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Overview

About this user’s guide

This is the official user’s guide for Stan. It provides example models and programming techniques for coding statistical models in Stan.

- Part 1 gives Stan code and discussions for several important classes of models.
- Part 2 discusses various general Stan programming techniques that are not tied to any particular model.
- Part 3 introduces algorithms for calibration and model checking that require multiple runs of Stan.
- The appendices provide a style guide and advice for users of BUGS and JAGS.

In addition to this user's guide, there are two reference manuals for the Stan language and algorithms. The Stan Reference Manual specifies the Stan programming language and inference algorithms. The Stan Functions Reference specifies the functions built into the Stan programming language.

There is also a separate installation and getting started guide for each of the Stan interfaces (R, Python, Julia, Stata, MATLAB, Mathematica, and command line).

We recommend working through this guide using the textbooks Bayesian Data Analysis and Statistical Rethinking: A Bayesian Course with Examples in R and Stan as references on the concepts, and using the Stan Reference Manual when necessary to clarify programming issues.

Web resources

Stan is an open-source software project, resources for which are hosted on various web sites:

- The Stan Web Site organizes all of the resources for the Stan project for users and developers. It contains links to the official Stan releases, source code, installation instructions, and full documentation, including the latest version of this manual, the user’s guide and the getting started guide for each interface, tutorials, case studies, and reference materials for developers.
• The Stan Forums provide structured message boards for questions, discussion, and announcements related to Stan for both users and developers.

• The Stan GitHub Organization hosts all of Stan’s code, documentation, wikis, and web site, as well as the issue trackers for bug reports and feature requests and interactive code review for pull requests.

Acknowledgements

The Stan project could not exist without developers, users, and funding. Stan is a highly collaborative project. The individual contributions of the Stan developers to code is tracked through GitHub and to the design conversation in the Wikis and forums.

Users have made extensive contributions to documentation in the way of case studies, tutorials and even books. They have also reported numerous bugs in both the code and documentation.

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Part 1. Example Models

In this part of the book, we survey a range of example models, with the goal of illustrating how to code them efficiently in Stan.
1. Regression Models

Stan supports regression models from simple linear regressions to multilevel generalized linear models.

1.1. Linear regression

The simplest linear regression model is the following, with a single predictor and a slope and intercept coefficient, and normally distributed noise. This model can be written using standard regression notation as

\[ y_n = \alpha + \beta x_n + \epsilon_n \quad \text{where} \quad \epsilon_n \sim \text{normal}(0, \sigma). \]

This is equivalent to the following sampling involving the residual,

\[ y_n - (\alpha + \beta X_n) \sim \text{normal}(0, \sigma), \]

and reducing still further, to

\[ y_n \sim \text{normal}(\alpha + \beta X_n, \sigma). \]

This latter form of the model is coded in Stan as follows.

```stan
data {
  int<lower=0> N;
  vector[N] x;
  vector[N] y;
}
parameters {
  real alpha;
  real beta;
  real<lower=0> sigma;
}
model {
  y ~ normal(alpha + beta * x, sigma);
}
```

There are \( N \) observations and for each observation, \( n \in N \), we have predictor \( x[n] \) and outcome \( y[n] \). The intercept and slope parameters are \( \alpha \) and \( \beta \). The model assumes a normally distributed noise term with scale \( \sigma \). This model has improper priors for the two regression coefficients.
**Matrix notation and vectorization**

The sampling statement in the previous model is vectorized, with

\[
y \sim \text{normal}(\alpha + \beta \times x, \sigma);
\]

providing the same model as the unvectorized version,

\[
\text{for (n in 1:N)} \\
\quad y[n] \sim \text{normal}(\alpha + \beta \times x[n], \sigma);
\]

In addition to being more concise, the vectorized form is much faster.\(^1\)

In general, Stan allows the arguments to distributions such as `normal` to be vectors. If any of the other arguments are vectors or arrays, they have to be the same size. If any of the other arguments is a scalar, it is reused for each vector entry. See the vectorization section for more information on vectorization of probability functions.

The other reason this works is that Stan’s arithmetic operators are overloaded to perform matrix arithmetic on matrices. In this case, because \(x\) is of type `vector` and \(\beta\) of type `real`, the expression \(\beta \times x\) is of type `vector`. Because Stan supports vectorization, a regression model with more than one predictor can be written directly using matrix notation.

```
data {
  int<lower=0> N; // number of data items
  int<lower=0> K; // number of predictors
  matrix[N, K] x; // predictor matrix
  vector[N] y; // outcome vector
}
parameters {
  real alpha; // intercept
  vector[K] beta; // coefficients for predictors
  real<lower=0> sigma; // error scale
}
model {
  y ~ normal(x \times beta + alpha, sigma); // likelihood
}
```

\(^1\)Unlike in Python and R, which are interpreted, Stan is translated to C++ and compiled, so loops and assignment statements are fast. Vectorized code is faster in Stan because (a) the expression tree used to compute derivatives can be simplified, leading to fewer virtual function calls, and (b) computations that would be repeated in the looping version, such as \(\log(\sigma)\) in the above model, will be computed once and reused.
The constraint \( \text{lower} = 0 \) in the declaration of \( \sigma \) constrains the value to be greater than or equal to 0. With no prior in the model block, the effect is an improper prior on non-negative real numbers. Although a more informative prior may be added, improper priors are acceptable as long as they lead to proper posteriors.

In the model above, \( x \) is an \( N \times K \) matrix of predictors and \( \beta \) a \( K \)-vector of coefficients, so \( x \times \beta \) is an \( N \)-vector of predictions, one for each of the \( N \) data items. These predictions line up with the outcomes in the \( N \)-vector \( y \), so the entire model may be written using matrix arithmetic as shown. It would be possible to include a column of ones in the data matrix \( x \) to remove the \( \alpha \) parameter.

The sampling statement in the model above is just a more efficient, vector-based approach to coding the model with a loop, as in the following statistically equivalent model.

```stan
model {
  for (n in 1:N)
    y[n] ~ normal(x[n] * beta, sigma);
}
```

With Stan’s matrix indexing scheme, \( x[n] \) picks out row \( n \) of the matrix \( x \); because \( \beta \) is a column vector, the product \( x[n] \times \beta \) is a scalar of type \( \text{real} \).

### Intercepts as inputs

In the model formulation

\[
y \sim \text{normal}(x \times \beta, \sigma);
\]

there is no longer an intercept coefficient \( \alpha \). Instead, we have assumed that the first column of the input matrix \( x \) is a column of 1 values. This way, \( \beta[1] \) plays the role of the intercept. If the intercept gets a different prior than the slope terms, then it would be clearer to break it out. It is also slightly more efficient in its explicit form with the intercept variable singled out because there’s one fewer multiplications; it should not make that much of a difference to speed, though, so the choice should be based on clarity.

#### 1.2. The QR reparameterization

In the previous example, the linear predictor can be written as \( \eta = x\beta \), where \( \eta \) is a \( N \)-vector of predictions, \( x \) is a \( N \times K \) matrix, and \( \beta \) is a \( K \)-vector of coefficients. Presuming \( N \geq K \), we can exploit the fact that any design matrix \( x \) can be decomposed using the thin QR decomposition into an orthogonal matrix \( Q \) and an upper-triangular matrix \( R \), i.e. \( x = QR \).
The functions `qr_thin_Q` and `qr_thin_R` implement the thin QR decomposition, which is to be preferred to the fat QR decomposition that would be obtained by using `qr_Q` and `qr_R`, as the latter would more easily run out of memory (see the Stan Functions Reference for more information on the `qr_thin_Q` and `qr_thin_R` functions). In practice, it is best to write \( x = Q^*R^\ast \) where \( Q^* = Q \star \sqrt{n - 1} \) and \( R^* = \frac{1}{\sqrt{n - 1}} R \). Thus, we can equivalently write \( \eta = x\beta = QR\beta = Q^*R^*\beta \). If we let \( \theta = R^*\beta \), then we have \( \eta = Q^*\theta \) and \( \beta = R^{-1}\theta \). In that case, the previous Stan program becomes

```stan
data {
    int<lower=0> N;      // number of data items
    int<lower=0> K;      // number of predictors
    matrix[N, K] x;       // predictor matrix
    vector[N] y;          // outcome vector
}
transformed data {
    matrix[N, K] Q_ast;
    matrix[K, K] R_ast;
    matrix[K, K] R_ast_inverse;
    // thin and scale the QR decomposition
    Q_ast = qr_thin_Q(x) * sqrt(N - 1);
    R_ast = qr_thin_R(x) / sqrt(N - 1);
    R_ast_inverse = inverse(R_ast);
}
parameters {
    real alpha;            // intercept
    vector[K] theta;       // coefficients on Q_ast
    real<lower=0> sigma;   // error scale
}
model {
    y ~ normal(Q_ast * theta + alpha, sigma); // likelihood
}
generated quantities {
    vector[K] beta;
    beta = R_ast_inverse * theta; // coefficients on x
}
```

Since this Stan program generates equivalent predictions for \( y \) and the same posterior distribution for \( \alpha, \beta, \) and \( \sigma \) as the previous Stan program, many wonder why the version with this QR reparameterization performs so much better in practice, often
both in terms of wall time and in terms of effective sample size. The reasoning is threefold:

1. The columns of $Q^*$ are orthogonal whereas the columns of $x$ generally are not. Thus, it is easier for a Markov Chain to move around in $\theta$-space than in $\beta$-space.

2. The columns of $Q^*$ have the same scale whereas the columns of $x$ generally do not. Thus, a Hamiltonian Monte Carlo algorithm can move around the parameter space with a smaller number of larger steps.

3. Since the covariance matrix for the columns of $Q^*$ is an identity matrix, $\theta$ typically has a reasonable scale if the units of $y$ are also reasonable. This also helps HMC move efficiently without compromising numerical accuracy.

Consequently, this QR reparameterization is recommended for linear and generalized linear models in Stan whenever $K > 1$ and you do not have an informative prior on the location of $\beta$. It can also be worthwhile to subtract the mean from each column of $x$ before obtaining the QR decomposition, which does not affect the posterior distribution of $\theta$ or $\beta$ but does affect $\alpha$ and allows you to interpret $\alpha$ as the expectation of $y$ in a linear model.

1.3. **Priors for coefficients and scales**

See our general discussion of priors for tips on priors for parameters in regression models.

Later sections discuss univariate hierarchical priors and multivariate hierarchical priors, as well as priors used to identify models.

However, as described in QR-reparameterization section, if you do not have an informative prior on the location of the regression coefficients, then you are better off reparameterizing your model so that the regression coefficients are a generated quantity. In that case, it usually does not matter much what prior is used on on the reparameterized regression coefficients and almost any weakly informative prior that scales with the outcome will do.

1.4. **Robust noise models**

The standard approach to linear regression is to model the noise term $\epsilon$ as having a normal distribution. From Stan’s perspective, there is nothing special about normally distributed noise. For instance, robust regression can be accommodated by giving the noise term a Student-$t$ distribution. To code this in Stan, the sampling distribution is changed to the following.
1.5. Logistic and probit regression

For binary outcomes, either of the closely related logistic or probit regression models may be used. These generalized linear models vary only in the link function they use to map linear predictions in $(-\infty, \infty)$ to probability values in $(0, 1)$. Their respective link functions, the logistic function and the standard normal cumulative distribution function, are both sigmoid functions (i.e., they are both S-shaped).

A logistic regression model with one predictor and an intercept is coded as follows.

```plaintext
data {
  int<lower=0> N;
  vector[N] x;
  int<lower=0,upper=1> y[N];
}
parameters {
  real alpha;
  real beta;
}
model {
  y ~ bernoulli_logit(alpha + beta * x);
}
```

The noise parameter is built into the Bernoulli formulation here rather than specified directly.

Logistic regression is a kind of generalized linear model with binary outcomes and the log odds (logit) link function, defined by

$$\text{logit}(v) = \log \left( \frac{v}{1 - v} \right).$$
The inverse of the link function appears in the model:

\[ \text{logit}^{-1}(u) = \text{inv\_logit}(u) = \frac{1}{1 + \exp(-u)}. \]

The model formulation above uses the logit-parameterized version of the Bernoulli distribution, which is defined by

\[ \text{bernoulli\_logit}(y | \alpha) = \text{bernoulli}(y | \text{logit}^{-1}(\alpha)). \]

The formulation is also vectorized in the sense that \( \alpha \) and \( \beta \) are scalars and \( x \) is a vector, so that \( \alpha + \beta \times x \) is a vector. The vectorized formulation is equivalent to the less efficient version

\[
\text{for} \ (n \ \text{in} \ 1:N) \\
\quad y[n] \sim \text{bernoulli\_logit}(\alpha + \beta \times x[n]);
\]

Expanding out the Bernoulli logit, the model is equivalent to the more explicit, but less efficient and less arithmetically stable

\[
\text{for} \ (n \ \text{in} \ 1:N) \\
\quad y[n] \sim \text{bernoulli}(\text{inv\_logit}(\alpha + \beta \times x[n]));
\]

Other link functions may be used in the same way. For example, probit regression uses the cumulative normal distribution function, which is typically written as

\[ \Phi(x) = \int_{-\infty}^{x} \text{normal}(y | 0, 1) \, dy. \]

The cumulative standard normal distribution function \( \Phi \) is implemented in Stan as the function \( \text{Phi} \). The probit regression model may be coded in Stan by replacing the logistic model’s sampling statement with the following.

\[ y[n] \sim \text{bernoulli}(\Phi(\alpha + \beta \times x[n])); \]

A fast approximation to the cumulative standard normal distribution function \( \Phi \) is implemented in Stan as the function \( \text{Phi\_approx} \).\(^2\) The approximate probit regression model may be coded with the following.

\[ y[n] \sim \text{bernoulli}(\text{Phi\_approx}(\alpha + \beta \times x[n])); \]

\(^2\)The \( \text{Phi\_approx} \) function is a rescaled version of the inverse logit function, so while the scale is roughly the same \( \Phi \), the tails do not match.
1.6. Multi-logit regression
Multiple outcome forms of logistic regression can be coded directly in Stan. For instance, suppose there are \( K \) possible outcomes for each output variable \( y_n \). Also suppose that there is a \( D \)-dimensional vector \( x_n \) of predictors for \( y_n \). The multi-logit model with normal(0,5) priors on the coefficients is coded as follows.

```stan
data {
  int K;
  int N;
  int D;
  int y[N];
  matrix[N, D] x;
}
parameters {
  matrix[D, K] beta;
}
model {
  matrix[N, K] x_beta = x * beta;
  to_vector(beta) ~ normal(0, 5);
  for (n in 1:N)
    y[n] ~ categorical_logit(x_beta[n]);
}
```

where \( x_{\text{beta}[n]}' \) is the transpose of \( x_{\text{beta}[n]} \). The prior on \( \beta \) is coded in vectorized form. As of Stan 2.18, the categorical-logit distribution is not vectorized for parameter arguments, so the loop is required. The matrix multiplication is pulled out to define a local variable for all of the predictors for efficiency. Like the Bernoulli-logit, the categorical-logit distribution applies softmax internally to convert an arbitrary vector to a simplex,

\[
categorical\_logit(y \mid \alpha) = categorical(y \mid \text{softmax}(\alpha)),
\]

where

\[
\text{softmax}(u) = \exp(u) / \text{sum}(\exp(u)).
\]

The categorical distribution with log-odds (logit) scaled parameters used above is equivalent to writing

\[
y[n] \sim \text{categorical(softmax(x[n] * beta))};
\]
Constraints on data declarations

The data block in the above model is defined without constraints on sizes $K$, $N$, and $D$ or on the outcome array $y$. Constraints on data declarations provide error checking at the point data are read (or transformed data are defined), which is before sampling begins. Constraints on data declarations also make the model author’s intentions more explicit, which can help with readability. The above model’s declarations could be tightened to

```plaintext
int<lower = 2> K;
int<lower = 0> N;
int<lower = 1> D;
int<lower = 1, upper = K> y[N];
```

These constraints arise because the number of categories, $K$, must be at least two in order for a categorical model to be useful. The number of data items, $N$, can be zero, but not negative; unlike R, Stan’s for-loops always move forward, so that a loop extent of $1:N$ when $N$ is equal to zero ensures the loop’s body will not be executed. The number of predictors, $D$, must be at least one in order for $\text{beta} \times x[n]$ to produce an appropriate argument for $\text{softmax()}$. The categorical outcomes $y[n]$ must be between 1 and $K$ in order for the discrete sampling to be well defined.

Constraints on data declarations are optional. Constraints on parameters declared in the `parameters` block, on the other hand, are not optional—they are required to ensure support for all parameter values satisfying their constraints. Constraints on transformed data, transformed parameters, and generated quantities are also optional.

Identifiability

Because $\text{softmax}$ is invariant under adding a constant to each component of its input, the model is typically only identified if there is a suitable prior on the coefficients.

An alternative is to use $(K - 1)$-vectors by fixing one of them to be zero. The partially known parameters section discusses how to mix constants and parameters in a vector. In the multi-logit case, the parameter block would be redefined to use $(K - 1)$-vectors

```plaintext
parameters {
  matrix[K - 1, D] beta_raw;
}
```

and then these are transformed to parameters to use in the model. First, a transformed data block is added before the parameters block to define a row vector of zero values,
transformed data {
    row_vector[D] zeros = rep_row_vector(0, D);
}

which can then be appended to beta_row to produce the coefficient matrix beta,

transformed parameters {
    matrix[K, D] beta;
    beta = append_row(beta_raw, zeros);
}

The `rep_row_vector(0, D)` call creates a row vector of size D with all entries set to zero. The derived matrix beta is then defined to be the result of appending the row-vector zeros as a new row at the end of beta_raw; the row vector zeros is defined as transformed data so that it doesn’t need to be constructed from scratch each time it is used.

This is not the same model as using K-vectors as parameters, because now the prior only applies to \((K - 1)\)-vectors. In practice, this will cause the maximum likelihood solutions to be different and also the posteriors to be slightly different when taking priors centered around zero, as is typical for regression coefficients.

### 1.7. Parameterizing centered vectors

It is often convenient to define a parameter vector \(\beta\) that is centered in the sense of satisfying the sum-to-zero constraint,

\[
\sum_{k=1}^{K} \beta_k = 0.
\]

Such a parameter vector may be used to identify a multi-logit regression parameter vector (see the multi-logit section for details), or may be used for ability or difficulty parameters (but not both) in an IRT model (see the item-response model section for details).

#### K − 1 degrees of freedom

There is more than one way to enforce a sum-to-zero constraint on a parameter vector, the most efficient of which is to define the \(K\)-th element as the negation of the sum of the elements 1 through \(K - 1\).

parameters {
    vector[K-1] beta_raw;
    ...
transformed parameters {
    vector[K] beta = append_row(beta_raw, -sum(beta_raw));
    ...
}

Placing a prior on beta_raw in this parameterization leads to a subtly different posterior than that resulting from the same prior on beta in the original parameterization without the sum-to-zero constraint. Most notably, a simple prior on each component of beta_raw produces different results than putting the same prior on each component of an unconstrained K-vector beta. For example, providing a normal(0, 5) prior on beta will produce a different posterior mode than placing the same prior on beta_raw.

**Marginal distribution of sum-to-zero components**

On the Stan forums, Aaron Goodman provided the following code to produce a prior with standard normal marginals on the components of beta,

```stan
model {
    beta ~ normal(0, inv(sqrt(1 - inv(K))));
    ...
}
```

The components are not independent, as they must sum zero. No Jacobian is required because summation and negation are linear operations (and thus have constant Jacobians).

To generate distributions with marginals other than standard normal, the resulting beta may be scaled by some factor sigma and translated to some new location mu.

**QR decomposition**

Aaron Goodman, on the Stan forums, also provided this approach, which calculates a QR decomposition in the transformed data block, then uses it to transform to a sum-to-zero parameter x,

```stan
transformed data{
    matrix[K, K] A = diag_matrix(rep_vector(1,K));
    matrix[K, K-1] A_qr;
    for (i in 1:K-1) A[K,i] = -1;
    A[K,K] = 0;
    A_qr = qr_Q(A)[ , 1:(K-1)];
}
parameters {
    vector[K-1] beta_raw;
}
transformed parameters{
    vector[K] beta = A_qr * beta_raw;
}

model {
    beta_raw ~ normal(0, inv(sqrt(1 - inv(K))));
}

This produces a marginal standard normal distribution on the values of \( \beta \), which will sum to zero by construction of the QR decomposition.

**Translated and scaled simplex**

An alternative approach that’s less efficient, but amenable to a symmetric prior, is to offset and scale a simplex.

parameters {
    simplex[K] beta_raw;
    real beta_scale;
    ...
}

transformed parameters {
    vector[K] beta;
    beta = beta_scale * (beta_raw - inv(K));
    ...
}

Here \( \text{inv}(K) \) is just a short way to write \( 1.0 / K \). Given that \( \beta_{\text{raw}} \) sums to 1 because it is a simplex, the elementwise subtraction of \( \text{inv}(K) \) is guaranteed to sum to zero. Because the magnitude of the elements of the simplex is bounded, a scaling factor is required to provide \( \beta \) with \( K \) degrees of freedom necessary to take on every possible value that sums to zero.

With this parameterization, a Dirichlet prior can be placed on \( \beta_{\text{raw}} \), perhaps uniform, and another prior put on \( \beta_{\text{scale}} \), typically for “shrinkage.”

**Soft centering**

Adding a prior such as \( \beta \sim \text{normal}(0, \sigma) \) will provide a kind of soft centering of a parameter vector \( \beta \) by preferring, all else being equal, that \( \sum_{k=1}^{K} \beta_k = 0 \). This approach is only guaranteed to roughly center if \( \beta \) and the elementwise addition \( \beta + c \) for a scalar constant \( c \) produce the same likelihood (perhaps by another vector \( \alpha \) being transformed to \( \alpha - c \), as in the IRT models). This is another way of achieving a symmetric prior.
1.8. Ordered logistic and probit regression

Ordered regression for an outcome $y_n \in \{1, \ldots, k\}$ with predictors $x_n \in \mathbb{R}^D$ is determined by a single coefficient vector $\beta \in \mathbb{R}^D$ along with a sequence of cutpoints $c \in \mathbb{R}^{K-1}$ sorted so that $c_d < c_{d+1}$. The discrete output is $k$ if the linear predictor $x_n \beta$ falls between $c_{k-1}$ and $c_k$, assuming $c_0 = -\infty$ and $c_K = \infty$. The noise term is fixed by the form of regression, with examples for ordered logistic and ordered probit models.

**Ordered logistic regression**

The ordered logistic model can be coded in Stan using the ordered data type for the cutpoints and the built-in `ordered_logistic` distribution.

```stan
data {
  int<lower=2> K;
  int<lower=0> N;
  int<lower=1> D;
  int<lower=1,upper=K> y[N];
  row_vector[D] x[N];
}
parameters {
  vector[D] beta;
  ordered[K-1] c;
}
model {
  for (n in 1:N)
    y[n] ~ ordered_logistic(x[n] * beta, c);
}
```

The vector of cutpoints $c$ is declared as `ordered[K-1]`, which guarantees that $c[k]$ is less than $c[k+1]$.

If the cutpoints were assigned independent priors, the constraint effectively truncates the joint prior to support over points that satisfy the ordering constraint. Luckily, Stan does not need to compute the effect of the constraint on the normalizing term because the probability is needed only up to a proportion.

**Ordered probit**

An ordered probit model could be coded in exactly the same way by swapping the cumulative logistic (`inv_logit`) for the cumulative normal (`Phi`).

```stan
data {
  int<lower=2> K;
  int<lower=0> N;
  int<lower=1> D;
  int<lower=1,upper=K> y[N];
  row_vector[D] x[N];
}
parameters {
  vector[D] beta;
  ordered[K-1] c;
}
model {
  for (n in 1:N)
    y[n] ~ ordered_logistic(x[n] * beta, c);
}
```
\begin{verbatim}
int<lower=2>  K;
int<lower=0>  N;
int<lower=1>  D;
int<lower=1,upper=K>  y[N];
row_vector[D]  x[N];
}\}
parameters {
  vector[D]  beta;
  ordered[K-1]  c;
}
model {
  vector[K]  theta;
  for (n in 1:N) {
    real  eta;
    eta = x[n] * beta;
    theta[1] = 1 - Phi(eta - c[1]);
    for (k in 2:(K-1))
      theta[k] = Phi(eta - c[k-1]) - Phi(eta - c[k]);
    theta[K] = Phi(eta - c[K-1]);
    y[n] ~ categorical(theta);
  }
}
\end{verbatim}

The logistic model could also be coded this way by replacing \texttt{Phi} with \texttt{inv\_logit}, though the built-in encoding based on the softmax transform is more efficient and more numerically stable. A small efficiency gain could be achieved by computing the values \(\Phi(\eta - c[k])\) once and storing them for re-use.

1.9. Hierarchical logistic regression

The simplest multilevel model is a hierarchical model in which the data are grouped into \(L\) distinct categories (or levels). An extreme approach would be to completely pool all the data and estimate a common vector of regression coefficients \(\beta\). At the other extreme, an approach with no pooling assigns each level \(l\) its own coefficient vector \(\beta_l\) that is estimated separately from the other levels. A hierarchical model is an intermediate solution where the degree of pooling is determined by the data and a prior on the amount of pooling.

Suppose each binary outcome \(y_n \in \{0, 1\}\) has an associated level, \(l_n \in \{1, \ldots, L\}\). Each outcome will also have an associated predictor vector \(x_n \in \mathbb{R}^D\). Each level \(l\) gets its own coefficient vector \(\beta_l \in \mathbb{R}^D\). The hierarchical structure involves drawing the coefficients \(\beta_{l,d} \in \mathbb{R}\) from a prior that is also estimated with the data. This
hierarchically estimated prior determines the amount of pooling. If the data in each level are similar, strong pooling will be reflected in low hierarchical variance. If the data in the levels are dissimilar, weaker pooling will be reflected in higher hierarchical variance.

The following model encodes a hierarchical logistic regression model with a hierarchical prior on the regression coefficients.

```stan
data {
  int<lower=1> D;
  int<lower=0> N;
  int<lower=1> L;
  int<lower=0,upper=1> y[N];
  int<lower=1,upper=L> ll[N];
  row_vector[D] x[N];
}
parameters {
  real mu[D];
  real<lower=0> sigma[D];
  vector[D] beta[L];
}
model {
  for (d in 1:D) {
    mu[d] ~ normal(0, 100);
    for (l in 1:L)
      beta[l,d] ~ normal(mu[d], sigma[d]);
  }
  for (n in 1:N)
    y[n] ~ bernoulli(inv_logit(x[n] * beta[ll[n]]));
}
```

The standard deviation parameter $\sigma$ gets an implicit uniform prior on $(0, \infty)$ because of its declaration with a lower-bound constraint of zero. Stan allows improper priors as long as the posterior is proper. Nevertheless, it is usually helpful to have informative or at least weakly informative priors for all parameters; see the regression priors section for recommendations on priors for regression coefficients and scales.
Optimizing the model

Where possible, vectorizing sampling statements leads to faster log probability and derivative evaluations. The speed boost is not because loops are eliminated, but because vectorization allows sharing subcomputations in the log probability and gradient calculations and because it reduces the size of the expression tree required for gradient calculations.

The first optimization vectorizes the for-loop over $D$ as

```latex
mu ~ normal(0, 100);
for (l in 1:L)
  beta[l] ~ normal(mu, sigma);
```

The declaration of $\beta$ as an array of vectors means that the expression $\beta[l]$ denotes a vector. Although $\beta$ could have been declared as a matrix, an array of vectors (or a two-dimensional array) is more efficient for accessing rows; see the indexing efficiency section for more information on the efficiency tradeoffs among arrays, vectors, and matrices.

This model can be further sped up and at the same time made more arithmetically stable by replacing the application of inverse-logit inside the Bernoulli distribution with the logit-parameterized Bernoulli,$^3$

```latex
for (n in 1:N)
  y[n] ~ bernoulli_logit(x[n] * beta[ll[n]]);
```

Unlike in R or BUGS, loops, array access and assignments are fast in Stan because they are translated directly to C++. In most cases, the cost of allocating and assigning to a container is more than made up for by the increased efficiency due to vectorizing the log probability and gradient calculations. Thus the following version is faster than the original formulation as a loop over a sampling statement.

```latex
{ 
  vector[N] x_beta_ll;
  for (n in 1:N)
    x_beta_ll[n] = x[n] * beta[ll[n]];
  y ~ bernoulli_logit(x_beta_ll);
}
```

---

$^3$The Bernoulli-logit distribution builds in the log link function, taking

$$bernoulli\_logit(y \mid \alpha) = bernoulli(y \mid \logit^{-1}(\alpha)).$$
The brackets introduce a new scope for the local variable $x_{\text{beta}_1}$; alternatively, the variable may be declared at the top of the model block.

In some cases, such as the above, the local variable assignment leads to models that are less readable. The recommended practice in such cases is to first develop and debug the more transparent version of the model and only work on optimizations when the simpler formulation has been debugged.

1.10. Hierarchical priors

Priors on priors, also known as “hyperpriors,” should be treated the same way as priors on lower-level parameters in that as much prior information as is available should be brought to bear. Because hyperpriors often apply to only a handful of lower-level parameters, care must be taken to ensure the posterior is both proper and not overly sensitive either statistically or computationally to wide tails in the priors.

Boundary-avoiding priors for MLE in hierarchical models

The fundamental problem with maximum likelihood estimation (MLE) in the hierarchical model setting is that as the hierarchical variance drops and the values cluster around the hierarchical mean, the overall density grows without bound. As an illustration, consider a simple hierarchical linear regression (with fixed prior mean) of $y_n \in \mathbb{R}$ on $x_n \in \mathbb{R}^K$, formulated as

\[
\begin{align*}
y_n &\sim \text{normal}(x_n \beta, \sigma) \\
\beta_k &\sim \text{normal}(0, \tau) \\
\tau &\sim \text{Cauchy}(0, 2.5)
\end{align*}
\]

In this case, as $\tau \to 0$ and $\beta_k \to 0$, the posterior density

\[
p(\beta, \tau, \sigma|y, x) \propto p(y|x, \beta, \tau, \sigma)
\]

grows without bound. See the plot of Neal's funnel density, which has similar behavior.

There is obviously no MLE estimate for $\beta, \tau, \sigma$ in such a case, and therefore the model must be modified if posterior modes are to be used for inference. The approach recommended by Chung et al. (2013) is to use a gamma distribution as a prior, such as

\[
\sigma \sim \text{Gamma}(2, 1/A),
\]

for a reasonably large value of $A$, such as $A = 10$.  


1.11. Item-response theory models
Item-response theory (IRT) models the situation in which a number of students each answer one or more of a group of test questions. The model is based on parameters for the ability of the students, the difficulty of the questions, and in more articulated models, the discriminativeness of the questions and the probability of guessing correctly; see Andrew Gelman and Hill (2007, pps. 314–320) for a textbook introduction to hierarchical IRT models and Curtis (2010) for encodings of a range of IRT models in BUGS.

**Data declaration with missingness**
The data provided for an IRT model may be declared as follows to account for the fact that not every student is required to answer every question.

```stan
data {
  int<lower=1> J; // number of students
  int<lower=1> K; // number of questions
  int<lower=1> N; // number of observations
  int<lower=1,upper=J> jj[N]; // student for observation n
  int<lower=1,upper=K> kk[N]; // question for observation n
  int<lower=0,upper=1> y[N]; // correctness for observation n
}
```

This declares a total of \( N \) student-question pairs in the data set, where each \( n \) in \( 1:N \) indexes a binary observation \( y[n] \) of the correctness of the answer of student \( jj[n] \) on question \( kk[n] \).

The prior hyperparameters will be hard coded in the rest of this section for simplicity, though they could be coded as data in Stan for more flexibility.

**1PL (Rasch) model**
The 1PL item-response model, also known as the Rasch model, has one parameter (1P) for questions and uses the logistic link function (L).

The model parameters are declared as follows.

```stan
parameters {
  real delta; // mean student ability
  real alpha[J]; // ability of student j - mean ability
  real beta[K]; // difficulty of question k
}
```

The parameter \( \alpha[J] \) is the ability coefficient for student \( j \) and \( \beta[k] \) is the difficulty coefficient for question \( k \). The non-standard parameterization used
here also includes an intercept term \( \delta \), which represents the average student’s response to the average question.\(^4\)

The model itself is as follows.

```stan
model {
  alpha ~ std_normal(); // informative true prior
  beta ~ std_normal(); // informative true prior
  delta ~ normal(0.75, 1); // informative true prior
  for (n in 1:N)
    y[n] ~ bernoulli_logit(alpha[jj[n]] - beta[kk[n]] + delta);
}
```

This model uses the logit-parameterized Bernoulli distribution, where

\[
\text{bernoulli\_logit}(y \mid \alpha) = \text{bernoulli}(y \mid \logit^{-1}(\alpha)).
\]

The key to understanding it is the term inside the \text{bernoulli\_logit} distribution, from which it follows that

\[
\Pr[y_n = 1] = \logit^{-1}(\alpha_{jj[n]} - \beta_{kk[n]} + \delta).
\]

The model suffers from additive identifiability issues without the priors. For example, adding a term \( \xi \) to each \( \alpha_j \) and \( \beta_k \) results in the same predictions. The use of priors for \( \alpha \) and \( \beta \) located at 0 identifies the parameters; see Andrew Gelman and Hill (2007) for a discussion of identifiability issues and alternative approaches to identification.

For testing purposes, the IRT 1PL model distributed with Stan uses informative priors that match the actual data generation process used to simulate the data in R (the simulation code is supplied in the same directory as the models). This is unrealistic for most practical applications, but allows Stan’s inferences to be validated. A simple sensitivity analysis with fatter priors shows that the posterior is fairly sensitive to the prior even with 400 students and 100 questions and only 25% missingness at random. For real applications, the priors should be fit hierarchically along with the other parameters, as described in the next section.

**Multilevel 2PL model**

The simple 1PL model described in the previous section is generalized in this section with the addition of a discrimination parameter to model how noisy a question is and

\(^4\)Andrew Gelman and Hill (2007) treat the \( \delta \) term equivalently as the location parameter in the distribution of student abilities.
by adding multilevel priors for the question difficulty and discrimination parameters. The model parameters are declared as follows.

```plaintext
parameters {
  real mu_beta; // mean question difficulty
  vector[J] alpha; // ability for j - mean
  vector[K] beta; // difficulty for k
  vector<lower=0>[K] gamma; // discrimination of k
  real<lower=0> sigma_beta; // scale of difficulties
  real<lower=0> sigma_gamma; // scale of log discrimination
}
```

The parameters should be clearer after the model definition.

```plaintext
model {
  alpha ~ std_normal();
  beta ~ normal(0, sigma_beta);
  gamma ~ lognormal(0, sigma_gamma);
  mu_beta ~ cauchy(0, 5);
  sigma_beta ~ cauchy(0, 5);
  sigma_gamma ~ cauchy(0, 5);
  y ~ bernoulli_logit(gamma[kk] .* (alpha[jj] - (beta[kk] + mu_beta)));
}
```

The `std_normal` function is used here, defined by

\[
\text{std\_normal}(y) = \text{normal}(y \mid 0, 1).
\]

The sampling statement is also vectorized using elementwise multiplication; it is equivalent to

```plaintext
for (n in 1:N)
  y[n] ~ bernoulli_logit(gamma[kk[n]]
  * (alpha[jj[n]] - (beta[kk[n]] + mu_beta));
```

The 2PL model is similar to the 1PL model, with the additional parameter `gamma[k]` modeling how discriminative question `k` is. If `gamma[k]` is greater than 1, responses are more attenuated with less chance of getting a question right at random. The parameter `gamma[k]` is constrained to be positive, which prohibits there being questions that are easier for students of lesser ability; such questions are not unheard of, but they tend to be eliminated from most testing situations where an IRT model would be applied.
The model is parameterized here with student abilities \( \alpha \) being given a standard normal prior. This is to identify both the scale and the location of the parameters, both of which would be unidentified otherwise; see the problematic posteriors chapter for further discussion of identifiability. The difficulty and discrimination parameters \( \beta \) and \( \gamma \) then have varying scales given hierarchically in this model. They could also be given weakly informative non-hierarchical priors, such as

\[
\begin{align*}
\beta & \sim \text{normal}(0, 5); \\
\gamma & \sim \text{lognormal}(0, 2);
\end{align*}
\]

The point is that the \( \alpha \) determines the scale and location and \( \beta \) and \( \gamma \) are allowed to float.

The \( \beta \) parameter is here given a non-centered parameterization, with parameter \( \mu_\beta \) serving as the mean \( \beta \) location. An alternative would’ve been to take:

\[
\begin{align*}
\beta & \sim \text{normal}(\mu_\beta, \sigma_\beta); \\
\end{align*}
\]

and

\[
\begin{align*}
y[n] & \sim \text{bernoulli_logit}(\gamma[kk[n]] \times (\alpha[jj[n]] - \beta[kk[n]]));
\end{align*}
\]

Non-centered parameterizations tend to be more efficient in hierarchical models; see the reparameterization section for more information on non-centered reparameterizations.

The intercept term \( \mu_\beta \) can’t itself be modeled hierarchically, so it is given a weakly informative \( \text{Cauchy}(0, 5) \) prior. Similarly, the scale terms, \( \sigma_\beta \), and \( \sigma_\gamma \), are given half-Cauchy priors. As mentioned earlier, the scale and location for \( \alpha \) are fixed to ensure identifiability. The truncation in the half-Cauchy prior is implicit; explicit truncation is not necessary because the log probability need only be calculated up to a proportion and the scale variables are constrained to \((0, \infty)\) by their declarations.

### 1.12. Priors for identifiability

#### Location and scale invariance

One application of (hierarchical) priors is to identify the scale and/or location of a group of parameters. For example, in the IRT models discussed in the previous section, there is both a location and scale non-identifiability. With uniform priors, the posteriors will float in terms of both scale and location. See the collinearity section for a simple example of the problems this poses for estimation.

The non-identifiability is resolved by providing a standard normal (i.e., \( \text{normal}(0, 1) \)) prior on one group of coefficients, such as the student abilities. With a standard
normal prior on the student abilities, the IRT model is identified in that the posterior will produce a group of estimates for student ability parameters that have a sample mean of close to zero and a sample variance of close to one. The difficulty and discrimination parameters for the questions should then be given a diffuse, or ideally a hierarchical prior, which will identify these parameters by scaling and locating relative to the student ability parameters.

Collinearity
Another case in which priors can help provide identifiability is in the case of collinearity in a linear regression. In linear regression, if two predictors are collinear (i.e., one is a linear function of the other), then their coefficients will have a correlation of 1 (or -1) in the posterior. This leads to non-identifiability. By placing normal priors on the coefficients, the maximum likelihood solution of two duplicated predictors (trivially collinear) will be half the value than would be obtained by only including one.

Separability
In a logistic regression, if a predictor is positive in cases of 1 outcomes and negative in cases of 0 outcomes, then the maximum likelihood estimate for the coefficient for that predictor diverges to infinity. This divergence can be controlled by providing a prior for the coefficient, which will “shrink” the estimate back toward zero and thus identify the model in the posterior.

Similar problems arise for sampling with improper flat priors. The sampler will try to draw large values. By providing a prior, the posterior will be concentrated around finite values, leading to well-behaved sampling.

1.13. Multivariate priors for hierarchical models
In hierarchical regression models (and other situations), several individual-level variables may be assigned hierarchical priors. For example, a model with multiple varying intercepts and slopes within might assign them a multivariate prior.

As an example, the individuals might be people and the outcome income, with predictors such as education level and age, and the groups might be states or other geographic divisions. The effect of education level and age as well as an intercept might be allowed to vary by state. Furthermore, there might be state-level predictors, such as average state income and unemployment level.

Multivariate regression example
Andrew Gelman and Hill (2007, chap. 13, Chapter 17) provide a discussion of a hierarchical model with $N$ individuals organized into $J$ groups. Each individual has a predictor row vector $x_n$ of size $K$; to unify the notation, they assume that
$x_{n,1} = 1$ is a fixed “intercept” predictor. To encode group membership, they assume individual $n$ belongs to group $j$ if $n \in \{1, \ldots, J\}$. Each individual $n$ also has an observed outcome $y_n$ taking on real values.

**Likelihood**

The model is a linear regression with slope and intercept coefficients varying by group, so that $\beta_j$ is the coefficient $K$-vector for group $j$. The likelihood function for individual $n$ is then just

$$y_n \sim \text{normal}(x_n \beta_{jj[n]}, \sigma) \quad \text{for} \quad n \in \{1, \ldots, N\}.$$  

**Coefficient prior**

Gelman and Hill model the coefficient vectors $\beta_j$ as being drawn from a multivariate distribution with mean vector $\mu$ and covariance matrix $\Sigma$,

$$\beta_j \sim \text{multivariate normal}(\mu_j, \Sigma) \quad \text{for} \quad j \in \{1, \ldots, J\}.$$

Below, we discuss the full model of Gelman and Hill, which uses group-level predictors to model $\mu$; for now, we assume $\mu$ is a simple vector parameter.

**Hyperpriors**

For hierarchical modeling, the group-level mean vector $\mu$ and covariance matrix $\Sigma$ must themselves be given priors. The group-level mean vector can be given a reasonable weakly-informative prior for independent coefficients, such as

$$\mu_j \sim \text{normal}(0, 5).$$

If more is known about the expected coefficient values $\beta_{j,k}$, this information can be incorporated into the prior for $\mu_j$.

For the prior on the covariance matrix, Gelman and Hill suggest using a scaled inverse Wishart. That choice was motivated primarily by convenience as it is conjugate to the multivariate likelihood function and thus simplifies Gibbs sampling.

In Stan, there is no restriction to conjugacy for multivariate priors, and we in fact recommend a slightly different approach. Like Gelman and Hill, we decompose our prior into a scale and a matrix, but are able to do so in a more natural way based on the actual variable scales and a correlation matrix. Specifically, we define

$$\Sigma = \text{diag}_\text{matrix}(\tau) \times \Omega \times \text{diag}_\text{matrix}(\tau),$$
where $\Omega$ is a correlation matrix and $\tau$ is the vector of coefficient scales. This mapping from scale vector $\tau$ and correlation matrix $\Omega$ can be inverted, using

$$\tau_k = \sqrt{\Sigma_{k,k}} \quad \text{and} \quad \Omega_{i,j} = \frac{\Sigma_{i,j}}{\tau_i \tau_j}.$$ 

The components of the scale vector $\tau$ can be given any reasonable prior for scales, but we recommend something weakly informative like a half-Cauchy distribution with a small scale, such as

$$\tau_k \sim \text{Cauchy}(0, 2.5) \quad \text{for} \quad k \in \{1, \ldots, K\} \quad \text{constrained by} \quad \tau_k > 0.$$ 

As for the prior means, if there is information about the scale of variation of coefficients across groups, it should be incorporated into the prior for $\tau$. For large numbers of exchangeable coefficients, the components of $\tau$ itself (perhaps excluding the intercept) may themselves be given a hierarchical prior.

Our final recommendation is to give the correlation matrix $\Omega$ an LKJ prior with shape $\eta \geq 1$,

$$\Omega \sim \text{LKJCorr}(\eta).$$

The LKJ correlation distribution is defined by

$$\text{LKJCorr}(\Sigma \mid \eta) \propto \det(\Sigma)^{\eta-1}.$$ 

The basic behavior of the LKJ correlation distribution is similar to that of a beta distribution. For $\eta = 1$, the result is a uniform distribution. Despite being the identity over correlation matrices, the marginal distribution over the entries in that matrix (i.e., the correlations) is not uniform between -1 and 1. Rather, it concentrates around zero as the dimensionality increases due to the complex constraints.

For $\eta > 1$, the density increasingly concentrates mass around the unit matrix, i.e., favoring less correlation. For $\eta < 1$, it increasingly concentrates mass in the other direction, i.e., favoring more correlation.

The LKJ prior may thus be used to control the expected amount of correlation among the parameters $\beta_j$. For a discussion of decomposing a covariance prior into a prior on correlation matrices and an independent prior on scales, see Barnard, McCulloch, and Meng (2000).

---

5The prior is named for Lewandowski, Kurowicka, and Joe, as it was derived by inverting the random correlation matrix generation strategy of Lewandowski, Kurowicka, and Joe (2009).
Group-level predictors for prior mean

To complete Gelman and Hill's model, suppose each group \( j \in \{1, \ldots, J\} \) is supplied with an \( L \)-dimensional row-vector of group-level predictors \( u_j \). The prior mean for the \( \beta_j \) can then itself be modeled as a regression, using an \( L \)-dimensional coefficient vector \( \gamma \). The prior for the group-level coefficients then becomes

\[
\beta_j \sim \text{multivariate normal}(u_j \gamma, \Sigma)
\]

The group-level coefficients \( \gamma \) may themselves be given independent weakly informative priors, such as

\[
\gamma_l \sim \text{normal}(0, 5).
\]

As usual, information about the group-level means should be incorporated into this prior.

Coding the model in Stan

The Stan code for the full hierarchical model with multivariate priors on the group-level coefficients and group-level prior means follows its definition.

data {
  int<lower=0> N; // num individuals
  int<lower=1> K; // num ind predictors
  int<lower=1> J; // num groups
  int<lower=1> L; // num group predictors
  int<lower=1,upper=J> jj[N]; // group for individual
  matrix[N, K] x; // individual predictors
  row_vector[L] u[J]; // group predictors
  vector[N] y; // outcomes
}

parameters {
  corr_matrix[K] Omega; // prior correlation
  vector<lower=0>[K] tau; // prior scale
  matrix[L, K] gamma; // group coeffs
  vector[K] beta[J]; // indiv coeffs by group
  real<lower=0> sigma; // prediction error scale
}

model {
  tau ~ cauchy(0, 2.5);
  Omega ~ lkj_corr(2);
  y ~ normal(x * beta[jj], sigma);
  for (j in 1:J) {
    beta[j] ~ normal(u[j] * gamma, tau);
  }
}
to_vector(gamma) ~ normal(0, 5);
{
  row_vector[K] u_gamma[J];
  for (j in 1:J)
    u_gamma[j] = u[j] * gamma;
  beta ~ multi_normal(u_gamma, quad_form_diag(Omega, tau));
}
for (n in 1:N)
  y[n] ~ normal(x[n] * beta[jj[n]], sigma);

The hyperprior covariance matrix is defined implicitly through the a quadratic form in the code because the correlation matrix Omega and scale vector tau are more natural to inspect in the output; to output Sigma, define it as a transformed parameter. The function quad_form_diag is defined so that quad_form_diag(Sigma, tau) is equivalent to diag_matrix(tau) * Sigma * diag_matrix(tau), where diag_matrix(tau) returns the matrix with tau on the diagonal and zeroes off diagonal; the version using quad_form_diag should be faster. For details on these and other matrix arithmetic operators and functions, see the function reference manual.

Optimization through vectorization

The code in the Stan program above can be sped up dramatically by replacing:

```stan
for (n in 1:N)
  y[n] ~ normal(x[n] * beta[jj[n]], sigma);
```

with the vectorized form:

```stan
{
  vector[N] x_beta_jj;
  for (n in 1:N)
    x_beta_jj[n] = x[n] * beta[jj[n]];
  y ~ normal(x_beta_jj, sigma);
}
```

The outer brackets create a local scope in which to define the variable x_beta_jj, which is then filled in a loop and used to define a vectorized sampling statement. The reason this is such a big win is that it allows us to take the log of sigma only once and it greatly reduces the size of the resulting expression graph by packing all of the work into a single density function.
Although it is tempting to redeclare \( \beta \) and include a revised model block sampling statement,

\[
\text{parameters} \{ \\
\quad \text{matrix}[J, K] \beta; \\
\}
\]

\[
\text{model} \{ \\
\quad y \sim \text{normal}(\text{rows_dot_product}(x, \beta[jj]), \sigma); \\
\}
\]

this fails because it breaks the vectorization of sampling for \( \beta \),

\[
\beta \sim \text{multi_normal}(\ldots);
\]

which requires \( \beta \) to be an array of vectors. Both vectorizations are important, so the best solution is to just use the loop above, because \text{rows_dot_product} cannot do much optimization in and of itself because there are no shared computations.

The code in the Stan program above also builds up an array of vectors for the outcomes and for the multivariate normal, which provides a major speedup by reducing the number of linear systems that need to be solved and differentiated.

\[
\{ \\
\quad \text{matrix}[K, K] \text{Sigma}_\beta; \\
\quad \text{Sigma}_\beta = \text{quad_form_diag}(\Omega, \tau); \\
\quad \text{for (j in 1:J)} \\
\quad \quad \beta[j] \sim \text{multi_normal}((u[j] \times \gamma)' \times \text{Sigma}_\beta); \\
\}
\]

In this example, the covariance matrix \( \text{Sigma}_\beta \) is defined as a local variable so as not to have to repeat the quadratic form computation \( J \) times. This vectorization can be combined with the Cholesky-factor optimization in the next section.

**Optimization through Cholesky factorization**

The multivariate normal density and LKJ prior on correlation matrices both require their matrix parameters to be factored. Vectorizing, as in the previous section, ensures this is only done once for each density. An even better solution, both in terms of efficiency and numerical stability, is to parameterize the model directly in terms of Cholesky factors of correlation matrices using the multivariate version of the non-centered parameterization. For the model in the previous section, the program

\[\text{6}\text{Thanks to Mike Lawrence for pointing this out in the GitHub issue for the manual.}\]
fragment to replace the full matrix prior with an equivalent Cholesky factorized prior is as follows.

data {
  matrix[L, J] u; // group predictors transposed
...
}

parameters {
  matrix[K, J] z;
  cholesky_factor_corr[K] L_Omega;
  matrix[K, L] gamma;
...
}

transformed parameters {
  matrix[K, J] beta;
  beta = gamma * u + diag_pre_multiply(tau, L_Omega) * z;
}

model {
  to_vector(z) ~ std_normal();
  L_Omega ~ lkj_corr_cholesky(2);
  ...
}

The data variable $u$ was originally an array of vectors, which is efficient for access; here it is redeclared as a matrix in order to use it in matrix arithmetic. Moreover, it is transposed, along with $gamma$ and $beta$, to minimize the number of transposition operations. The new parameter $L_Omega$ is the Cholesky factor of the original correlation matrix $Omega$, so that

$Omega = L_Omega * L_Omega'$

The prior scale vector $tau$ is unchanged, and furthermore, pre-multiplying the Cholesky factor by the scale produces the Cholesky factor of the final covariance matrix,

$Sigma_beta$

$= quad_form_diag(Omega, tau)$

$= diag_pre_multiply(tau, L_Omega) * diag_pre_multiply(tau, L_Omega)'$

where the diagonal pre-multiply compound operation is defined by

$diag_pre_multiply(a, b) = diag_matrix(a) * b$

The new variable $z$ is declared as a matrix, the entries of which are given independent standard normal priors; the to_vector operation turns the matrix into a vector so that it can be used as a vectorized argument to the univariate normal density. This
results in every column of \( z \) being a \( K \)-variate normal random vector with the identity as covariance matrix. Therefore, multiplying \( z \) by the Cholesky factor of the covariance matrix and adding the mean \((u \ast \text{gamma})'\) produces a beta distributed as in the original model, where the variance is, letting \( L = \text{diag}(\tau) \Omega_L \),

\[
\nabla[\beta] = \mathbb{E}\left((Lz)(Lz)^\top\right) \\
= \mathbb{E}\left((Lz)^\top Lz\right) \\
= L \mathbb{E}(z z^\top) L^\top \\
= LL^\top = (\text{diag}(\tau) \Omega_L) (\text{diag}(\tau) \Omega_L)^\top \\
= \text{diag}(\tau) \Omega \text{diag}(\tau) \\
= \Sigma.
\]

Where we have used the linearity of expectations (line 2 to 3), the definition of \( \Omega = \Omega_L \Omega_L^\top \), and the fact that \( \mathbb{E}(z z^\top) = I \) since \( z \sim \mathcal{N}(0, I) \).

Omitting the remaining data declarations, which are the same as before with the exception of \( u \), the optimized model is as follows.

parameters {
  matrix[K, J] z;
  cholesky_factor_corr[K] L_Omega;
  vector<lower=0,upper=pi()/2>[K] tau_unif; // prior scale
  matrix[K, L] gamma; // group coeffs
  real<lower=0> sigma; // prediction error scale
}
transformed parameters {
  vector<lower=0>[K] tau = 2.5 * tan(tau_unif);
  matrix[K, J] beta = gamma * u + diag_pre_multiply(tau, L_Omega) * z;
}
model {
  vector[N] mu;
  for(n in 1:N) {
    mu[n] = x[n, ] * beta[, jj[n]];
  }
  to_vector(z) ~ std_normal();
  L_Omega ~ lkj_corr_cholesky(2);
  to_vector(gamma) ~ normal(0, 5);
  y ~ normal(mu, sigma);
}
This model also reparameterizes the prior scale $\tau$ to avoid potential problems with the heavy tails of the Cauchy distribution. The statement $\tau_{\text{unif}} \sim \text{uniform}(0, \pi() / 2)$ can be omitted from the model block because Stan increments the log posterior for parameters with uniform priors without it.

1.14. Prediction, forecasting, and backcasting

Stan models can be used for “predicting” the values of arbitrary model unknowns. When predictions are about the future, they’re called “forecasts;” when they are predictions about the past, as in climate reconstruction or cosmology, they are sometimes called “backcasts” (or “aftcasts” or “hindcasts” or “antecasts,” depending on the author’s feelings about the opposite of “fore”).

**Programming predictions**

As a simple example, the following linear regression provides the same setup for estimating the coefficients $\beta$ as in our very first example, using $y$ for the $N$ observations and $x$ for the $N$ predictor vectors. The model parameters and model for observations are exactly the same as before.

To make predictions, we need to be given the number of predictions, $N_{\text{new}}$, and their predictor matrix, $x_{\text{new}}$. The predictions themselves are modeled as a parameter $y_{\text{new}}$. The model statement for the predictions is exactly the same as for the observations, with the new outcome vector $y_{\text{new}}$ and prediction matrix $x_{\text{new}}$.

```stan
data {
    int<lower=1> K;
    int<lower=0> N;
    matrix[N, K] x;
    vector[N] y;

    int<lower=0> N_new;
    matrix[N_new, K] x_new;
}
parameters {
    vector[K] beta;
    real<lower=0> sigma;

    vector[N_new] y_new; // predictions
}
model {
    y ~ normal(x * beta, sigma); // observed model
    y_new ~ normal(x_new * beta, sigma); // predictions
}
```

Predictions as generated quantities
Where possible, the most efficient way to generate predictions is to use the generated quantities block. This provides proper Monte Carlo (not Markov chain Monte Carlo) inference, which can have a much higher effective sample size per iteration.

...data as above...

parameters {
  vector[K] beta;
  real<lower=0> sigma;
}
model {
  y ~ normal(x * beta, sigma);
}
generated quantities {
  vector[N_new] y_new;
  for (n in 1:N_new)
    y_new[n] = normal_rng(x_new[n] * beta, sigma);
}

Now the data are just as before, but the parameter \( y_{\text{new}} \) is now declared as a generated quantity, and the prediction model is removed from the model and replaced by a pseudo-random draw from a normal distribution.

Overflow in generated quantities
It is possible for values to overflow or underflow in generated quantities. The problem is that if the result is NaN, then any constraints placed on the variables will be violated. It is possible to check a value assigned by an RNG and reject it if it overflows, but this is both inefficient and leads to biased posterior estimates. Instead, the conditions causing overflow, such as trying to generate a negative binomial random variate with a mean of \( 2^{31} \), must be intercepted and dealt with. This is typically done by reparameterizing or reimplementing the random number generator using real values rather than integers, which are upper-bounded by \( 2^{31} - 1 \) in Stan.

1.15. Multivariate outcomes
Most regressions are set up to model univariate observations (be they scalar, boolean, categorical, ordinal, or count). Even multinomial regressions are just repeated
categorical regressions. In contrast, this section discusses regression when each observed value is multivariate. To relate multiple outcomes in a regression setting, their error terms are provided with covariance structure.

This section considers two cases, seemingly unrelated regressions for continuous multivariate quantities and multivariate probit regression for boolean multivariate quantities.

**Seemingly unrelated regressions**
The first model considered is the “seemingly unrelated” regressions (SUR) of econometrics where several linear regressions share predictors and use a covariance error structure rather than independent errors (Zellner 1962; Greene 2011).

The model is easy to write down as a regression,

\[ y_n = x_n \beta + \epsilon_n \quad \epsilon_n \sim \text{multivariate normal}(0, \Sigma) \]

where \( x_n \) is a \( J \)-row-vector of predictors (\( x \) is an \( N \times J \) matrix), \( y_n \) is a \( K \)-vector of observations, \( \beta \) is a \( K \times J \) matrix of regression coefficients (vector \( \beta_k \) holds coefficients for outcome \( k \)), and \( \Sigma \) is covariance matrix governing the error. As usual, the intercept can be rolled into \( x \) as a column of ones.

The basic Stan code is straightforward (though see below for more optimized code for use with LKJ priors on correlation).

```stan
data {
  int<lower=1> K;
  int<lower=1> J;
  int<lower=0> N;
  vector[J] x[N];
  vector[K] y[N];
}
parameters {
  matrix[K, J] beta;
  cov_matrix[K] Sigma;
}
model {
  vector[K] mu[N];
  for (n in 1:N)
    mu[n] = beta * x[n];
  y ~ multi_normal(mu, Sigma);
}
```
For efficiency, the multivariate normal is vectorized by precomputing the array of mean vectors and sharing the same covariance matrix.

Following the advice in the multivariate hierarchical priors section, we will place a weakly informative normal prior on the regression coefficients, an LKJ prior on the correlations and a half-Cauchy prior on standard deviations. The covariance structure is parameterized in terms of Cholesky factors for efficiency and arithmetic stability.

```plaintext
parameters {
  matrix[K, J] beta;
  cholesky_factor_corr[K] L_Omega;
  vector<lower=0>[K] L_sigma;
}

model {
  vector[K] mu[N];
  matrix[K, K] L_Sigma;

  for (n in 1:N)
    mu[n] = beta * x[n];

  L_Sigma = diag_pre_multiply(L_sigma, L_Omega);

  to_vector(beta) ~ normal(0, 5);
  L_Omega ~ lkj_corr_cholesky(4);
  L_sigma ~ cauchy(0, 2.5);

  y ~ multi_normal_cholesky(mu, L_Sigma);
}
```

The Cholesky factor of the covariance matrix is then reconstructed as a local variable and used in the model by scaling the Cholesky factor of the correlation matrices. The regression coefficients get a prior all at once by converting the matrix beta to a vector.

If required, the full correlation or covariance matrices may be reconstructed from their Cholesky factors in the generated quantities block.
Multivariate probit regression

The multivariate probit model generates sequences of boolean variables by applying a step function to the output of a seemingly unrelated regression.

The observations $y_n$ are $D$-vectors of boolean values (coded 0 for false, 1 for true). The values for the observations $y_n$ are based on latent values $z_n$ drawn from a seemingly unrelated regression model (see the previous section),

$$z_n = x_n \beta + \epsilon_n$$
$$\epsilon_n \sim \text{multivariate normal}(0, \Sigma)$$

These are then put through the step function to produce a $K$-vector $z_n$ of boolean values with elements defined by

$$y_{n,k} = I(z_{n,k} > 0),$$

where $I()$ is the indicator function taking the value 1 if its argument is true and 0 otherwise.

Unlike in the seemingly unrelated regressions case, here the covariance matrix $\Sigma$ has unit standard deviations (i.e., it is a correlation matrix). As with ordinary probit and logistic regressions, letting the scale vary causes the model (which is defined only by a cutpoint at 0, not a scale) to be unidentified (see Greene (2011)).

Multivariate probit regression can be coded in Stan using the trick introduced by Albert and Chib (1993), where the underlying continuous value vectors $y_n$ are coded as truncated parameters. The key to coding the model in Stan is declaring the latent vector $z$ in two parts, based on whether the corresponding value of $y$ is 0 or 1. Otherwise, the model is identical to the seemingly unrelated regression model in the previous section.

First, we introduce a sum function for two-dimensional arrays of integers; this is going to help us calculate how many total 1 values there are in $y$.

```c++
functions {
  int sum2d(int[,] a) {
    int s = 0;
    for (i in 1:size(a))
      s += sum(a[i]);
    return s;
  }
}
```
The function is trivial, but it’s not a built-in for Stan and it’s easier to understand the rest of the model if it’s pulled into its own function so as not to create a distraction.

The data declaration block is much like for the seemingly unrelated regressions, but the observations $y$ are now integers constrained to be 0 or 1.

```stan
data {
  int<lower=1> K;
  int<lower=1> D;
  int<lower=0> N;
  int<lower=0,upper=1> y[N,D];
  vector[K] x[N];
}
```

After declaring the data, there is a rather involved transformed data block whose sole purpose is to sort the data array $y$ into positive and negative components, keeping track of indexes so that $z$ can be easily reassembled in the transformed parameters block.

```stan
transformed data {
  int<lower=0> N_pos;
  int<lower=1,upper=N> n_pos[sum2d(y)];
  int<lower=1,upper=D> d_pos[size(n_pos)];
  int<lower=0> N_neg;
  int<lower=1,upper=N> n_neg[(N * D) - size(n_pos)];
  int<lower=1,upper=D> d_neg[size(n_neg)];

  N_pos = size(n_pos);
  N_neg = size(n_neg);
  {
    int i;
    int j;
    i = 1;
    j = 1;
    for (n in 1:N) {
      for (d in 1:D) {
        if (y[n,d] == 1) {
          n_pos[i] = n;
          d_pos[i] = d;
          i += 1;
        } else {
```

The variables $N_{\text{pos}}$ and $N_{\text{neg}}$ are set to the number of true (1) and number of false (0) observations in $y$. The loop then fills in the sequence of indexes for the positive and negative values in four arrays.

The parameters are declared as follows.

```stan
parameters {
  matrix[D, K] beta;
  cholesky_factor_corr[D] L_Omega;
  vector<lower=0>[N_{\text{pos}}] z_{\text{pos}};
  vector<upper=0>[N_{\text{neg}}] z_{\text{neg}};
}
```

These include the regression coefficients $\beta$ and the Cholesky factor of the correlation matrix, $L_\Omega$. This time there is no scaling because the covariance matrix has unit scale (i.e., it is a correlation matrix; see above).

The critical part of the parameter declaration is that the latent real value $z$ is broken into positive-constrained and negative-constrained components, whose size was conveniently calculated in the transformed data block. The transformed data block's real work was to allow the transformed parameter block to reconstruct $z$.

```stan
transformed parameters {
  vector[D] z[N];
  for (n in 1:N_{\text{pos}})
    z[n_{\text{pos}}[n], d_{\text{pos}}[n]] = z_{\text{pos}}[n];
  for (n in 1:N_{\text{neg}})
    z[n_{\text{neg}}[n], d_{\text{neg}}[n]] = z_{\text{neg}}[n];
}
```

At this point, the model is simple, pretty much recreating the seemingly unrelated regression.

```stan
model {
  

```
L_Omega ~ lkj_corr_cholesky(4);
to_vector(beta) ~ normal(0, 5);
{
  vector[D] beta_x[N];
  for (n in 1:N)
    beta_x[n] = beta * x[n];
  z ~ multi_normal_cholesky(beta_x, L_Omega);
}
}

This simple form of model is made possible by the Albert and Chib-style constraints on z.

Finally, the correlation matrix itself can be put back together in the generated quantities block if desired.

generated quantities {
  corr_matrix[D] Omega;
  Omega = multiply_lower_tri_self_transpose(L_Omega);
}

The same could be done for the seemingly unrelated regressions in the previous section.

1.16. Applications of pseudorandom number generation

The main application of pseudorandom number generator (PRNGs) is for posterior inference, including prediction and posterior predictive checks. They can also be used for pure data simulation, which is like a posterior predictive check with no conditioning. See the function reference manual for a complete description of the syntax and usage of pseudorandom number generators.

Prediction

Consider predicting unobserved outcomes using linear regression. Given predictors $x_1, \ldots, x_N$ and observed outcomes $y_1, \ldots, y_N$, and assuming a standard linear regression with intercept $\alpha$, slope $\beta$, and error scale $\sigma$, along with improper uniform priors, the posterior over the parameters given $x$ and $y$ is

$$p(\alpha, \beta, \sigma \mid x, y) \propto \prod_{n=1}^{N} \text{normal}(y_n \mid \alpha + \beta x_n, \sigma).$$

For this model, the posterior predictive inference for a new outcome $\tilde{y}_m$ given a
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predictor \( \bar{x}_m \), conditioned on the observed data \( x \) and \( y \), is

\[
p(\bar{y}_n | \bar{x}_n, x, y) = \int_{(\alpha, \beta, \sigma)} \text{normal}(\bar{y}_n | \alpha + \beta \bar{x}_n, \sigma) \times p(\alpha, \beta, \sigma | x, y) \, d(\alpha, \beta, \sigma).
\]

To code the posterior predictive inference in Stan, a standard linear regression is combined with a random number in the generated quantities block.

data {
  int<lower=0> N;
  vector[N] y;
  vector[N] x;
  int<lower=0> N_tilde;
  vector[N_tilde] x_tilde;
}
parameters {
  real alpha;
  real beta;
  real<lower=0> sigma;
}
model {
  y ~ normal(alpha + beta * x, sigma);
}
generated quantities {
  vector[N_tilde] y_tilde;
  for (n in 1:N_tilde)
    y_tilde[n] = normal_rng(alpha + beta * x_tilde[n], sigma);
}

Given observed predictors \( x \) and outcomes \( y \), \( y_{\text{tilde}} \) will be drawn according to \( p(\bar{y} | \bar{x}, y, x) \). This means that, for example, the posterior mean for \( y_{\text{tilde}} \) is the estimate of the outcome that minimizes expected square error (conditioned on the data and model).

**Posterior predictive checks**

A good way to investigate the fit of a model to the data, a critical step in Bayesian data analysis, is to generate simulated data according to the parameters of the model. This is carried out with exactly the same procedure as before, only the observed data predictors \( x \) are used in place of new predictors \( \bar{x} \) for unobserved outcomes. If the model fits the data well, the predictions for \( \bar{y} \) based on \( x \) should match the observed data \( y \).
To code posterior predictive checks in Stan requires only a slight modification of the prediction code to use $x$ and $N$ in place of $\tilde{x}$ and $\tilde{N}$,

```stan
generated quantities {
  vector[N] y_tilde;
  for (n in 1:N)
    y_tilde[n] = normal_rng(alpha + beta * x[n], sigma);
}
```

Andrew Gelman et al. (2013) recommend choosing several posterior draws $\tilde{y}^{(1)}, \ldots, \tilde{y}^{(M)}$ and plotting each of them alongside the data $y$ that was actually observed. If the model fits well, the simulated $\tilde{y}$ will look like the actual data $y$. 
2. Time-Series Models

Times series data come arranged in temporal order. This chapter presents two kinds of time series models, regression-like models such as autoregressive and moving average models, and hidden Markov models.

The Gaussian processes chapter presents Gaussian processes, which may also be used for time-series (and spatial) data.

2.1. Autoregressive models

A first-order autoregressive model (AR(1)) with normal noise takes each point $y_n$ in a sequence $y$ to be generated according to

$$y_n \sim \text{normal}(\alpha + \beta y_{n-1}, \sigma).$$

That is, the expected value of $y_n$ is $\alpha + \beta y_{n-1}$, with noise scaled as $\sigma$.

**AR(1) models**

With improper flat priors on the regression coefficients $\alpha$ and $\beta$ and on the positively-constrained noise scale ($\sigma$), the Stan program for the AR(1) model is as follows.\(^1\)

```stan
data {
  int<lower=0> N;
  vector[N] y;
}
parameters {
  real alpha;
  real beta;
  real<lower=0> sigma;
}
model {
  for (n in 2:N)
    y[n] ~ normal(alpha + beta * y[n-1], sigma);
}
```

The first observed data point, $y[1]$, is not modeled here because there is nothing to condition on; instead, it acts to condition $y[2]$. This model also uses an improper

\(^1\)The intercept in this model is $\alpha/(1 - \beta)$. An alternative parameterization in terms of an intercept $\gamma$ suggested Mark Scheuerell on GitHub is $y_n \sim \text{normal}(\gamma + \beta \cdot (y_{n-1} - \gamma), \sigma)$. 

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prior for \( \sigma \), but there is no obstacle to adding an informative prior if information is available on the scale of the changes in \( y \) over time, or a weakly informative prior to help guide inference if rough knowledge of the scale of \( y \) is available.

**Slicing for efficiency**

Although perhaps a bit more difficult to read, a much more efficient way to write the above model is by slicing the vectors, with the model above being replaced with the one-liner

```
model {
    y[2:N] ~ normal(alpha + beta * y[1:(N - 1)], sigma);
}
```

The left-hand side slicing operation pulls out the last \( N - 1 \) elements and the right-hand side version pulls out the first \( N - 1 \).

**Extensions to the AR(1) model**

Proper priors of a range of different families may be added for the regression coefficients and noise scale. The normal noise model can be changed to a Student-\( t \) distribution or any other distribution with unbounded support. The model could also be made hierarchical if multiple series of observations are available.

To enforce the estimation of a stationary AR(1) process, the slope coefficient \( \beta \) may be constrained with bounds as follows.

```
real<lower=-1,upper=1> beta;
```

In practice, such a constraint is not recommended. If the data are not well fit by a stationary model it is best to know this. Stationary parameter estimates can be encouraged with a prior favoring values of \( \beta \) near zero.

**AR(2) models**

Extending the order of the model is also straightforward. For example, an AR(2) model could be coded with the second-order coefficient \( \gamma \) and the following model statement.

```
for (n in 3:N)
    y[n] ~ normal(alpha + beta*y[n-1] + gamma*y[n-2], sigma);
```

**AR(\( K \)) models**

A general model where the order is itself given as data can be coded by putting the coefficients in an array and computing the linear predictor in a loop.

```
data {
```
ARCH(1) models
Econometric and financial time-series models usually assume heteroscedasticity: they allow the scale of the noise terms defining the series to vary over time. The simplest such model is the autoregressive conditional heteroscedasticity (ARCH) model (Engle 1982). Unlike the autoregressive model AR(1), which modeled the mean of the series as varying over time but left the noise term fixed, the ARCH(1) model takes the scale of the noise terms to vary over time but leaves the mean term fixed. Models could be defined where both the mean and scale vary over time; the econometrics literature presents a wide range of time-series modeling choices.

The ARCH(1) model is typically presented as the following sequence of equations, where \( r_t \) is the observed return at time point \( t \) and \( \mu, \alpha_0, \) and \( \alpha_1 \) are unknown regression coefficient parameters.

\[
\begin{align*}
    r_t &= \mu + a_t \\
    a_t &= \sigma_t \epsilon_t \\
    \epsilon_t &\sim \text{normal}(0, 1) \\
    \sigma_t^2 &= \alpha_0 + \alpha_1 a_{t-1}^2
\end{align*}
\]

In order to ensure the noise terms \( \sigma_t^2 \) are positive, the scale coefficients are constrained to be positive, \( \alpha_0, \alpha_1 > 0 \). To ensure stationarity of the time series, the slope
is constrained to be less than one, i.e., $\alpha_1 < 1$.\footnote{In practice, it can be useful to remove the constraint to test whether a non-stationary set of coefficients provides a better fit to the data. It can also be useful to add a trend term to the model, because an unfitted trend will manifest as non-stationarity.}

The ARCH(1) model may be coded directly in Stan as follows.

```stan
data {
  int<lower=0> T; // number of time points
  real r[T]; // return at time t
}
parameters {
  real mu; // average return
  real<lower=0> alpha0; // noise intercept
  real<lower=0,upper=1> alpha1; // noise slope
}
model {
  for (t in 2:T)
    r[t] ~ normal(mu, sqrt(alpha0 + alpha1 * pow(r[t-1] - mu,2)));
}
```

The loop in the model is defined so that the return at time $t = 1$ is not modeled; the model in the next section shows how to model the return at $t = 1$. The model can be vectorized to be more efficient; the model in the next section provides an example.

### 2.2. Modeling temporal heteroscedasticity

A set of variables is homoscedastic if their variances are all the same; the variables are heteroscedastic if they do not all have the same variance. Heteroscedastic time-series models allow the noise term to vary over time.

**GARCH(1,1) models**

The basic generalized autoregressive conditional heteroscedasticity (GARCH) model, GARCH(1,1), extends the ARCH(1) model by including the squared previous difference in return from the mean at time $t - 1$ as a predictor of volatility at time $t$, defining

$$
\sigma_t^2 = \alpha_0 + \alpha_1 a_{t-1}^2 + \beta_1 \sigma_{t-1}^2.
$$

To ensure the scale term is positive and the resulting time series stationary, the coefficients must all satisfy $\alpha_0, \alpha_1, \beta_1 > 0$ and the slopes $\alpha_1 + \beta_1 < 1$.\footnote{In practice, it can be useful to remove the constraint to test whether a non-stationary set of coefficients provides a better fit to the data. It can also be useful to add a trend term to the model, because an unfitted trend will manifest as non-stationarity.}
CHAPTER 2. TIME-SERIES MODELS

real r[T];
real<lower=0> sigma1;
}
parameters {
  real mu;
  real<lower=0> alpha0;
  real<lower=0,upper=1> alpha1;
  real<lower=0,upper=(1-alpha1)> beta1;
}
transformed parameters {
  real<lower=0> sigma[T];
  sigma[1] = sigma1;
  for (t in 2:T)
    sigma[t] = sqrt(alpha0
               + alpha1 * pow(r[t-1] - mu, 2)
               + beta1 * pow(sigma[t-1], 2));
}
model {
  r ~ normal(mu, sigma);
}

To get the recursive definition of the volatility regression off the ground, the data declaration includes a non-negative value sigma1 for the scale of the noise at $t = 1$. The constraints are coded directly on the parameter declarations. This declaration is order-specific in that the constraint on beta1 depends on the value of alpha1.

A transformed parameter array of non-negative values sigma is used to store the scale values at each time point. The definition of these values in the transformed parameters block is where the regression is now defined. There is an intercept alpha0, a slope alpha1 for the squared difference in return from the mean at the previous time, and a slope beta1 for the previous noise scale squared. Finally, the whole regression is inside the sqrt function because Stan requires scale (deviation) parameters (not variance parameters) for the normal distribution.

With the regression in the transformed parameters block, the model reduces a single vectorized sampling statement. Because $r$ and $sigma$ are of length $T$, all of the data are modeled directly.

2.3. Moving average models

A moving average model uses previous errors as predictors for future outcomes. For a moving average model of order $Q$, MA($Q$), there is an overall mean parameter $\mu$
and regression coefficients $\theta_q$ for previous error terms. With $\epsilon_t$ being the noise at time $t$, the model for outcome $y_t$ is defined by

$$ y_t = \mu + \theta_1 \epsilon_{t-1} + \cdots + \theta_Q \epsilon_{t-Q} + \epsilon_t, $$

with the noise term $\epsilon_t$ for outcome $y_t$ modeled as normal,

$$ \epsilon_t \sim \text{normal}(0, \sigma). $$

In a proper Bayesian model, the parameters $\mu$, $\theta$, and $\sigma$ must all be given priors.

**MA(2) example**

An MA(2) model can be coded in Stan as follows.

```stan
data {
  int<lower=3> T; // number of observations
  vector[T] y; // observation at time T
}
parameters {
  real mu; // mean
  real<lower=0> sigma; // error scale
  vector[2] theta; // lag coefficients
}
transformed parameters {
  vector[T] epsilon; // error terms
  for (t in 3:T)
    epsilon[t] = (y[t] - mu
                  - theta[1] * epsilon[t - 1]
                  - theta[2] * epsilon[t - 2]);
}
model {
  mu ~ cauchy(0, 2.5);
  theta ~ cauchy(0, 2.5);
  sigma ~ cauchy(0, 2.5);
  for (t in 3:T)
    y[t] ~ normal(mu
                   + theta[1] * epsilon[t - 1]
                   + theta[2] * epsilon[t - 2],
                   sigma);
}
```
The error terms $\epsilon_t$ are defined as transformed parameters in terms of the observations and parameters. The definition of the sampling statement (defining the likelihood) follows the definition, which can only be applied to $y_n$ for $n > Q$. In this example, the parameters are all given Cauchy (half-Cauchy for $\sigma$) priors, although other priors can be used just as easily.

This model could be improved in terms of speed by vectorizing the sampling statement in the model block. Vectorizing the calculation of the $\epsilon_t$ could also be sped up by using a dot product instead of a loop.

**Vectorized MA(Q) model**

A general MA($Q$) model with a vectorized sampling probability may be defined as follows.

```stan
data {
  int<lower=0> Q; // num previous noise terms
  int<lower=3> T; // num observations
  vector[T] y; // observation at time t
}
parameters {
  real mu; // mean
  real<lower=0> sigma; // error scale
  vector[Q] theta; // error coeff, lag -t
}
transformed parameters {
  vector[T] epsilon; // error term at time t
  for (t in 1:T) {
    epsilon[t] = y[t] - mu;
    for (q in 1:min(t - 1, Q))
      epsilon[t] = epsilon[t] - theta[q] * epsilon[t - q];
  }
}
model {
  vector[T] eta;
  mu ~ cauchy(0, 2.5);
  theta ~ cauchy(0, 2.5);
  sigma ~ cauchy(0, 2.5);
  for (t in 1:T) {
    eta[t] = mu;
    for (q in 1:min(t - 1, Q))
      eta[t] = eta[t] + theta[q] * epsilon[t - q];
  }
}
```
Here all of the data are modeled, with missing terms just dropped from the regressions as in the calculation of the error terms. Both models converge quickly and mix well at convergence, with the vectorized model being faster (per iteration, not to converge—they compute the same model).

### 2.4. Autoregressive moving average models

Autoregressive moving-average models (ARMA), combine the predictors of the autoregressive model and the moving average model. An ARMA(1,1) model, with a single state of history, can be encoded in Stan as follows.

```stan
data {
  int<lower=1> T; // num observations
  real y[T]; // observed outputs
}
parameters {
  real mu; // mean coeff
  real phi; // autoregression coeff
  real theta; // moving avg coeff
  real<lower=0> sigma; // noise scale
}
model {
  vector[T] nu; // prediction for time t
  vector[T] err; // error for time t
  nu[1] = mu + phi * mu; // assume err[0] == 0
  for (t in 2:T) {
    nu[t] = mu + phi * y[t-1] + theta * err[t-1];
    err[t] = y[t] - nu[t];
  }
  mu ~ normal(0, 10); // priors
  phi ~ normal(0, 2);
  theta ~ normal(0, 2);
  sigma ~ cauchy(0, 5);
  err ~ normal(0, sigma); // likelihood
}
```

The data are declared in the same way as the other time-series regressions and the
parameters are documented in the code.

In the model block, the local vector nu stores the predictions and err the errors. These are computed similarly to the errors in the moving average models described in the previous section.

The priors are weakly informative for stationary processes. The likelihood only involves the error term, which is efficiently vectorized here.

Often in models such as these, it is desirable to inspect the calculated error terms. This could easily be accomplished in Stan by declaring err as a transformed parameter, then defining it the same way as in the model above. The vector nu could still be a local variable, only now it will be in the transformed parameter block.

Wayne Folta suggested encoding the model without local vector variables as follows.

```stan
model {
  real err;
  mu ~ normal(0, 10);
  phi ~ normal(0, 2);
  theta ~ normal(0, 2);
  sigma ~ cauchy(0, 5);
  err = y[1] - mu + phi * mu;
  err ~ normal(0, sigma);
  for (t in 2:T) {
    err = y[t] - (mu + phi * y[t-1] + theta * err);
    err ~ normal(0, sigma);
  }
}
```

This approach to ARMA models illustrates how local variables, such as err in this case, can be reused in Stan. Folta’s approach could be extended to higher order moving-average models by storing more than one error term as a local variable and reassigning them in the loop.

Both encodings are fast. The original encoding has the advantage of vectorizing the normal distribution, but it uses a bit more memory. A halfway point would be to vectorize just err.

**Identifiability and stationarity**

MA and ARMA models are not identifiable if the roots of the characteristic polynomial for the MA part lie inside the unit circle, so it’s necessary to add the following
constraint

real<lower = -1, upper = 1> theta;

When the model is run without the constraint, using synthetic data generated from
the model, the simulation can sometimes find modes for \((\theta, \phi)\) outside the
\([-1, 1]\) interval, which creates a multiple mode problem in the posterior and also
causes the NUTS tree depth to get large (often above 10). Adding the constraint
both improves the accuracy of the posterior and dramatically reduces the tree depth,
which speeds up the simulation considerably (typically by much more than an order
of magnitude).

Further, unless one thinks that the process is really non-stationary, it’s worth adding
the following constraint to ensure stationarity.

real<lower = -1, upper = 1> phi;

### 2.5. Stochastic volatility models

Stochastic volatility models treat the volatility (i.e., variance) of a return on an asset,
such as an option to buy a security, as following a latent stochastic process in discrete
time (Kim, Shephard, and Chib 1998). The data consist of mean corrected (i.e.,
centered) returns \(y_t\) on an underlying asset at \(T\) equally spaced time points. Kim et
al. formulate a typical stochastic volatility model using the following regression-like
equations, with a latent parameter \(h_t\) for the log volatility, along with parameters \(\mu\)
for the mean log volatility, and \(\phi\) for the persistence of the volatility term. The
variable \(\epsilon_t\) represents the white-noise shock (i.e., multiplicative error) on the asset
return at time \(t\), whereas \(\delta_t\) represents the shock on volatility at time \(t\).

\[
\begin{align*}
y_t &= \epsilon_t \exp(h_t/2) \\
h_{t+1} &= \mu + \phi(h_t - \mu) + \delta_t \sigma \\
h_1 &\sim \text{normal}\left(\mu, \frac{\sigma}{\sqrt{1 - \phi^2}}\right) \\
\epsilon_t &\sim \text{normal}(0, 1) \\
\delta_t &\sim \text{normal}(0, 1)
\end{align*}
\]

Rearranging the first line, \(\epsilon_t = y_t \exp(-h_t/2)\), allowing the sampling distribution for
\(y_t\) to be written as

\[
y_t \sim \text{normal}(0, \exp(h_t/2)).
\]

---

3This subsection is a lightly edited comment of Jonathan Gilligan’s on GitHub; see https://github.com/stan-dev/stan/issues/1617#issuecomment-160249142
The recurrence equation for \( h_{t+1} \) may be combined with the scaling and sampling of \( \delta_t \) to yield the sampling distribution

\[
h_t \sim \text{normal}(\mu + \phi(h_{t-1} - \mu), \sigma).
\]

This formulation can be directly encoded, as shown in the following Stan model.

```stan
data {
  int<lower=0> T; // # time points (equally spaced)
  vector[T] y; // mean corrected return at time t
}
parameters {
  real mu; // mean log volatility
  real<lower=-1,upper=1> phi; // persistence of volatility
  real<lower=0> sigma; // white noise shock scale
  vector[T] h; // log volatility at time t
}
model {
  phi ~ uniform(-1, 1);
  sigma ~ cauchy(0, 5);
  mu ~ cauchy(0, 10);
  h[1] ~ normal(mu, sigma / sqrt(1 - phi * phi));
  for (t in 2:T)
    h[t] ~ normal(mu + phi * (h[t - 1] - mu), sigma);
  for (t in 1:T)
    y[t] ~ normal(0, exp(h[t] / 2));
}
```

Compared to the Kim et al. formulation, the Stan model adds priors for the parameters \( \phi, \sigma, \) and \( \mu \). The shock terms \( \varepsilon_t \) and \( \delta_t \) do not appear explicitly in the model, although they could be calculated efficiently in a generated quantities block.

The posterior of a stochastic volatility model such as this one typically has high posterior variance. For example, simulating 500 data points from the above model with \( \mu = -1.02, \phi = 0.95, \) and \( \sigma = 0.25 \) leads to 95\% posterior intervals for \( \mu \) of \((-1.23, -0.54)\), for \( \phi \) of \((0.82, 0.98)\), and for \( \sigma \) of \((0.16, 0.38)\).

The samples using NUTS show a high degree of autocorrelation among the samples, both for this model and the stochastic volatility model evaluated in (Hoffman and Gelman 2014). Using a non-diagonal mass matrix provides faster convergence and more effective samples than a diagonal mass matrix, but will not scale to large values of \( T \).
It is relatively straightforward to speed up the effective samples per second generated by this model by one or more orders of magnitude. First, the sampling statements for return $y$ is easily vectorized to

$$y \sim \text{normal}(0, \exp(h / 2));$$

This speeds up the iterations, but does not change the effective sample size because the underlying parameterization and log probability function have not changed. Mixing is improved by by reparameterizing in terms of a standardized volatility, then rescaling. This requires a standardized parameter $h_{\text{std}}$ to be declared instead of $h$.

```plaintext
parameters {
  ...
  vector[T] h_std; // std log volatility time t
}
```

The original value of $h$ is then defined in a transformed parameter block.

```plaintext
transformed parameters {
  vector[T] h = h_std * sigma; // now $h \sim \text{normal}(0, \sigma)$
  h[1] /= sqrt(1 - phi * phi); // rescale $h[1]$
  h += mu;
  for (t in 2:T)
    h[t] += phi * (h[t-1] - mu);
}
```

The first assignment rescales $h_{\text{std}}$ to have a $\text{normal}(0, \sigma)$ distribution and temporarily assigns it to $h$. The second assignment rescales $h[1]$ so that its prior differs from that of $h[2]$ through $h[T]$. The next assignment supplies a $\mu$ offset, so that $h[2]$ through $h[T]$ are now distributed $\text{normal}(\mu, \sigma)$; note that this shift must be done after the rescaling of $h[1]$. The final loop adds in the moving average so that $h[2]$ through $h[T]$ are appropriately modeled relative to $\phi$ and $\mu$.

As a final improvement, the sampling statement for $h[1]$ and loop for sampling $h[2]$ to $h[T]$ are replaced with a single vectorized standard normal sampling statement.

```plaintext
model {
  ...
  h_std ~ std_normal();
}
```

Although the original model can take hundreds and sometimes thousands of iterations to converge, the reparameterized model reliably converges in tens of iterations. Mixing is also dramatically improved, which results in higher effective sample sizes per iteration. Finally, each iteration runs in roughly a quarter of the time of the original iterations.
2.6. Hidden Markov models

A hidden Markov model (HMM) generates a sequence of $T$ output variables $y_t$ conditioned on a parallel sequence of latent categorical state variables $z_t \in \{1, \ldots, K\}$. These “hidden” state variables are assumed to form a Markov chain so that $z_t$ is conditionally independent of other variables given $z_{t-1}$. This Markov chain is parameterized by a transition matrix $\theta$ where $\theta_k$ is a $K$-simplex for $k \in \{1, \ldots, K\}$. The probability of transitioning to state $z_t$ from state $z_{t-1}$ is

$$z_t \sim \text{categorical}(\theta_{z[t-1]}).$$

The output $y_t$ at time $t$ is generated conditionally independently based on the latent state $z_t$.

This section describes HMMs with a simple categorical model for outputs $y_t \in \{1, \ldots, V\}$. The categorical distribution for latent state $k$ is parameterized by a $V$-simplex $\phi_k$. The observed output $y_t$ at time $t$ is generated based on the hidden state indicator $z_t$ at time $t$,

$$y_t \sim \text{categorical}(\phi_{z[t]}).$$

In short, HMMs form a discrete mixture model where the mixture component indicators form a latent Markov chain.

**Supervised parameter estimation**

In the situation where the hidden states are known, the following naive model can be used to fit the parameters $\theta$ and $\phi$.

```stan
data {
  int<lower=1> K; // num categories
  int<lower=1> V; // num words
  int<lower=0> T; // num instances
  int<lower=1,upper=V> w[T]; // words
  int<lower=1,upper=K> z[T]; // categories
  vector<lower=0>[K] alpha; // transit prior
  vector<lower=0>[V] beta; // emit prior
}
parameters {
  simplex[K] theta[K]; // transit probs
  simplex[V] phi[K]; // emit probs
}
model {
  for (k in 1:K)
```
Explicit Dirichlet priors have been provided for $\theta_k$ and $\phi_k$; dropping these two statements would implicitly take the prior to be uniform over all valid simplexes.

**Start-state and end-state probabilities**

Although workable, the above description of HMMs is incomplete because the start state $z_1$ is not modeled (the index runs from 2 to $T$). If the data are conceived as a subsequence of a long-running process, the probability of $z_1$ should be set to the stationary state probabilities in the Markov chain. In this case, there is no distinct end to the data, so there is no need to model the probability that the sequence ends at $z_T$.

An alternative conception of HMMs is as models of finite-length sequences. For example, human language sentences have distinct starting distributions (usually a capital letter) and ending distributions (usually some kind of punctuation). The simplest way to model the sequence boundaries is to add a new latent state $K + 1$, generate the first state from a categorical distribution with parameter vector $\theta_{K+1}$, and restrict the transitions so that a transition to state $K + 1$ is forced to occur at the end of the sentence and is prohibited elsewhere.

**Calculating sufficient statistics**

The naive HMM estimation model presented above can be sped up dramatically by replacing the loops over categorical distributions with a single multinomial distribution.

The data are declared as before. The transformed data block computes the sufficient statistics for estimating the transition and emission matrices.

```r
transformed data {
  int<lower=0> trans[K, K];
  int<lower=0> emit[K, V];
  for (k1 in 1:K)
    for (k2 in 1:K)
      trans[k1, k2] = 0;
}
for (t in 2:T)
    trans[z[t - 1], z[t]] += 1;
for (k in 1:K)
    for (v in 1:V)
        emit[k,v] = 0;
for (t in 1:T)
    emit[z[t], w[t]] += 1;
}

The likelihood component of the model based on looping over the input is replaced with multinomials as follows.

model {
...
    for (k in 1:K)
        trans[k] ~ multinomial(theta[k]);
    for (k in 1:K)
        emit[k] ~ multinomial(phi[k]);
}

In a continuous HMM with normal emission probabilities could be sped up in the same way by computing sufficient statistics.

Analytic posterior
With the Dirichlet-multinomial HMM, the posterior can be computed analytically because the Dirichlet is the conjugate prior to the multinomial. The following example illustrates how a Stan model can define the posterior analytically. This is possible in the Stan language because the model only needs to define the conditional probability of the parameters given the data up to a proportion, which can be done by defining the (unnormalized) joint probability or the (unnormalized) conditional posterior, or anything in between.

The model has the same data and parameters as the previous models, but now computes the posterior Dirichlet parameters in the transformed data block.

transformed data {
    vector<lower=0>[K] alpha_post[K];
    vector<lower=0>[V] beta_post[K];
    for (k in 1:K)
        alpha_post[k] = alpha;
    for (t in 2:T)
        alpha_post[z[t-1], z[t]] += 1;
for (k in 1:K)
  beta_post[k] = beta;
for (t in 1:T)
  beta_post[z[t], w[t]] += 1;
}
The posterior can now be written analytically as follows.

model {
  for (k in 1:K)
    theta[k] ~ dirichlet(alpha_post[k]);
  for (k in 1:K)
    phi[k] ~ dirichlet(beta_post[k]);
}

**Semisupervised estimation**

HMMs can be estimated in a fully unsupervised fashion without any data for which latent states are known. The resulting posteriors are typically extremely multimodal. An intermediate solution is to use semisupervised estimation, which is based on a combination of supervised and unsupervised data. Implementing this estimation strategy in Stan requires calculating the probability of an output sequence with an unknown state sequence. This is a marginalization problem, and for HMMs, it is computed with the so-called forward algorithm.

In Stan, the forward algorithm is coded as follows. First, two additional data variable are declared for the unsupervised data.

data {
  ...
  int<lower=1> T_unsup; // num unsupervised items
  int<lower=1,upper=V> u[T_unsup]; // unsup words
  ...
}

The model for the supervised data does not change; the unsupervised data are handled with the following Stan implementation of the forward algorithm.

model {
  ...
  real acc[K];
  real gamma[T_unsup, K];
  for (k in 1:K)
    gamma[1, k] = log(phi[k, u[1]]);
  for (t in 2:T_unsup) {
for (k in 1:K) {
    for (j in 1:K)
        acc[j] = gamma[t-1, j] + log(theta[j, k]) + log(phi[k, u[t]]);
    gamma[t, k] = log_sum_exp(acc);
}
}
target += log_sum_exp(gamma[T_unsup]);

The forward values gamma[t, k] are defined to be the log marginal probability of the inputs u[1],...,u[t] up to time t and the latent state being equal to k at time t; the previous latent states are marginalized out. The first row of gamma is initialized by setting gamma[1, k] equal to the log probability of latent state k generating the first output u[1]; as before, the probability of the first latent state is not itself modeled. For each subsequent time t and output j, the value acc[j] is set to the probability of the latent state at time t-1 being j, plus the log transition probability from state j at time t-1 to state k at time t, plus the log probability of the output u[t] being generated by state k. The log_sum_exp operation just multiplies the probabilities for each prior state j on the log scale in an arithmetically stable way.

The brackets provide the scope for the local variables acc and gamma; these could have been declared earlier, but it is clearer to keep their declaration near their use.

**Predictive inference**

Given the transition and emission parameters, \( \theta_{k,k'} \) and \( \phi_{k,v} \) and an observation sequence \( u_1,...,u_T \in \{1,...,V\} \), the Viterbi (dynamic programming) algorithm computes the state sequence which is most likely to have generated the observed output \( u \).

The Viterbi algorithm can be coded in Stan in the generated quantities block as follows. The predictions here is the most likely state sequence \( y_{\text{star}}[1], \ldots, y_{\text{star}}[T_{\text{unsup}}] \) underlying the array of observations \( u[1], \ldots, u[T_{\text{unsup}}] \). Because this sequence is determined from the transition probabilities theta and emission probabilities phi, it may be different from sample to sample in the posterior.

```stan
generated quantities {
    int<lower=1,upper=K> y_star[T_unsup];
    real log_p_y_star;
    {
        int back_ptr[T_unsup, K];
        real best_logp[T_unsup, K];
```
real best_total_logp;
for (k in 1:K)
    best_logp[1, k] = log(phi[k, u[1]]);
for (t in 2:T_unsup) {
    for (k in 1:K) {
        best_logp[t, k] = negative_infinity();
        for (j in 1:K) {
            real logp;
            logp = best_logp[t-1, j]
                + log(theta[j, k]) + log(phi[k, u[t]]);
            if (logp > best_logp[t, k]) {
                back_ptr[t, k] = j;
                best_logp[t, k] = logp;
            }
        }
    }
}
log_p_y_star = max(best_logp[T_unsup]);
for (k in 1:K)
    if (best_logp[T_unsup, k] == log_p_y_star)
        y_star[T_unsup] = k;
for (t in 1:(T_unsup - 1))
    y_star[T_unsup - t] = back_ptr[T_unsup - t + 1,
                                  y_star[T_unsup - t + 1]];
}

The bracketed block is used to make the three variables back_ptr, best_logp, and best_total_logp local so they will not be output. The variable y_star will hold the label sequence with the highest probability given the input sequence u. Unlike the forward algorithm, where the intermediate quantities were total probability, here they consist of the maximum probability best_logp[t, k] for the sequence up to time t with final output category k for time t, along with a backpointer to the source of the link. Following the backpointers from the best final log probability for the final time t yields the optimal state sequence.

This inference can be run for the same unsupervised outputs u as are used to fit the semisupervised model. The above code can be found in the same model file as the unsupervised fit. This is the Bayesian approach to inference, where the data being reasoned about is used in a semisupervised way to train the model. It is not
“cheating” because the underlying states for $u$ are never observed — they are just estimated along with all of the other parameters.

If the outputs $u$ are not used for semisupervised estimation but simply as the basis for prediction, the result is equivalent to what is represented in the BUGS modeling language via the cut operation. That is, the model is fit independently of $u$, then those parameters used to find the most likely state to have generated $u$. 
3. Missing Data and Partially Known Parameters

Bayesian inference supports a general approach to missing data in which any missing data item is represented as a parameter that is estimated in the posterior (Andrew Gelman et al. 2013). If the missing data are not explicitly modeled, as in the predictors for most regression models, then the result is an improper prior on the parameter representing the missing predictor.

Mixing arrays of observed and missing data can be difficult to include in Stan, partly because it can be tricky to model discrete unknowns in Stan and partly because unlike some other statistical languages (for example, R and Bugs), Stan requires observed and unknown quantities to be defined in separate places in the model. Thus it can be necessary to include code in a Stan program to splice together observed and missing parts of a data structure. Examples are provided later in the chapter.

3.1. Missing data

Stan treats variables declared in the data and transformed data blocks as known and the variables in the parameters block as unknown.

An example involving missing normal observations could be coded as follows.¹

```stan
data {
  int<lower=0> N_obs;
  int<lower=0> N_mis;
  real y_obs[N_obs];
}
parameters {
  real mu;
  real<lower=0> sigma;
  real y_mis[N_mis];
}
model {
  y_obs ~ normal(mu, sigma);
  y_mis ~ normal(mu, sigma);
}
```

¹A more meaningful estimation example would involve a regression of the observed and missing observations using predictors that were known for each and specified in the data block.
The number of observed and missing data points are coded as data with non-negative integer variables $N_{\text{obs}}$ and $N_{\text{mis}}$. The observed data are provided as an array data variable $y_{\text{obs}}$. The missing data are coded as an array parameter, $y_{\text{mis}}$. The ordinary parameters being estimated, the location $\mu$ and scale $\sigma$, are also coded as parameters. The model is vectorized on the observed and missing data; combining them in this case would be less efficient because the data observations would be promoted and have needless derivatives calculated.

### 3.2. Partially known parameters

In some situations, such as when a multivariate probability function has partially observed outcomes or parameters, it will be necessary to create a vector mixing known (data) and unknown (parameter) values. This can be done in Stan by creating a vector or array in the transformed parameters block and assigning to it.

The following example involves a bivariate covariance matrix in which the variances are known, but the covariance is not.

data {
  int<lower=0> N;
  vector[2] y[N];
  real<lower=0> var1;
  real<lower=0> var2;
}

transformed data {
  real<lower=0> max_cov = sqrt(var1 * var2);
  real<upper=0> min_cov = -max_cov;
}

parameters {
  vector[2] mu;
  real<lower=min_cov, upper=max_cov> cov;
}

transformed parameters {
  matrix[2, 2] Sigma;
  Sigma[1, 1] = var1; Sigma[1, 2] = cov;
  Sigma[2, 1] = cov; Sigma[2, 2] = var2;
}

model {
  y ~ multi_normal(mu, Sigma);
}
The variances are defined as data in variables `var1` and `var2`, whereas the covariance is defined as a parameter in variable `cov`. The $2 \times 2$ covariance matrix $\Sigma$ is defined as a transformed parameter, with the variances assigned to the two diagonal elements and the covariance to the two off-diagonal elements.

The constraint on the covariance declaration ensures that the resulting covariance matrix $\sigma$ is positive definite. The bound, plus or minus the square root of the product of the variances, is defined as transformed data so that it is only calculated once.

The vectorization of the multivariate normal is critical for efficiency here. The transformed parameter $\Sigma$ could be defined as a local variable within the model block if it does not need to be included in the sampler’s output.

### 3.3. Sliced missing data

If the missing data are part of some larger data structure, then it can often be effectively reassembled using index arrays and slicing. Here’s an example for time-series data, where only some entries in the series are observed.

```stan
data {
  int<lower = 0> N_obs;
  int<lower = 0> N_mis;
  int<lower = 1, upper = N_obs + N_mis> ii_obs[N_obs];
  int<lower = 1, upper = N_obs + N_mis> ii_mis[N_mis];
  real y_obs[N_obs];
}
transformed data {
  int<lower = 0> N = N_obs + N_mis;
}
parameters {
  real y_mis[N_mis];
  real<lower=0> sigma;
}
transformed parameters {
  real y[N];
  y[ii_obs] = y_obs;
  y[ii_mis] = y_mis;
}
model {
  sigma ~ gamma(1, 1);
  y[1] ~ normal(0, 100);
}
The index arrays \( ii_{\text{obs}} \) and \( ii_{\text{mis}} \) contain the indexes into the final array \( y \) of the observed data (coded as a data vector \( y_{\text{obs}} \)) and the missing data (coded as a parameter vector \( y_{\text{mis}} \)). See the time series chapter for further discussion of time-series model and specifically the autoregression section for an explanation of the vectorization for \( y \) as well as an explanation of how to convert this example to a full AR(1) model. To ensure \( y[1] \) has a proper posterior in case it is missing, we have given it an explicit, albeit broad, prior.

Another potential application would be filling the columns of a data matrix of predictors for which some predictors are missing; matrix columns can be accessed as vectors and assigned the same way, as in

\[
\begin{align*}
x[N_{\text{obs}}_2, 2] &= x_{\text{obs}}_2; \\
x[N_{\text{mis}}_2, 2] &= x_{\text{mis}}_2;
\end{align*}
\]

where the relevant variables are all hard coded with index 2 because Stan doesn’t support ragged arrays. These could all be packed into a single array with more fiddly indexing that slices out vectors from longer vectors (see the ragged data structures section for a general discussion of coding ragged data structures in Stan).

### 3.4. Loading matrix for factor analysis

Rick Farouni, on the Stan users group, inquired as to how to build a Cholesky factor for a covariance matrix with a unit diagonal, as used in Bayesian factor analysis (Aguilar and West 2000). This can be accomplished by declaring the below-diagonal elements as parameters, then filling the full matrix as a transformed parameter.

```stan
data {
  int<lower=2> K;
}
transformed data {
  int<lower=1> K_choose_2;
  K_choose_2 = (K * (K - 1)) / 2;
}
parameters {
  vector[K_choose_2] L_lower;
}
transformed parameters {
  cholesky_factor_cov[K] L;
  for (k in 1:K)
```
\[ L[k, k] = 1; \]
\[
\begin{align*}
\text{int } & i; \\
\text{for (m in 2:K) } \{ \\
& \text{for (n in 1:(m - 1)) } \{ \\
& \quad L[m, n] = L_{\text{lower}}[i]; \\
& \quad L[n, m] = 0; \\
& \quad i += 1; \\
& \} \\
& \}
\}
\]

It is most convenient to place a prior directly on \( L_{\text{lower}} \). An alternative would be a prior for the full Cholesky factor \( L \), because the transform from \( L_{\text{lower}} \) to \( L \) is just the identity and thus does not require a Jacobian adjustment (despite the warning from the parser, which is not smart enough to do the code analysis to infer that the transform is linear). It would not be at all convenient to place a prior on the full covariance matrix \( L * L' \), because that would require a Jacobian adjustment; the exact adjustment is detailed in the reference manual.

### 3.5. Missing multivariate data

It’s often the case that one or more components of a multivariate outcome are missing.\(^2\)

As an example, we’ll consider the bivariate distribution, which is easily marginalized. The coding here is brute force, representing both an array of vector observations \( y \) and a boolean array \( y_{\text{observed}} \) to indicate which values were observed (others can have dummy values in the input).

\[
\text{vector}[2] \ y[N]; \\
\text{int<lower=0, upper=1>} \ y_{\text{observed}}[N, 2];
\]

If both components are observed, we model them using the full multi-normal, otherwise we model the marginal distribution of the component that is observed.

\[
\begin{align*}
\text{for (n in 1:N) } \{ \\
& \quad \text{if (y_{\text{observed}}[n, 1] } && \text{ y_{\text{observed}}[n, 2])} \\
& \quad \quad \text{y}[n] \sim \text{multi_normal}(\mu, \Sigma);
\}
\]

\(^2\)This is not the same as missing components of a multivariate predictor in a regression problem; in that case, you will need to represent the missing data as a parameter and impute missing values in order to feed them into the regression.
else if (y_observed[n, 1])
    y[n, 1] ~ normal(mu[1], sqrt(Sigma[1, 1]));
else if (y_observed[n, 2])
    y[n, 2] ~ normal(mu[2], sqrt(Sigma[2, 2]));
}

It's a bit more work, but much more efficient to vectorize these sampling statements. In transformed data, build up three vectors of indices, for the three cases above:

```plaintext
transformed data {
    int ns12[observed_12(y_observed)];
    int ns1[observed_1(y_observed)];
    int ns2[observed_2(y_observed)];
}
```

You will need to write functions that pull out the count of observations in each of the three sampling situations. This must be done with functions because the result needs to go in top-level block variable size declaration. Then the rest of transformed data just fills in the values using three counters.

```plaintext
int n12 = 1;
int n1 = 1;
int n2 = 1;
for (n in 1:N) {
    if (y_observed[n, 1] && y_observed[n, 2]) {
        ns12[n12] = n;
        n12 += 1;
    } else if (y_observed[n, 1]) {
        ns1[n1] = n;
        n1 += 1;
    } else if (y_observed[n, 2]) {
        ns2[n2] = n;
        n2 += 1;
    }
}
```

Then, in the model block, everything is vectorizable using those indexes constructed once in transformed data:

```plaintext
y[ns12] ~ multi_normal(mu, Sigma);
y[ns1] ~ normal(mu[1], sqrt(Sigma[1, 1]));
y[ns2] ~ normal(mu[2], sqrt(Sigma[2, 2]));
```
The result will be much more efficient than using latent variables for the missing data, but it requires the multivariate distribution to be marginalized analytically. It’d be more efficient still to precompute the three arrays in the transformed data block, though the efficiency improvement will be relatively minor compared to vectorizing the probability functions.

This approach can easily be generalized with some index fiddling to the general multivariate case. The trick is to pull out entries in the covariance matrix for the missing components. It can also be used in situations such as multivariate differential equation solutions where only one component is observed, as in a phase-space experiment recording only time and position of a pendulum (and not recording momentum).
4. Truncated or Censored Data

Data in which measurements have been truncated or censored can be coded in Stan following their respective probability models.

4.1. Truncated distributions

Truncation in Stan is restricted to univariate distributions for which the corresponding log cumulative distribution function (CDF) and log complementary cumulative distribution (CCDF) functions are available. See the reference manual section on truncated distributions for more information on truncated distributions, CDFs, and CCDFs.

4.2. Truncated data

Truncated data are data for which measurements are only reported if they fall above a lower bound, below an upper bound, or between a lower and upper bound.

Truncated data may be modeled in Stan using truncated distributions. For example, suppose the truncated data are \( y_n \) with an upper truncation point of \( U = 300 \) so that \( y_n < 300 \). In Stan, this data can be modeled as following a truncated normal distribution for the observations as follows.

data {
  int<lower=0> N;
  real U;
  real<upper=U> y[N];
}
parameters {
  real mu;
  real<lower=0> sigma;
}
model {
  for (n in 1:N)
    y[n] ~ normal(mu, sigma) T[,U];
}

The model declares an upper bound \( U \) as data and constrains the data for \( y \) to respect the constraint; this will be checked when the data are loaded into the model before sampling begins.
This model implicitly uses an improper flat prior on the scale and location parameters; these could be given priors in the model using sampling statements.

**Constraints and out-of-bounds returns**

If the sampled variate in a truncated distribution lies outside of the truncation range, the probability is zero, so the log probability will evaluate to $-\infty$. For instance, if variate $y$ is sampled with the statement.

```plaintext
for (n in 1:N)
    y[n] ~ normal(mu, sigma) T[L,U];
```

then if the value of $y[n]$ is less than the value of $L$ or greater than the value of $U$, the sampling statement produces a zero-probability estimate. For user-defined truncation, this zeroing outside of truncation bounds must be handled explicitly.

To avoid variables straying outside of truncation bounds, appropriate constraints are required. For example, if $y$ is a parameter in the above model, the declaration should constrain it to fall between the values of $L$ and $U$.

```plaintext
parameters {
    real<lower=L, upper=U> y[N];
    ...
}
```

If in the above model, $L$ or $U$ is a parameter and $y$ is data, then $L$ and $U$ must be appropriately constrained so that all data are in range and the value of $L$ is less than that of $U$ (if they are equal, the parameter range collapses to a single point and the Hamiltonian dynamics used by the sampler break down). The following declarations ensure the bounds are well behaved.

```plaintext
parameters {
    real<upper=min(y)> L; // L < y[n]
    real<lower=max(L, max(y))> U; // L < U; y[n] < U
}
```

For pairs of real numbers, the function `fmax` is used rather than `max`.

**Unknown truncation points**

If the truncation points are unknown, they may be estimated as parameters. This can be done with a slight rearrangement of the variable declarations from the model in the previous section with known truncation points.

```plaintext
data {
    int<lower=1> N;
    real y[N];
}
```
parameters {
    real<upper = min(y)> L;
    real<lower = max(y)> U;
    real mu;
    real<lower=0> sigma;
}
model {
    L ~ ...;
    U ~ ...;
    for (n in 1:N)
        y[n] ~ normal(mu, sigma) T[L,U];
}

Here there is a lower truncation point $L$ which is declared to be less than or equal to the minimum value of $y$. The upper truncation point $U$ is declared to be larger than the maximum value of $y$. This declaration, although dependent on the data, only enforces the constraint that the data fall within the truncation bounds. With $N$ declared as type `int<lower=1>`, there must be at least one data point. The constraint that $L$ is less than $U$ is enforced indirectly, based on the non-empty data.

The ellipses where the priors for the bounds $L$ and $U$ should go should be filled in with a an informative prior in order for this model to not concentrate $L$ strongly around $\min(y)$ and $U$ strongly around $\max(y)$.

### 4.3. Censored data

Censoring hides values from points that are too large, too small, or both. Unlike with truncated data, the number of data points that were censored is known. The textbook example is the household scale which does not report values above 300 pounds.

**Estimating censored values**

One way to model censored data is to treat the censored data as missing data that is constrained to fall in the censored range of values. Since Stan does not allow unknown values in its arrays or matrices, the censored values must be represented explicitly, as in the following right-censored case.

data {
    int<lower=0> N_obs;
    int<lower=0> N_cens;
    real y_obs[N_obs];
    real<lower=max(y_obs)> U;
}
parameters {
  real<lower=U> y_cens[N_cens];
  real mu;
  real<lower=0> sigma;
}
model {
  y_obs ~ normal(mu, sigma);
  y_cens ~ normal(mu, sigma);
}

Because the censored data array y_cens is declared to be a parameter, it will be sampled along with the location and scale parameters mu and sigma. Because the censored data array y_cens is declared to have values of type real<lower=U>, all imputed values for censored data will be greater than U. The imputed censored data affects the location and scale parameters through the last sampling statement in the model.

**Integrating out censored values**

Although it is wrong to ignore the censored values in estimating location and scale, it is not necessary to impute values. Instead, the values can be integrated out. Each censored data point has a probability of

$$
\Pr[y > U] = \int_U^\infty \text{normal}(y | \mu, \sigma) \, dy \\
= 1 - \Phi \left( \frac{y - \mu}{\sigma} \right),
$$

where \( \Phi() \) is the standard normal cumulative distribution function. With \( M \) censored observations, the total probability on the log scale is

$$
\log \prod_{m=1}^M \Pr[y_m > U] = \log \left( 1 - \Phi \left( \frac{y - \mu}{\sigma} \right) \right)^M \\
= M \times \text{normal}_\text{lccdf}(y | \mu, \sigma),
$$

where \text{normal}_\text{lccdf} is the log of complementary CDF (Stan provides \text{<distr>_lccdf} for each distribution implemented in Stan).

The following right-censored model assumes that the censoring point is known, so it is declared as data.

data {

int<lower=0> N_obs;
int<lower=0> N_cens;
real y_obs[N_obs];
real<lower=max(y_obs)> U;
}
parameters {
  real mu;
  real<lower=0> sigma;
}
model {
  y_obs ~ normal(mu, sigma);
  target += N_cens * normal_lccdf(U | mu, sigma);
}

For the observed values in \( y_{\text{obs}} \), the normal sampling model is used without truncation. The log probability is directly incremented using the calculated log cumulative normal probability of the censored data items.

For the left-censored data the CDF (\texttt{normal_lccdf}) has to be used instead of complementary CDF. If the censoring point variable (\( L \)) is unknown, its declaration should be moved from the data to the parameters block.

data {
  int<lower=0> N_obs;
  int<lower=0> N_cens;
  real y_obs[N_obs];
}
parameters {
  real<upper=min(y_obs)> L;
  real mu;
  real<lower=0> sigma;
}
model {
  L ~ normal(mu, sigma);
  y_obs ~ normal(mu, sigma);
  target += N_cens * normal_lccdf(L | mu, sigma);
}
5. Finite Mixtures

Finite mixture models of an outcome assume that the outcome is drawn from one of several distributions, the identity of which is controlled by a categorical mixing distribution. Mixture models typically have multimodal densities with modes near the modes of the mixture components. Mixture models may be parameterized in several ways, as described in the following sections. Mixture models may be used directly for modeling data with multimodal distributions, or they may be used as priors for other parameters.

5.1. Relation to clustering
Clustering models, as discussed in the clustering chapter, are just a particular class of mixture models that have been widely applied to clustering in the engineering and machine-learning literature. The normal mixture model discussed in this chapter reappears in multivariate form as the statistical basis for the $K$-means algorithm; the latent Dirichlet allocation model, usually applied to clustering problems, can be viewed as a mixed-membership multinomial mixture model.

5.2. Latent discrete parameterization
One way to parameterize a mixture model is with a latent categorical variable indicating which mixture component was responsible for the outcome. For example, consider $K$ normal distributions with locations $\mu_k \in \mathbb{R}$ and scales $\sigma_k \in (0, \infty)$. Now consider mixing them in proportion $\lambda$, where $\lambda_k \geq 0$ and $\sum_{k=1}^{K} \lambda_k = 1$ (i.e., $\lambda$ lies in the unit $K$-simplex). For each outcome $y_n$ there is a latent variable $z_n$ in $\{1, \ldots, K\}$ with a categorical distribution parameterized by $\lambda$,

$$z_n \sim \text{categorical}(\lambda).$$

The variable $y_n$ is distributed according to the parameters of the mixture component $z_n$,

$$y_n \sim \text{normal}(\mu_{z[n]}, \sigma_{z[n]}).$$

This model is not directly supported by Stan because it involves discrete parameters $z_n$, but Stan can sample $\mu$ and $\sigma$ by summing out the $z$ parameter as described in the next section.
5.3. Summing out the responsibility parameter

To implement the normal mixture model outlined in the previous section in Stan, the discrete parameters can be summed out of the model. If \( Y \) is a mixture of \( K \) normal distributions with locations \( \mu_k \) and scales \( \sigma_k \) with mixing proportions \( \lambda \) in the unit \( K \)-simplex, then

\[
p_Y (y \mid \lambda, \mu, \sigma) = \sum_{k=1}^{K} \lambda_k \text{normal} \ (y \mid \mu_k, \sigma_k).
\]

**Log sum of exponentials: linear Sums on the log scale**

The log sum of exponentials function is used to define mixtures on the log scale. It is defined for two inputs by

\[
\log \_sum \_exp(a, b) = \log (\exp(a) + \exp(b)).
\]

If \( a \) and \( b \) are probabilities on the log scale, then \( \exp(a) + \exp(b) \) is their sum on the linear scale, and the outer log converts the result back to the log scale; to summarize, \( \log \_sum \_exp \) does linear addition on the log scale. The reason to use Stan’s built-in \( \log \_sum \_exp \) function is that it can prevent underflow and overflow in the exponentiation, by calculating the result as

\[
\log (\exp(a) + \exp(b)) = c + \log (\exp(a - c) + \exp(b - c)),
\]

where \( c = \max(a, b) \). In this evaluation, one of the terms, \( a - c \) or \( b - c \), is zero and the other is negative, thus eliminating the possibility of overflow or underflow in the leading term while extracting the most arithmetic precision possible by pulling the \( \max(a, b) \) out of the log-exp round trip.

For example, the mixture of \( \text{normal}(-1, 2) \) with \( \text{normal}(3, 1) \), with mixing proportion \( \lambda = [0.3, 0.7]^{\top} \), can be implemented in Stan as follows.

```stan
parameters {
  real y;
}
model {
  target += log_sum_exp(log(0.3) + normal_lpdf(y | -1, 2),
                        log(0.7) + normal_lpdf(y | 3, 1));
}
```
The log probability term is derived by taking

\[
\log p(y \mid \lambda, \mu, \sigma) = \log \left( 0.3 \times \text{normal}(y \mid -1, 2) + 0.7 \times \text{normal}(y \mid 3, 1) \right)
\]

\[
= \log \left( \exp \left( \log \left( 0.3 \times \text{normal}(y \mid -1, 2) \right) \right) + \exp \left( \log \left( 0.7 \times \text{normal}(y \mid 3, 1) \right) \right) \right)
\]

\[
= \log\text{sum}\_\text{exp}\left( \log(0.3) + \log \text{normal}(y \mid -1, 2), \log(0.7) + \log \text{normal}(y \mid 3, 1) \right).
\]

**Dropping uniform mixture ratios**

If a two-component mixture has a mixing ratio of 0.5, then the mixing ratios can be dropped, because

\[
\log\text{half} = \log(0.5);
\]

for (n in 1:N)

    target += \text{log\_sum\_exp}(\text{neg\_log\_half} + \text{normal\_lpdf}(y[n] \mid \mu[1], \sigma[1]), \log\_half + \text{normal\_lpdf}(y[n] \mid \mu[2], \sigma[2]));

then the \log0.5 term isn’t contributing to the proportional density, and the above can be replaced with the more efficient version

for (n in 1:N)

    target += \text{log\_sum\_exp}(\text{normal\_lpdf}(y[n] \mid \mu[1], \sigma[1]), \text{normal\_lpdf}(y[n] \mid \mu[2], \sigma[2]));

The same result holds if there are \(K\) components and the mixing simplex \(\lambda\) is symmetric, i.e.,

\[
\lambda = \left( \frac{1}{K}, \ldots, \frac{1}{K} \right).
\]

The result follows from the identity

\[
\text{log\_sum\_exp}(c + a, c + b) = c + \text{log\_sum\_exp}(a, b)
\]

and the fact that adding a constant \(c\) to the log density accumulator has no effect because the log density is only specified up to an additive constant in the first place. There is nothing specific to the normal distribution here; constants may always be dropped from the target.
CHAPTER 5. FINITE MIXTURES

Recovering posterior mixture proportions
The posterior $p(z_n \mid y_n, \mu, \sigma)$ over the mixture indicator $z_n \in 1:K$ is often of interest as $p(z_n = k \mid y, \mu, \sigma)$ is the posterior probability that that observation $y_n$ was generated by mixture component $k$. The posterior can be computed via Bayes’s rule,

$$
\Pr(z_n = k \mid y_n, \mu, \sigma, \lambda) \propto p(y_n \mid z_n = k, \mu, \sigma) \cdot p(z_n = k \mid \lambda) = \text{normal}(y_n \mid \mu_k, \sigma_k) \cdot \lambda_k.
$$

The normalization can be done via summation, because $z_n \in 1:K$ only takes on finitely many values. In detail,

$$
p(z_n = k \mid y_n, \mu, \sigma, \lambda) = \frac{p(y_n \mid z_n = k, \mu, \sigma) \cdot p(z_n = k \mid \lambda)}{\sum_{k'=1}^K p(y_n \mid z_n = k', \mu, \sigma) \cdot p(z_n = k' \mid \lambda)}.
$$

On the log scale, the normalized probability is computed as

$$
\log \Pr(z_n = k \mid y_n, \mu, \sigma, \lambda) = \log p(y_n \mid z_n = k, \mu, \sigma) + \log \Pr(z_n = k \mid \lambda) - \log \sum_{k'=1}^K \left( \log p(y_n \mid z_n = k', \mu, \sigma) + \log p(z_n = k' \mid \lambda) \right).
$$

This can be coded up directly in Stan; the change-point model in the change point section provides an example.

Estimating parameters of a mixture
Given the scheme for representing mixtures, it may be moved to an estimation setting, where the locations, scales, and mixture components are unknown. Further generalizing to a number of mixture components specified as data yields the following model.

data {
    int<lower=1> K; // number of mixture components
    int<lower=1> N; // number of data points
    real y[N]; // observations
}
parameters {
    simplex[K] theta; // mixing proportions
    ordered[K] mu; // locations of mixture components
    vector<lower=0>[K] sigma; // scales of mixture components
}
model {


CHAPTER 5. FINITE MIXTURES

vector[K] log_theta = log(theta); // cache log calculation
sigma ~ lognormal(0, 2);
mu ~ normal(0, 10);
for (n in 1:N) {
    vector[K] lps = log_theta;
    for (k in 1:K)
        lps[k] += normal_lpdf(y[n] | mu[k], sigma[k]);
    target += log_sum_exp(lps);
}

The model involves $K$ mixture components and $N$ data points. The mixing proportion parameter $\theta$ is declared to be a unit $K$-simplex, whereas the component location parameter $\mu$ and scale parameter $\sigma$ are both defined to be $K$-vectors.

The location parameter $\mu$ is declared to be an ordered vector in order to identify the model. This will not affect inferences that do not depend on the ordering of the components as long as the prior for the components $\mu[k]$ is symmetric, as it is here (each component has an independent normal(0, 10) prior). It would even be possible to include a hierarchical prior for the components.

The values in the scale array $\sigma$ are constrained to be non-negative, and have a weakly informative prior given in the model chosen to avoid zero values and thus collapsing components.

The model declares a local array variable $lps$ to be size $K$ and uses it to accumulate the log contributions from the mixture components. The main action is in the loop over data points $n$. For each such point, the log of $\theta_k \times \text{normal}(y_n | \mu_k, \sigma_k)$ is calculated and added to the array $lps$. Then the log probability is incremented with the log sum of exponentials of those values.

5.4. Vectorizing mixtures

There is (currently) no way to vectorize mixture models at the observation level in Stan. This section is to warn users away from attempting to vectorize naively, as it results in a different model. A proper mixture at the observation level is defined as follows, where we assume that $\lambda$, $y[n]$, $\mu[1]$, $\mu[2]$, and $\sigma[1]$, $\sigma[2]$ are all scalars and $\lambda$ is between 0 and 1.

for (n in 1:N) {
    target += log_sum_exp(log(lambda)
                        + normal_lpdf(y[n] | mu[1], sigma[1]),
                        log1m(lambda))

+ normal_lpdf(y[n] | mu[2], sigma[2]));

or equivalently

for (n in 1:N)
    target += log_mix(lambda,
        normal_lpdf(y[n] | mu[1], sigma[1]),
        normal_lpdf(y[n] | mu[2], sigma[2]));

This definition assumes that each observation $y_n$ may have arisen from either of the mixture components. The density is

$$p(y | \lambda, \mu, \sigma) = \prod_{n=1}^{N} \left( \lambda \times \text{normal}(y_n | \mu_1, \sigma_1) + (1 - \lambda) \times \text{normal}(y_n | \mu_2, \sigma_2) \right).$$

Contrast the previous model with the following (erroneous) attempt to vectorize the model.

target += log_sum_exp(log(lambda)
    + normal_lpdf(y | mu[1], sigma[1]),
    log1m(lambda)
    + normal_lpdf(y | mu[2], sigma[2]));

or equivalently,

target += log_mix(lambda,
    normal_lpdf(y | mu[1], sigma[1]),
    normal_lpdf(y | mu[2], sigma[2]));

This second definition implies that the entire sequence $y_1, \ldots, y_n$ of observations comes from one component or the other, defining a different density,

$$p(y | \lambda, \mu, \sigma) = \lambda \times \prod_{n=1}^{N} \text{normal}(y_n | \mu_1, \sigma_1) + (1 - \lambda) \times \prod_{n=1}^{N} \text{normal}(y_n | \mu_2, \sigma_2).$$

5.5. Inferences supported by mixtures

In many mixture models, the mixture components are underlyingly exchangeable in the model and thus not identifiable. This arises if the parameters of the mixture components have exchangeable priors and the mixture ratio gets a uniform prior so that the parameters of the mixture components are also exchangeable in the likelihood.
We have finessed this basic problem by ordering the parameters. This will allow us in some cases to pick out mixture components either ahead of time or after fitting (e.g., male vs. female, or Democrat vs. Republican).

In other cases, we do not care about the actual identities of the mixture components and want to consider inferences that are independent of indexes. For example, we might only be interested in posterior predictions for new observations.

**Mixtures with unidentifiable components**

As an example, consider the normal mixture from the previous section, which provides an exchangeable prior on the pairs of parameters \((\mu_1, \sigma_1)\) and \((\mu_2, \sigma_2)\),

\[
\begin{align*}
\mu_1, \mu_2 &\sim \text{normal}(0, 10) \\
\sigma_1, \sigma_2 &\sim \text{halfnormal}(0, 10)
\end{align*}
\]

The prior on the mixture ratio is uniform,

\[
\lambda \sim \text{uniform}(0, 1)
\]

so that with the likelihood

\[
p(y_n \mid \mu, \sigma) = \lambda \times \text{normal}(y_n \mid \mu_1, \sigma_1) + (1 - \lambda) \times \text{normal}(y_n \mid \mu_2, \sigma_2),
\]

the joint distribution \(p(y, \mu, \sigma, \lambda)\) is exchangeable in the parameters \((\mu_1, \sigma_1)\) and \((\mu_2, \sigma_2)\) with \(\lambda\) flipping to \(1 - \lambda\).\(^1\)

**Inference under label switching**

In cases where the mixture components are not identifiable, it can be difficult to diagnose convergence of sampling or optimization algorithms because the labels will switch, or be permuted, in different MCMC chains or different optimization runs. Luckily, posterior inferences which do not refer to specific component labels are invariant under label switching and may be used directly. This subsection considers a pair of examples.

**Predictive likelihood**

Predictive likelihood for a new observation \(\tilde{y}\) given the complete parameter vector \(\theta\) will be

\[
p(\tilde{y} \mid y) = \int_{\theta} p(\tilde{y} \mid \theta) p(\theta \mid y) \, d\theta.
\]

\(^1\)Imposing a constraint such as \(\theta < 0.5\) will resolve the symmetry, but fundamentally changes the model and its posterior inferences.
The normal mixture example from the previous section, with \( \theta = (\mu, \sigma, \lambda) \), shows that the likelihood returns the same density under label switching and thus the predictive inference is sound. In Stan, that predictive inference can be done either by computing \( p(\tilde{y} \mid y) \), which is more efficient statistically in terms of effective sample size, or simulating draws of \( \tilde{y} \), which is easier to plug into other inferences. Both approaches can be coded directly in the generated quantities block of the program. Here’s an example of the direct (non-sampling) approach.

```stan
data {
  int<lower = 0> N_tilde;
  vector[N_tilde] y_tilde;
  ...
}

generated quantities {
  vector[N_tilde] log_p_y_tilde;
  for (n in 1:N_tilde)
    log_p_y_tilde[n] = log_mix(lambda,
                               normal_lpdf(y_tilde[n] | mu[1], sigma[1])
                               normal_lpdf(y_tilde[n] | mu[2], sigma[2]));
}
```

It is a bit of a bother afterwards, because the logarithm function isn’t linear and hence doesn’t distribute through averages (Jensen’s inequality shows which way the inequality goes). The right thing to do is to apply `log_sum_exp` of the posterior draws of \( \log_p_y_tilde \). The average log predictive density is then given by subtracting \( \log(N_{\text{new}}) \).

**Clustering and similarity**

Often a mixture model will be applied to a clustering problem and there might be two data items \( y_i \) and \( y_j \) for which there is a question of whether they arose from the same mixture component. If we take \( z_i \) and \( z_j \) to be the component responsibility discrete variables, then the quantity of interest is \( z_i = z_j \), which can be summarized as an event probability

\[
\Pr[z_i = z_j \mid y] = \int \sum_{k=0}^{1} \sum_{m=0}^{1} \frac{p(z_i = k, z_j = k, y_i, y_j \mid \theta)}{p(\theta \mid y)} \, d\theta.
\]

As with other event probabilities, this can be calculated in the generated quantities block either by sampling \( z_i \) and \( z_j \) and using the indicator function on their equality, or by computing the term inside the integral as a generated quantity. As with
predictive likelihood, working in expectation is more statistically efficient than sampling.

5.6. Zero-inflated and hurdle models
Zero-inflated and hurdle models both provide mixtures of a Poisson and Bernoulli probability mass function to allow more flexibility in modeling the probability of a zero outcome. Zero-inflated models, as defined by Lambert (1992), add additional probability mass to the outcome of zero. Hurdle models, on the other hand, are formulated as pure mixtures of zero and non-zero outcomes.

Zero inflation and hurdle models can be formulated for discrete distributions other than the Poisson. Zero inflation does not work for continuous distributions in Stan because of issues with derivatives; in particular, there is no way to add a point mass to a continuous distribution, such as zero-inflating a normal as a regression coefficient prior.

Zero inflation
Consider the following example for zero-inflated Poisson distributions. It uses a parameter \( \theta \) here there is a probability \( \theta \) of drawing a zero, and a probability \( 1 - \theta \) of drawing from \( \text{Poisson}(\lambda) \) (now \( \theta \) is being used for mixing proportions because \( \lambda \) is the traditional notation for a Poisson mean parameter). The probability function is thus

\[
p(y_n \mid \theta, \lambda) = \begin{cases} 
\theta + (1 - \theta) \times \text{Poisson}(0 \mid \lambda) & \text{if } y_n = 0, \text{ and} \\
(1 - \theta) \times \text{Poisson}(y_n \mid \lambda) & \text{if } y_n > 0.
\end{cases}
\]

The log probability function can be implemented directly in Stan as follows.

```stan
data {
  int<lower=0> N;
  int<lower=0> y[N];
}
parameters {
  real<lower=0, upper=1> theta;
  real<lower=0> lambda;
}
model {
  for (n in 1:N) {
    if (y[n] == 0)
      target += log_sum_exp(bernoulli_lpmf(1 | theta), bernoulli_lpmf(0 | theta))
    else
      target += bernoulli_lpmf(y[n] | theta) * poisson_lpmf(y[n] | lambda);
  }
}
```
+ poisson_lpmf(y[n] | lambda));

else
    target += bernoulli_lpmf(0 | theta)
    + poisson_lpmf(y[n] | lambda);
}
}

The `log_sum_exp(lp1,lp2)` function adds the log probabilities on the linear scale; it is defined to be equal to \( \log(\exp(lp1) + \exp(lp2)) \), but is more arithmetically stable and faster.

**Optimizing the zero-inflated Poisson model**

The code given above to compute the zero-inflated Poisson redundantly calculates all of the Bernoulli terms and also `poisson_lpmf(0 | lambda)` every time the first condition body executes. The use of the redundant terms is conditioned on \( y \), which is known when the data are read in. This allows the transformed data block to be used to compute some more convenient terms for expressing the log density each iteration.

The number of zero cases is computed and handled separately. Then the nonzero cases are collected into their own array for vectorization. The number of zeros is required to declare `y_nonzero`, so it must be computed in a function.

```cpp
functions {
    int num_zeros(int[] y) {
        int sum = 0;
        for (n in 1:size(y))
            sum += (y[n] == 0);
        return sum;
    }
}

transformed data {
    int<lower = 0> N_zero = num_zeros(y);
    int<lower = 1> y_nonzero[N - N_zero];
    int N_nonzero = 0;
    for (n in 1:N) {
        if (y[n] == 0) continue;
        N_nonzero += 1;
        y_nonzero[N_nonzero] = y[n];
    }
    ...
model {
  ...
  target += N_zero * log_sum_exp(bernoulli_lpmf(1 | theta),
                         bernoulli_lpmf(0 | theta)
                         + poisson_lpmf(0 | lambda));
  target += N_nonzero * bernoulli_lpmf(0 | theta);
  target += poisson_lpmf(y_nonzero | lambda);
  ...
}

The boundary conditions of all zeros and no zero outcomes is handled appropriately;
in the vectorized case, if y_nonzero is empty, N_nonzero will be zero, and the last
two target increment terms will add zeros.

**Hurdle models**

The hurdle model is similar to the zero-inflated model, but more flexible in that the
zero outcomes can be deflated as well as inflated. The probability mass function for
the hurdle likelihood is defined by

\[
p(y | \theta, \lambda) = \begin{cases} 
\theta & \text{if } y = 0, \text{ and} \\
(1 - \theta) \frac{\text{Poisson}(y | \lambda)}{1 - \text{PoissonCDF}(0 | \lambda)} & \text{if } y > 0,
\end{cases}
\]

where \text{PoissonCDF} is the cumulative distribution function for the Poisson distribution.
The hurdle model is even more straightforward to program in Stan, as it does not
require an explicit mixture.

```stancode
if (y[n] == 0)
  1 ~ bernoulli(theta);
else {
  0 ~ bernoulli(theta);
  y[n] ~ poisson(lambda) T[1, ];
}
```

The Bernoulli statements are just shorthand for adding \( \log \theta \) and \( \log(1 - \theta) \) to the log
density. The T[1,] after the Poisson indicates that it is truncated below at 1; see the
truncation section for more about truncation and the Poisson regression section for
the specifics of the Poisson CDF. The net effect is equivalent to the direct definition of the log likelihood.

```plaintext
if (y[n] == 0)
    target += log(theta);
else
    target += log1m(theta) + poisson_lpmf(y[n] | lambda)
              - poisson_lccdf(0 | lambda));
```

Julian King pointed out that because

\[
\log(1 - \text{PoissonCDF}(0 | \lambda)) = \log(1 - \text{Poisson}(0 | \lambda)) = \log(1 - \exp(-\lambda))
\]

the CCDF in the else clause can be replaced with a simpler expression.

```plaintext
target += log1m(theta) + poisson_lpmf(y[n] | lambda)
              - log1m_exp(-lambda));
```

The resulting code is about 15% faster than the code with the CCDF.

This is an example where collecting counts ahead of time can also greatly speed up the execution speed without changing the density. For data size \(N = 200\) and parameters \(\theta = 0.3\) and \(\lambda = 8\), the speedup is a factor of 10; it will be lower for smaller \(N\) and greater for larger \(N\); it will also be greater for larger \(\theta\).

To achieve this speedup, it helps to have a function to count the number of non-zero entries in an array of integers,

```plaintext
functions {
    int num_zero(int[] y) {
        int nz = 0;
        for (n in 1:size(y))
            if (y[n] == 0)
                nz += 1;
        return nz;
    }
}
```

Then a transformed data block can be used to store the sufficient statistics,

```plaintext
transformed data {
    int<lower=0, upper=N> NO = num_zero(y);
    int<lower=0, upper=N> Ngt0 = N - NO;
```
int<lower=1> y_nz[N - num_zero(y)];
{
    int pos = 1;
    for (n in 1:N) {
        if (y[n] != 0) {
            y_nz[pos] = y[n];
            pos += 1;
        }
    }
}

The model block is then reduced to three statements.

model {
    N0 ~ binomial(N, theta);
    y_nz ~ poisson(lambda);
    target += -Ngt0 * log1m_exp(-lambda);
}

The first statement accounts for the Bernoulli contribution to both the zero and non-zero counts. The second line is the Poisson contribution from the non-zero counts, which is now vectorized. Finally, the normalization for the truncation is a single line, so that the expression for the log CCDF at 0 isn’t repeated. Also note that the negation is applied to the constant $Ngt0$; whenever possible, leave subexpressions constant because then gradients need not be propagated until a non-constant term is encountered.

5.7. Priors and effective data size in mixture models
Suppose we have a two-component mixture model with mixing rate $\lambda \in (0, 1)$. Because the likelihood for the mixture components is proportionally weighted by the mixture weights, the effective data size used to estimate each of the mixture components will also be weighted as a fraction of the overall data size. Thus although there are $N$ observations, the mixture components will be estimated with effective data sizes of $\theta N$ and $(1 - \theta) N$ for the two components for some $\theta \in (0, 1)$. The effective weighting size is determined by posterior responsibility, not simply by the mixing rate $\lambda$.

Comparison to model averaging
In contrast to mixture models, which create mixtures at the observation level, model averaging creates mixtures over the posteriors of models separately fit with the entire data set. In this situation, the priors work as expected when fitting the models.
independently, with the posteriors being based on the complete observed data $y$.

If different models are expected to account for different observations, we recommend building mixture models directly. If the models being mixed are similar, often a single expanded model will capture the features of both and may be used on its own for inferential purposes (estimation, decision making, prediction, etc.). For example, rather than fitting an intercept-only regression and a slope-only regression and averaging their predictions, even as a mixture model, we would recommend building a single regression with both a slope and an intercept. Model complexity, such as having more predictors than data points, can be tamed using appropriately regularizing priors. If computation becomes a bottleneck, the only recourse can be model averaging, which can be calculated after fitting each model independently (see Hoeting et al. (1999) and Andrew Gelman et al. (2013) for theoretical and computational details).
6. Measurement Error and Meta-Analysis

Most quantities used in statistical models arise from measurements. Most of these measurements are taken with some error. When the measurement error is small relative to the quantity being measured, its effect on a model is usually small. When measurement error is large relative to the quantity being measured, or when precise relations can be estimated being measured quantities, it is useful to introduce an explicit model of measurement error. One kind of measurement error is rounding.

Meta-analysis plays out statistically much like measurement error models, where the inferences drawn from multiple data sets are combined to do inference over all of them. Inferences for each data set are treated as providing a kind of measurement error with respect to true parameter values.

6.1. Bayesian measurement error model

A Bayesian approach to measurement error can be formulated directly by treating the true quantities being measured as missing data (Clayton 1992; Richardson and Gilks 1993). This requires a model of how the measurements are derived from the true values.

Regression with measurement error

Before considering regression with measurement error, first consider a linear regression model where the observed data for \( N \) cases includes a predictor \( x_n \) and outcome \( y_n \). In Stan, a linear regression for \( y \) based on \( x \) with a slope and intercept is modeled as follows.

```stan
data {
  int<lower=0> N; // number of cases
  vector[N] x; // predictor (covariate)
  vector[N] y; // outcome (variate)
}
parameters {
  real alpha; // intercept
  real beta; // slope
  real<lower=0> sigma; // outcome noise
}
model {
  y ~ normal(alpha + beta * x, sigma);
}
```
Now suppose that the true values of the predictors $x_n$ are not known, but for each $n$, a measurement $x_{n}^{\text{meas}}$ of $x_n$ is available. If the error in measurement can be modeled, the measured value $x_{n}^{\text{meas}}$ can be modeled in terms of the true value $x_n$ plus measurement noise. The true value $x_n$ is treated as missing data and estimated along with other quantities in the model. A simple approach is to assume the measurement error is normal with known deviation $\tau$. This leads to the following regression model with constant measurement error.

```plaintext
data { ... 
  real x_meas[N]; // measurement of x 
  real<lower=0> tau; // measurement noise 
}
parameters { 
  real x[N]; // unknown true value 
  real mu_x; // prior location 
  real sigma_x; // prior scale 
  ... 
}
model { 
  x ~ normal(mu_x, sigma_x); // prior 
  x_meas ~ normal(x, tau); // measurement model 
  y ~ normal(alpha + beta * x, sigma); 
  ... 
}
```

The regression coefficients alpha and beta and regression noise scale sigma are the same as before, but now $x$ is declared as a parameter rather than as data. The data are now $x_{n}^{\text{meas}}$, which is a measurement of the true $x$ value with noise scale $\tau$. The model then specifies that the measurement error for $x_{\text{meas}}[n]$ given true value $x[n]$ is normal with deviation $\tau$. Furthermore, the true values $x$ are given a hierarchical prior here.

In cases where the measurement errors are not normal, richer measurement error models may be specified. The prior on the true values may also be enriched. For instance, Clayton (1992) introduces an exposure model for the unknown (but noisily
measured) risk factors $x$ in terms of known (without measurement error) risk factors $c$. A simple model would regress $x_n$ on the covariates $c_n$ with noise term $v$,

$$x_n \sim \text{normal}(\gamma^T c, v).$$

This can be coded in Stan just like any other regression. And, of course, other exposure models can be provided.

**Rounding**

A common form of measurement error arises from rounding measurements. Rounding may be done in many ways, such as rounding weights to the nearest milligram, or to the nearest pound; rounding may even be done by rounding down to the nearest integer.

Exercise 3.5(b) by Andrew Gelman et al. (2013) provides an example.

3.5. Suppose we weigh an object five times and measure weights, rounded to the nearest pound, of 10, 10, 12, 11, 9. Assume the unrounded measurements are normally distributed with a noninformative prior distribution on $\mu$ and $\sigma^2$.

(b) Give the correct posterior distribution for $(\mu, \sigma^2)$, treating the measurements as rounded.

Letting $z_n$ be the unrounded measurement for $y_n$, the problem as stated assumes the likelihood

$$z_n \sim \text{normal}(\mu, \sigma).$$

The rounding process entails that $z_n \in (y_n - 0.5, y_n + 0.5)$. The probability mass function for the discrete observation $y$ is then given by marginalizing out the unrounded measurement, producing the likelihood

$$p(y_n \mid \mu, \sigma) = \int_{y_n - 0.5}^{y_n + 0.5} \text{normal}(z_n \mid \mu, \sigma) \, dz_n$$

$$= \Phi \left( \frac{y_n + 0.5 - \mu}{\sigma} \right) - \Phi \left( \frac{y_n - 0.5 - \mu}{\sigma} \right).$$

Gelman’s answer for this problem took the noninformative prior to be uniform in the variance $\sigma^2$ on the log scale, which yields (due to the Jacobian adjustment), the prior density

$$p(\mu, \sigma^2) \propto \frac{1}{\sigma^2}. $$
The posterior after observing \( y = (10, 10, 12, 11, 9) \) can be calculated by Bayes’s rule as

\[
p(\mu, \sigma^2 | y) \propto p(\mu, \sigma^2) \, p(y | \mu, \sigma^2)
\]

\[
\propto \frac{1}{\sigma^2} \prod_{n=1}^{5} \left( \Phi \left( \frac{y_n + 0.5 - \mu}{\sigma} \right) - \Phi \left( \frac{y_n - 0.5 - \mu}{\sigma} \right) \right).
\]

The Stan code simply follows the mathematical definition, providing an example of the direct definition of a probability function up to a proportion.

```stan
data {
  int<lower=0> N;
  vector[N] y;
}
parameters {
  real mu;
  real<lower=0> sigma_sq;
}
transformed parameters {
  real<lower=0> sigma;
  sigma = sqrt(sigma_sq);
}
model {
  target += -2 * log(sigma);
  for (n in 1:N)
    target += log(Phi((y[n] + 0.5 - mu) / sigma)
      - Phi((y[n] - 0.5 - mu) / sigma));
}
```

Alternatively, the model may be defined with latent parameters for the unrounded measurements \( z_n \). The Stan code in this case uses the likelihood for \( z_n \) directly while respecting the constraint \( z_n \in (y_n - 0.5, y_n + 0.5) \). Because Stan does not allow varying upper- and lower-bound constraints on the elements of a vector (or array), the parameters are declared to be the rounding error \( y - z \), and then \( z \) is defined as a transformed parameter.

```stan
data {
  int<lower=0> N;
  vector[N] y;
}
```
parameters {
  real mu;
  real<lower=0> sigma_sq;
  vector<lower=-0.5, upper=0.5>[N] y_err;
}
transformed parameters {
  real<lower=0> sigma;
  vector[N] z;
  sigma = sqrt(sigma_sq);
  z = y + y_err;
}
model {
  target += -2 * log(sigma);
  z ~ normal(mu, sigma);
}

This explicit model for the unrounded measurements $z$ produces the same posterior for $\mu$ and $\sigma$ as the previous model that marginalizes $z$ out. Both approaches mix well, but the latent parameter version is about twice as efficient in terms of effective samples per iteration, as well as providing a posterior for the unrounded parameters.

6.2. Meta-analysis
Meta-analysis aims to pool the data from several studies, such as the application of a tutoring program in several schools or treatment using a drug in several clinical trials.

The Bayesian framework is particularly convenient for meta-analysis, because each previous study can be treated as providing a noisy measurement of some underlying quantity of interest. The model then follows directly from two components, a prior on the underlying quantities of interest and a measurement-error style model for each of the studies being analyzed.

Treatment effects in controlled studies
Suppose the data in question arise from a total of $M$ studies providing paired binomial data for a treatment and control group. For instance, the data might be post-surgical pain reduction under a treatment of ibuprofen (Warn, Thompson, and Spiegelhalter 2002) or mortality after myocardial infarction under a treatment of beta blockers (Andrew Gelman et al. 2013, sec. 5.6).
Data

The clinical data consists of $J$ trials, each with $n^t$ treatment cases, $n^c$ control cases, $r^t$ successful outcomes among those treated and $r^c$ successful outcomes among those in the control group. This data can be declared in Stan as follows.\(^1\)

```stan
data {
  int<lower=0> J;
  int<lower=0> n_t[J];   // num cases, treatment
  int<lower=0> r_t[J];   // num successes, treatment
  int<lower=0> n_c[J];   // num cases, control
  int<lower=0> r_c[J];   // num successes, control
}
```

Converting to log odds and standard error

Although the clinical trial data are binomial in its raw format, it may be transformed to an unbounded scale by considering the log odds ratio

$$y_j = \log \left( \frac{r^t_j / (n^t_j - r^t_j)}{r^c_j / (n^c_j - r^c_j)} \right)$$

$$= \log \left( \frac{r^t_j}{n^t_j - r^t_j} \right) - \log \left( \frac{r^c_j}{n^c_j - r^c_j} \right)$$

and corresponding standard errors

$$\sigma_j = \sqrt{\frac{1}{r^T_i} + \frac{1}{n^T_i - r^T_i} + \frac{1}{r^C_i} + \frac{1}{n^C_i - r^C_i}}.$$

The log odds and standard errors can be defined in a transformed parameter block, though care must be taken not to use integer division.\(^2\)

```stan
transformed data {
  real y[J];
  real<lower=0> sigma[J];
  for (j in 1:J)
```

\(^1\)Stan’s integer constraints are not powerful enough to express the constraint that $r_t[j] \leq n_t[j]$, but this constraint could be checked in the transformed data block.

\(^2\)When dividing two integers, the result type is an integer and rounding will ensue if the result is not exact. See the discussion of primitive arithmetic types in the reference manual for more information.
\[
y[j] = \log(r_t[j]) - \log(n_t[j] - r_t[j]) - (\log(r_c[j]) - \log(n_c[j] - r_c[j])
\]

\[
\text{for } (j \in 1:J)
\]
\[
sigma[j] = \sqrt{\frac{1}{r_t[j]} + \frac{1}{n_t[j] - r_t[j]} + \frac{1}{r_c[j]} + \frac{1}{n_c[j] - r_c[j]}};
\]

This definition will be problematic if any of the success counts is zero or equal to the number of trials. If that arises, a direct binomial model will be required or other transforms must be used than the unregularized sample log odds.

**Non-hierarchical model**

With the transformed data in hand, two standard forms of meta-analysis can be applied. The first is a so-called “fixed effects” model, which assumes a single parameter for the global odds ratio. This model is coded in Stan as follows.

parameters {
  real theta; // global treatment effect, log odds
}

model {
  y ~ normal(theta, sigma);
}

The sampling statement for \( y \) is vectorized; it has the same effect as the following.

\[
\text{for } (j \in 1:J)
\]
\[
y[j] \sim normal(\theta, \sigma[j]);
\]

It is common to include a prior for \( \theta \) in this model, but it is not strictly necessary for the model to be proper because \( y \) is fixed and \( \text{normal}(y | \mu, \sigma) = \text{normal}(\mu | y, \sigma) \).

**Hierarchical model**

To model so-called “random effects,” where the treatment effect may vary by clinical trial, a hierarchical model can be used. The parameters include per-trial treatment effects and the hierarchical prior parameters, which will be estimated along with other unknown quantities.

parameters {
  real theta[J]; // per-trial treatment effect
  real mu; // mean treatment effect
  real<lower=0> tau; // deviation of treatment effects
}
model {
  y ~ normal(theta, sigma);
  theta ~ normal(mu, tau);
  mu ~ normal(0, 10);
  tau ~ cauchy(0, 5);
}

Although the vectorized sampling statement for $y$ appears unchanged, the parameter theta is now a vector. The sampling statement for theta is also vectorized, with the hyperparameters mu and tau themselves being given wide priors compared to the scale of the data.

Donald B. Rubin (1981) provided a hierarchical Bayesian meta-analysis of the treatment effect of Scholastic Aptitude Test (SAT) coaching in eight schools based on the sample treatment effect and standard error in each school.

Extensions and alternatives

Smith, Spiegelhalter, and Thomas (1995) and Andrew Gelman et al. (2013, sec. 19.4) provide meta-analyses based directly on binomial data. Warn, Thompson, and Spiegelhalter (2002) consider the modeling implications of using alternatives to the log-odds ratio in transforming the binomial data.

If trial-specific predictors are available, these can be included directly in a regression model for the per-trial treatment effects $\theta_j$. 
7. Latent Discrete Parameters

Stan does not support sampling discrete parameters. So it is not possible to directly translate BUGS or JAGS models with discrete parameters (i.e., discrete stochastic nodes). Nevertheless, it is possible to code many models that involve bounded discrete parameters by marginalizing out the discrete parameters.¹

This chapter shows how to code several widely-used models involving latent discrete parameters. The next chapter, the clustering chapter, on clustering models, considers further models involving latent discrete parameters.

7.1. The benefits of marginalization

Although it requires some algebra on the joint probability function, a pleasant byproduct of the required calculations is the posterior expectation of the marginalized variable, which is often the quantity of interest for a model. This allows far greater exploration of the tails of the distribution as well as more efficient sampling on an iteration-by-iteration basis because the expectation at all possible values is being used rather than itself being estimated through sampling a discrete parameter.

Standard optimization algorithms, including expectation maximization (EM), are often provided in applied statistics papers to describe maximum likelihood estimation algorithms. Such derivations provide exactly the marginalization needed for coding the model in Stan.

7.2. Change point models

The first example is a model of coal mining disasters in the U.K. for the years 1851–1962.²

Model with latent discrete parameter

Fonnesbeck et al. (2013, sec. 3.1) provides a Poisson model of disaster \( D_t \) in year \( t \) with two rate parameters, an early rate \( e \) and late rate \( l \), that change at a given

---

¹The computations are similar to those involved in expectation maximization (EM) algorithms (Dempster, Laird, and Rubin 1977).
²The source of the data is (Jarrett 1979), which itself is a note correcting an earlier data collection.
point in time \( s \). The full model expressed using a latent discrete parameter \( s \) is

\[
\begin{align*}
    e & \sim \text{exponential}(r_e) \\
    l & \sim \text{exponential}(r_l) \\
    s & \sim \text{uniform}(1, T) \\
    D_t & \sim \text{Poisson}(t < s ? e : l)
\end{align*}
\]

The last line uses the conditional operator (also known as the ternary operator), which is borrowed from C and related languages. The conditional operator has the same behavior as its counterpart in C++.\(^3\)

It uses a compact notation involving separating its three arguments by a question mark (?) and a colon (:) as the conditional operator is defined by

\[
c ? x_1 : x_2 = \begin{cases} 
    x_1 & \text{if } c \text{ is true (i.e., non-zero), and} \\
    x_2 & \text{if } c \text{ is false (i.e., zero).}
\end{cases}
\]

Marginalizing out the discrete parameter

To code this model in Stan, the discrete parameter \( s \) must be marginalized out to produce a model defining the log of the probability function \( p(e, l, D_t) \). The full joint probability factors as

\[
p(e, l, D) = p(e) p(l) p(s) p(D | s, e, l) = \text{exponential}(e | r_e) \text{exponential}(l | r_l) \text{uniform}(s | 1, T)
\]

\[
\prod_{t=1}^{T} \text{Poisson}(D_t | t < s ? e : l),
\]

To marginalize, an alternative factorization into prior and likelihood is used,

\[
p(e, l, D) = p(e, l) p(D | e, l),
\]

\(^3\)The R counterpart, ifelse, is slightly different in that it is typically used in a vectorized situation. The conditional operator is not (yet) vectorized in Stan.
where the likelihood is defined by marginalizing $s$ as

$$p(D \mid e, l) = \sum_{s=1}^{T} p(s, D \mid e, l)$$

$$= \sum_{s=1}^{T} p(s) p(D \mid s, e, l)$$

$$= \sum_{s=1}^{T} \text{uniform}(s \mid 1, T) \prod_{t=1}^{T} \text{Poisson}(D_t \mid t < s ? e : l).$$

Stan operates on the log scale and thus requires the log likelihood,

$$\log p(D \mid e, l) = \log \text{sum}_{s=1}^{T} \left( \log \text{uniform}(s \mid 1, T) \right.$$  

$$+ \sum_{t=1}^{T} \log \text{Poisson}(D_t \mid t < s ? e : l) \left. \right),$$

where the log sum of exponents function is defined by

$$\log \text{sum}_{n=1}^{N} \alpha_n = \log \sum_{n=1}^{N} \exp(\alpha_n).$$

The log sum of exponents function allows the model to be coded directly in Stan using the built-in function \text{log\_sum\_exp}, which provides both arithmetic stability and efficiency for mixture model calculations.

**Coding the model in Stan**

The Stan program for the change point model is shown in the figure below. The transformed parameter `lp[s]` stores the quantity $\log p(s, D \mid e, l)$.

```stan
data {
  real<lower=0> r_e;
  real<lower=0> r_l;
  int<lower=1> T;
  int<lower=0> D[T];
}
transformed data {
```
A change point model in which disaster rates $D[t]$ have one rate, $e$, before the change point and a different rate, $l$, after the change point. The change point itself, $s$, is marginalized out as described in the text.

Although the change-point model is coded directly, the doubly nested loop used for $s$ and $t$ is quadratic in $T$. Luke Wiklendt pointed out that a linear alternative can be achieved by the use of dynamic programming similar to the forward-backward algorithm for Hidden Markov models; he submitted a slight variant of the following code to replace the transformed parameters block of the above Stan program.
\[ \text{lp} = \text{rep_vector}(\log_{\text{unif}} + \text{lp}_1[T + 1], T) \]
\[ + \text{head}(\text{lp}_e, T) - \text{head}(\text{lp}_1, T); \]
\}
\}

As should be obvious from looking at it, it has linear complexity in \( T \) rather than quadratic. The result for the mining-disaster data is about 20 times faster; the improvement will be greater for larger \( T \).

The key to understanding Wiklendt’s dynamic programming version is to see that \( \text{head}(\text{lp}_e) \) holds the forward values, whereas \( \text{lp}_1[T + 1] - \text{head}(\text{lp}_1, T) \) holds the backward values; the clever use of subtraction allows \( \text{lp}_1 \) to be accumulated naturally in the forward direction.

**Fitting the model with MCMC**

This model is easy to fit using MCMC with NUTS in its default configuration. Convergence is fast and sampling produces roughly one effective sample every two iterations. Because it is a relatively small model (the inner double loop over time is roughly 20,000 steps), it is fast.

The value of \( \text{lp} \) for each iteration for each change point is available because it is declared as a transformed parameter. If the value of \( \text{lp} \) were not of interest, it could be coded as a local variable in the model block and thus avoid the I/O overhead of saving values every iteration.

**Posterior distribution of the discrete change point**

The value of \( \text{lp}[s] \) in a given iteration is given by \( \log p(s, D \mid e, l) \) for the values of the early and late rates, \( e \) and \( l \), in the iteration. In each iteration after convergence, the early and late disaster rates, \( e \) and \( l \), are drawn from the posterior \( p(e, l \mid D) \) by MCMC sampling and the associated \( \text{lp} \) calculated. The value of \( \text{lp} \) may be normalized to calculate \( p(s \mid e, l, D) \) in each iteration, based on on the current values of \( e \) and \( l \). Averaging over iterations provides an unnormalized probability estimate of the change point being \( s \) (see below for the normalizing constant),

\[
p(s \mid D) \propto q(s \mid D)
\]
\[
= \frac{1}{M} \sum_{m=1}^{M} \exp(\text{lp}[m, s])
\]

where \( \text{lp}[m, s] \) represents the value of \( \text{lp} \) in posterior draw \( m \) for change point \( s \). By averaging over draws, \( e \) and \( l \) are themselves marginalized out, and the result has no dependence on a given iteration’s value for \( e \) and \( l \). A final normalization then produces the quantity of interest, the posterior probability of the change point being
s conditioned on the data $D$,

$$p(s \mid D) = \frac{q(s \mid D)}{\sum_{s' = 1}^{T} q(s' \mid D)}.$$  

A plot of the values of $\log p(s \mid D)$ computed using Stan 2.4’s default MCMC implementation is shown in the posterior plot.

Log probability of change point being in year, calculated analytically.

Figure 7.1: Analytical change-point posterior

The frequency of change points generated by sampling the discrete change points. In order their range of estimates be visible, the first plot is on the log scale and the second plot on the linear scale; note the narrower range of years in the second plot resulting from sampling. The posterior mean of $s$ is roughly 1891.
**Discrete sampling**

The generated quantities block may be used to draw discrete parameter values using the built-in pseudo-random number generators. For example, with `lp` defined as above, the following program draws a random value for `s` at every iteration.

```plaintext
generated quantities {
  int<lower=1,upper=T> s;
  s = categorical_logit_rng(lp);
}
```

A posterior histogram of draws for `s` is shown on the second change point posterior figure above.

Compared to working in terms of expectations, discrete sampling is highly inefficient, especially for tails of distributions, so this approach should only be used if draws from a distribution are explicitly required. Otherwise, expectations should be computed in the generated quantities block based on the posterior distribution for `s` given by
softmax(lp).

**Posterior covariance**
The discrete sample generated for \( s \) can be used to calculate covariance with other parameters. Although the sampling approach is straightforward, it is more statistically efficient (in the sense of requiring far fewer iterations for the same degree of accuracy) to calculate these covariances in expectation using \( lp \).

**Multiple change points**
There is no obstacle in principle to allowing multiple change points. The only issue is that computation increases from linear to quadratic in marginalizing out two change points, cubic for three change points, and so on. There are three parameters, \( e, m, \) and \( l \), and two loops for the change point and then one over time, with log densities being stored in a matrix.

```plaintext
matrix[T, T] lp;
lp = rep_matrix(log_unif, T);
for (s1 in 1:T)
  for (s2 in 1:T)
    for (t in 1:T)
      lp[s1,s2] = lp[s1,s2] + poisson_lpmf(D[t] | t < s1 ? e : (t < s2 ? m : l));
```

The matrix can then be converted back to a vector using `to_vector` before being passed to `log_sum_exp`.

**7.3. Mark-recapture models**
A widely applied field method in ecology is to capture (or sight) animals, mark them (e.g., by tagging), then release them. This process is then repeated one or more times, and is often done for populations on an ongoing basis. The resulting data may be used to estimate population size.

The first subsection describes a simple mark-recapture model that does not involve any latent discrete parameters. The following subsections describes the Cormack-Jolly-Seber model, which involves latent discrete parameters for animal death.

**Simple mark-recapture model**
In the simplest case, a one-stage mark-recapture study produces the following data

- \( M \) : number of animals marked in first capture,
- \( C \) : number animals in second capture, and
- \( R \) : number of marked animals in second capture.

The estimand of interest is
• \( N \): number of animals in the population.

Despite the notation, the model will take \( N \) to be a continuous parameter; just because the population must be finite doesn’t mean the parameter representing it must be. The parameter will be used to produce a real-valued estimate of the population size.

The Lincoln-Petersen (Lincoln 1930; Petersen 1896) method for estimating population size is

\[
\hat{N} = \frac{MC}{R}.
\]

This population estimate would arise from a probabilistic model in which the number of recaptured animals is distributed binomially,

\[
R \sim \text{binomial}(C, M/N)
\]

given the total number of animals captured in the second round (\( C \)) with a recapture probability of \( M/N \), the fraction of the total population \( N \) marked in the first round.

data {
  int<lower=0> M;
  int<lower=0> C;
  int<lower=0,upper=min(M,C)> R;
}

parameters {
  real<lower=(C - R + M)> N;
}

model {
  R ~ binomial(C, M / N);
}

A probabilistic formulation of the Lincoln-Petersen estimator for population size based on data from a one-step mark-recapture study. The lower bound on \( N \) is necessary to efficiently eliminate impossible values.

The probabilistic variant of the Lincoln-Petersen estimator can be directly coded in Stan as shown in the Lincon-Petersen model figure. The Lincoln-Petersen estimate is the maximum likelihood estimate (MLE) for this model.

To ensure the MLE is the Lincoln-Petersen estimate, an improper uniform prior for \( N \) is used; this could (and should) be replaced with a more informative prior if possible, based on knowledge of the population under study.
The one tricky part of the model is the lower bound $C - R + M$ placed on the population size $N$. Values below this bound are impossible because it is otherwise not possible to draw $R$ samples out of the $C$ animals recaptured. Implementing this lower bound is necessary to ensure sampling and optimization can be carried out in an unconstrained manner with unbounded support for parameters on the transformed (unconstrained) space. The lower bound in the declaration for $C$ implies a variable transform $f : (C - R + M, \infty) \to (-\infty, +\infty)$ defined by $f(N) = \log(N - (C - R + M))$; the reference manual contains full details of all constrained parameter transforms.

**Cormack-Jolly-Seber with discrete parameter**

The Cormack-Jolly-Seber (CJS) model (Cormack 1964; Jolly 1965; Seber 1965) is an open-population model in which the population may change over time due to death; the presentation here draws heavily on Schofield (2007).

The basic data are

- $I$: number of individuals,
- $T$: number of capture periods, and
- $y_{i,t}$: Boolean indicating if individual $i$ was captured at time $t$.

Each individual is assumed to have been captured at least once because an individual only contributes information conditionally after they have been captured the first time.

There are two Bernoulli parameters in the model,

- $\phi_t$: probability that animal alive at time $t$ survives until $t + 1$ and
- $p_t$: probability that animal alive at time $t$ is captured at time $t$.

These parameters will both be given uniform priors, but information should be used to tighten these priors in practice.

The CJS model also employs a latent discrete parameter $z_{i,t}$ indicating for each individual $i$ whether it is alive at time $t$, distributed as

$$z_{i,t} \sim \text{Bernoulli}(z_{i,t} \neq 0 : \phi_{t-1}).$$

The conditional prevents the model positing zombies; once an animal is dead, it stays dead. The data distribution is then simple to express conditional on $z$ as

$$y_{i,t} \sim \text{Bernoulli}(z_{i,t} \neq 0 : p_t).$$

The conditional enforces the constraint that dead animals cannot be captured.
Collective Cormack-Jolly-Seber model

This subsection presents an implementation of the model in terms of counts for different history profiles for individuals over three capture times. It assumes exchangeability of the animals in that each is assigned the same capture and survival probabilities.

In order to ease the marginalization of the latent discrete parameter $z_{i,t}$, the Stan models rely on a derived quantity $\chi_t$ for the probability that an individual is never captured again if it is alive at time $t$ (if it is dead, the recapture probability is zero). This quantity is defined recursively by

$$
\chi_t = \begin{cases} 
1 & \text{if } t = T \\
(1 - \phi_t) + \phi_t (1 - p_{t+1}) \chi_{t+1} & \text{if } t < T
\end{cases}
$$

The base case arises because if an animal was captured in the last time period, the probability it is never captured again is 1 because there are no more capture periods. The recursive case defining $\chi_t$ in terms of $\chi_{t+1}$ involves two possibilities: (1) not surviving to the next time period, with probability $(1 - \phi_t)$, or (2) surviving to the next time period with probability $\phi_t$, not being captured in the next time period with probability $(1 - p_{t+1})$, and not being captured again after being alive in period $t + 1$ with probability $\chi_{t+1}$.

With three capture times, there are eight captured/not-captured profiles an individual may have. These may be naturally coded as binary numbers as follows.

<table>
<thead>
<tr>
<th>profile</th>
<th>captures</th>
<th>probability</th>
</tr>
</thead>
<tbody>
<tr>
<td>0</td>
<td>– – –</td>
<td>n/a</td>
</tr>
<tr>
<td>1</td>
<td>– – +</td>
<td>n/a</td>
</tr>
<tr>
<td>2</td>
<td>– + –</td>
<td>$\chi_2$</td>
</tr>
<tr>
<td>3</td>
<td>– + +</td>
<td>$\phi_2 p_3$</td>
</tr>
<tr>
<td>4</td>
<td>+ – –</td>
<td>$\chi_1$</td>
</tr>
<tr>
<td>5</td>
<td>+ – +</td>
<td>$\phi_1 (1 - p_2) \phi_2 p_3$</td>
</tr>
<tr>
<td>6</td>
<td>+ + –</td>
<td>$\phi_1 p_2 \chi_2$</td>
</tr>
<tr>
<td>7</td>
<td>+ + +</td>
<td>$\phi_1 p_2 \phi_2 p_3$</td>
</tr>
</tbody>
</table>
last round, provides no information for the CJS model, because capture/non-capture status is only informative when conditioned on earlier captures. For the remaining cases, the contribution to the likelihood is provided in the final column.

By defining these probabilities in terms of $\chi$ directly, there is no need for a latent binary parameter indicating whether an animal is alive at time $t$ or not. The definition of $\chi$ is typically used to define the likelihood (i.e., marginalize out the latent discrete parameter) for the CJS model (Schofield 2007).

The Stan model defines $\chi$ as a transformed parameter based on parameters $\phi$ and $p$. In the model block, the log probability is incremented for each history based on its count. This second step is similar to collecting Bernoulli observations into a binomial or categorical observations into a multinomial, only it is coded directly in the Stan program using target += rather than being part of a built-in probability function.

The following is the Stan program for the Cormack-Jolly-Seber mark-recapture model that considers counts of individuals with observation histories of being observed or not in three capture periods

data {
  int<lower=0> history[7];
}
parameters {
  real<lower=0,upper=1> phi[2];
  real<lower=0,upper=1> p[3];
}
transformed parameters {
  real<lower=0,upper=1> chi[2];
}
model {
  target += history[3] * (log(phi[2]) + log(p[3]));
  target += history[4] * (log(chi[1]));
  target += history[5] * (log(phi[1]) + log1m(p[2])
      + log(phi[2]) + log(p[3]));
  target += history[6] * (log(phi[1]) + log(p[2])
      + log(chi[2]));
  target += history[7] * (log(phi[1]) + log(p[2])
      + log(phi[2]) + log(p[3]));
generated quantities {
  real<lower=0,upper=1> beta3;
  beta3 = phi[2] * p[3];
}

Identifiability

The parameters $\phi_2$ and $p_3$, the probability of death at time 2 and probability of capture at time 3 are not identifiable, because both may be used to account for lack of capture at time 3. Their product, $\beta_3 = \phi_2 p_3$, is identified. The Stan model defines $\beta_3$ as a generated quantity. Unidentified parameters pose a problem for Stan’s samplers’ adaptation. Although the problem posed for adaptation is mild here because the parameters are bounded and thus have proper uniform priors, it would be better to formulate an identified parameterization. One way to do this would be to formulate a hierarchical model for the $p$ and $\phi$ parameters.

Individual Cormack-Jolly-Seber model

This section presents a version of the Cormack-Jolly-Seber (CJS) model cast at the individual level rather than collectively as in the previous subsection. It also extends the model to allow an arbitrary number of time periods. The data will consist of the number $T$ of capture events, the number $I$ of individuals, and a boolean flag $y_{i,t}$ indicating if individual $i$ was observed at time $t$. In Stan,

data {
  int<lower=2> T;
  int<lower=0> I;
  int<lower=0,upper=1> y[I, T];
}

The advantages to the individual-level model is that it becomes possible to add individual “random effects” that affect survival or capture probability, as well as to avoid the combinatorics involved in unfolding $2^T$ observation histories for $T$ capture times.

Utility functions

The individual CJS model is written involves several function definitions. The first two are used in the transformed data block to compute the first and last time period
in which an animal was captured.\footnote{An alternative would be to compute this on the outside and feed it into the Stan model as preprocessed data. Yet another alternative encoding would be a sparse one recording only the capture events along with their time and identifying the individual captured.}

```stan
functions {
  int first_capture(int[] y_i) {
    for (k in 1:size(y_i))
      if (y_i[k])
        return k;
    return 0;
  }

  int last_capture(int[] y_i) {
    for (k_rev in 0:(size(y_i) - 1)) {
      int k;
      k = size(y_i) - k_rev;
      if (y_i[k])
        return k;
    }
    return 0;
  }
}
```

These two functions are used to define the first and last capture time for each individual in the transformed data block.\footnote{Both functions return 0 if the individual represented by the input array was never captured. Individuals with no captures are not relevant for estimating the model because all probability statements are conditional on earlier captures. Typically they would be removed from the data, but the program allows them to be included even though they make no contribution to the log probability function.}

```stan
transformed data {
  int<lower=0,upper=T> first[I];
  int<lower=0,upper=T> last[I];
  vector<lower=0,upper=I>[T] n_captured;
  for (i in 1:I)
    first[i] = first_capture(y[i]);
  for (i in 1:I)
    last[i] = last_capture(y[i]);
  n_captured = rep_vector(0, T);
  for (t in 1:T)
    for (i in 1:I)
      n_captured[t] += (y[i] == t);
}
```
if (y[i, t])
    n_captured[t] = n_captured[t] + 1;
}

The transformed data block also defines n_captured[t], which is the total number of captures at time t. The variable n_captured is defined as a vector instead of an integer array so that it can be used in an elementwise vector operation in the generated quantities block to model the population estimates at each time point.

The parameters and transformed parameters are as before, but now there is a function definition for computing the entire vector chi, the probability that if an individual is alive at t that it will never be captured again.

parameters {
  vector<lower=0,upper=1>[T-1] phi;
  vector<lower=0,upper=1>[T] p;
}
transformed parameters {
  vector<lower=0,upper=1>[T] chi;
  chi = prob_uncaptured(T,p,phi);
}

The definition of prob_uncaptured, from the functions block, is

functions {
  ...
  vector prob_uncaptured(int T, vector p, vector phi) {
    vector[T] chi;
    chi[T] = 1.0;
    for (t in 1:(T - 1)) {
      int t_curr;
      int t_next;
      t_curr = T - t;
      t_next = t_curr + 1;
      chi[t_curr] = (1 - phi[t_curr])
                    + phi[t_curr]
                    * (1 - p[t_next])
                    * chi[t_next];
    }
    return chi;
  }
}
The function definition directly follows the mathematical definition of $\chi_t$, unrolling
the recursion into an iteration and defining the elements of chi from $T$ down to 1.

**The model**

Given the precomputed quantities, the model block directly encodes the CJS model’s
log likelihood function. All parameters are left with their default uniform priors
and the model simply encodes the log probability of the observations $q$ given the
parameters $p$ and $\phi$ as well as the transformed parameter $\chi$ defined in terms of
$p$ and $\phi$.

```r
model {
  for (i in 1:I) {
    if (first[i] > 0) {
      for (t in (first[i]+1):last[i]) {
        1 ~ bernoulli(phi[t-1]);
        y[i, t] ~ bernoulli(p[t]);
      }
      1 ~ bernoulli(chi[last[i]]);
    }
  }
}
```

The outer loop is over individuals, conditional skipping individuals $i$ which are never
captured. The never-captured check depends on the convention of the first-capture
and last-capture functions returning 0 for `first` if an individual is never captured.

The inner loop for individual $i$ first increments the log probability based on the
survival of the individual with probability $\phi[t-1]$. The outcome of 1 is fixed
because the individual must survive between the first and last capture (i.e., no
zombies). The loop starts after the first capture, because all information in the CJS
model is conditional on the first capture.

In the inner loop, the observed capture status $y[i, t]$ for individual $i$ at time $t$ has
a Bernoulli distribution based on the capture probability $p[t]$ at time $t$.

After the inner loop, the probability of an animal never being seen again after being
observed at time `last[i]` is included, because `last[i]` was defined to be the last
time period in which animal $i$ was observed.
Identified parameters

As with the collective model described in the previous subsection, this model does not identify \( \phi[T-1] \) and \( p[T] \), but does identify their product, beta. Thus beta is defined as a generated quantity to monitor convergence and report.

```plaintext
generated quantities {
  real beta;
  ...

  beta = phi[T-1] * p[T];
  ...
}
```

The parameter \( p[1] \) is also not modeled and will just be uniform between 0 and 1. A more finely articulated model might have a hierarchical or time-series component, in which case \( p[1] \) would be an unknown initial condition and both \( \phi[T-1] \) and \( p[T] \) could be identified.

Population size estimates

The generated quantities also calculates an estimate of the population mean at each time \( t \) in the same way as in the simple mark-recapture model as the number of individuals captured at time \( t \) divided by the probability of capture at time \( t \). This is done with the elementwise division operation for vectors (./) in the generated quantities block.

```plaintext
generated quantities {
  ...
  vector<lower=0>[T] pop;
  ...
  pop = n_captured ./ p;
  pop[1] = -1;
}
```

Generalizing to individual effects

All individuals are modeled as having the same capture probability, but this model could be easily generalized to use a logistic regression here based on individual-level inputs to be used as predictors.
7.4. Data coding and diagnostic accuracy models
Although seemingly disparate tasks, the rating/coding/annotation of items with categories and diagnostic testing for disease or other conditions, share several characteristics which allow their statistical properties to be modeled similarly.

Diagnostic accuracy
Suppose you have diagnostic tests for a condition of varying sensitivity and specificity. Sensitivity is the probability a test returns positive when the patient has the condition and specificity is the probability that a test returns negative when the patient does not have the condition. For example, mammograms and puncture biopsy tests both test for the presence of breast cancer. Mammograms have high sensitivity and low specificity, meaning lots of false positives, whereas puncture biopsies are the opposite, with low sensitivity and high specificity, meaning lots of false negatives.

There are several estimands of interest in such studies. An epidemiological study may be interested in the prevalence of a kind of infection, such as malaria, in a population. A test development study might be interested in the diagnostic accuracy of a new test. A health care worker performing tests might be interested in the disease status of a particular patient.

Data coding
Humans are often given the task of coding (equivalently rating or annotating) data. For example, journal or grant reviewers rate submissions, a political study may code campaign commercials as to whether they are attack ads or not, a natural language processing study might annotate Tweets as to whether they are positive or negative in overall sentiment, or a dentist looking at an X-ray classifies a patient as having a cavity or not. In all of these cases, the data coders play the role of the diagnostic tests and all of the same estimands are in play — data coder accuracy and bias, true categories of items being coded, or the prevalence of various categories of items in the data.

Noisy categorical measurement model
In this section, only categorical ratings are considered, and the challenge in the modeling for Stan is to marginalize out the discrete parameters.

A. P. Dawid and Skene (1979) introduce a noisy-measurement model for coding and apply it in the epidemiological setting of coding what doctors say about patient histories; the same model can be used for diagnostic procedures.
Data

The data for the model consists of $J$ raters (diagnostic tests), $I$ items (patients), and $K$ categories (condition statuses) to annotate, with $y_{i,j} \in \{1, \ldots, K\}$ being the rating provided by rater $j$ for item $i$. In a diagnostic test setting for a particular condition, the raters are diagnostic procedures and often $K = 2$, with values signaling the presence or absence of the condition.\textsuperscript{6}

It is relatively straightforward to extend Dawid and Skene’s model to deal with the situation where not every rater rates each item exactly once.

Model parameters

The model is based on three parameters, the first of which is discrete:

- $z_i$: a value in $\{1, \ldots, K\}$ indicating the true category of item $i$,
- $\pi$: a $K$-simplex for the prevalence of the $K$ categories in the population, and
- $\theta_{j,k}$: a $K$-simplex for the response of annotator $j$ to an item of true category $k$.

Noisy measurement model

The true category of an item is assumed to be generated by a simple categorical distribution based on item prevalence,

$$z_i \sim \text{categorical}(\pi).$$

The rating $y_{i,j}$ provided for item $i$ by rater $j$ is modeled as a categorical response of rater $i$ to an item of category $z_i$,\textsuperscript{7}

$$y_{i,j} \sim \text{categorical}(\theta_{j,z_i}).$$

Priors and hierarchical modeling

Dawid and Skene provided maximum likelihood estimates for $\theta$ and $\pi$, which allows them to generate probability estimates for each $z_i$.

To mimic Dawid and Skene’s maximum likelihood model, the parameters $\theta_{j,k}$ and $\pi$ can be given uniform priors over $K$-simplexes. It is straightforward to generalize to Dirichlet priors,

$$\pi \sim \text{Dirichlet}(\alpha).$$

\textsuperscript{6}Diagnostic procedures are often ordinal, as in stages of cancer in oncological diagnosis or the severity of a cavity in dental diagnosis. Dawid and Skene’s model may be used as is or naturally generalized for ordinal ratings using a latent continuous rating and cutpoints as in ordinal logistic regression.

\textsuperscript{7}In the subscript, $z_i$ is written as $z[i]$ to improve legibility.
and

$$\theta_{j,k} \sim \text{Dirichlet}(\beta_k)$$

with fixed hyperparameters $\alpha$ (a vector) and $\beta$ (a matrix or array of vectors). The prior for $\theta_{j,k}$ must be allowed to vary in $k$, so that, for instance, $\beta_{k,k}$ is large enough to allow the prior to favor better-than-chance annotators over random or adversarial ones.

Because there are $J$ coders, it would be natural to extend the model to include a hierarchical prior for $\beta$ and to partially pool the estimates of coder accuracy and bias.

**Marginalizing out the true category**

Because the true category parameter $z$ is discrete, it must be marginalized out of the joint posterior in order to carry out sampling or maximum likelihood estimation in Stan. The joint posterior factors as

$$p(y, \theta, \pi) = p(y | \theta, \pi) p(\pi) p(\theta),$$

where $p(y | \theta, \pi)$ is derived by marginalizing $z$ out of

$$p(z, y | \theta, \pi) = \prod_{i=1}^{I} \left( \text{categorical}(z_i | \pi) \prod_{j=1}^{J} \text{categorical}(y_{i,j} | \theta_{j,z[i]}) \right).$$

This can be done item by item, with

$$p(y | \theta, \pi) = \prod_{i=1}^{I} \sum_{k=1}^{K} \left( \text{categorical}(k | \pi) \prod_{j=1}^{J} \text{categorical}(y_{i,j} | \theta_{j,k}) \right).$$

In the missing data model, only the observed labels would be used in the inner product.

A. P. Dawid and Skene (1979) derive exactly the same equation in their Equation (2.7), required for the E-step in their expectation maximization (EM) algorithm. Stan requires the marginalized probability function on the log scale,

$$\log p(y | \theta, \pi) = \sum_{i=1}^{I} \log \left( \sum_{k=1}^{K} \exp \left( \log \text{categorical}(k | \pi) + \sum_{j=1}^{J} \log \text{categorical}(y_{i,j} | \theta_{j,k}) \right) \right)$$

which can be directly coded using Stan’s built-in `log_sum_exp` function.
Stan implementation
The Stan program for the Dawid and Skene model is provided below (A. P. Dawid and Skene 1979).

data {
    int<lower=2> K;
    int<lower=1> I;
    int<lower=1> J;

    int<lower=1,upper=K> y[I, J];

    vector<lower=0>[K] alpha;
    vector<lower=0>[K] beta[K];
}
parameters {
    simplex[K] pi;
    simplex[K] theta[J, K];
}
transformed parameters {
    vector[K] log_q_z[I];
    for (i in 1:I) {
        log_q_z[i] = log(pi);
        for (j in 1:J)
            for (k in 1:K)
                log_q_z[i, k] = log_q_z[i, k]
                    + log(theta[j, k, y[i, j]]);
    }
}
model {
    pi ~ dirichlet(alpha);
    for (j in 1:J)
        for (k in 1:K)
            theta[j, k] ~ dirichlet(beta[k]);

    for (i in 1:I)
        target += log_sum_exp(log_q_z[i]);
}

The model marginalizes out the discrete parameter $z$, storing the unnormalized conditional probability $\log q(z_i = k|\theta, \pi)$ in $\log_q_z[i, k]$. 
The Stan model converges quickly and mixes well using NUTS starting at diffuse initial points, unlike the equivalent model implemented with Gibbs sampling over the discrete parameter. Reasonable weakly informative priors are $\alpha_k = 3$ and $\beta_{k,k'} = 2.5K$ and $\beta_{k,k'} = 1$ if $k \neq k'$. Taking $\alpha$ and $\beta_k$ to be unit vectors and applying optimization will produce the same answer as the expectation maximization (EM) algorithm of A. P. Dawid and Skene (1979).

**Inference for the true category**

The quantity $\log_q z[i]$ is defined as a transformed parameter. It encodes the (unnormalized) log of $p(z_i \mid \theta, \pi)$. Each iteration provides a value conditioned on that iteration's values for $\theta$ and $\pi$. Applying the softmax function to $\log_q z[i]$ provides a simplex corresponding to the probability mass function of $z_i$ in the posterior. These may be averaged across the iterations to provide the posterior probability distribution over each $z_i$. 
8. Sparse and Ragged Data Structures

Stan does not directly support either sparse or ragged data structures, though both can be accommodated with some programming effort. The sparse matrices chapter introduces a special-purpose sparse matrix times dense vector multiplication, which should be used where applicable; this chapter covers more general data structures.

8.1. Sparse data structures
Coding sparse data structures is as easy as moving from a matrix-like data structure to a database-like data structure. For example, consider the coding of sparse data for the IRT models discussed in the item-response model section. There are $J$ students and $K$ questions, and if every student answers every question, then it is practical to declare the data as a $J \times K$ array of answers.

```data{
  int<lower=1> J;
  int<lower=1> K;
  int<lower=0,upper=1> y[J, K];
  ...
model{
  for (j in 1:J)
    for (k in 1:K)
      y[j, k] ~ bernoulli_logit(delta[k] * (alpha[j] - beta[k]));
  ...
```

On the left is a definition of a sparse matrix $y$ using the NA notation from R (which is not supported by Stan). On the right is a database-like encoding of the same sparse matrix $y$ that can be used directly in Stan. The first two columns, $jj$ and $kk$, denote

<table>
<thead>
<tr>
<th>jj</th>
<th>kk</th>
<th>y</th>
</tr>
</thead>
<tbody>
<tr>
<td>1</td>
<td>1</td>
<td>0</td>
</tr>
<tr>
<td>1</td>
<td>2</td>
<td>1</td>
</tr>
<tr>
<td>1</td>
<td>4</td>
<td>1</td>
</tr>
<tr>
<td>2</td>
<td>1</td>
<td>0</td>
</tr>
<tr>
<td>2</td>
<td>4</td>
<td>1</td>
</tr>
<tr>
<td>3</td>
<td>2</td>
<td>0</td>
</tr>
</tbody>
</table>
the indexes and the final column, $y$, the value. For example, the fifth row of the database-like data structure on the right indicates that $y_{2,4} = 1$.

When not every student is given every question, the dense array coding will no longer work, because Stan does not support undefined values. The sparse data example shows an example with $J = 3$ and $K = 4$, with missing responses shown as NA, as in R. There is no support within Stan for R's NA values, so this data structure cannot be used directly. Instead, it must be converted to a “long form” as in a database, with columns indicating the $j$ and $k$ indexes along with the value. For instance, with $jj$ and $kk$ used for the indexes (following Andrew Gelman and Hill (2007)), the data structure can be coded as in the right-hand example in the example. This says that $y_{1,1} = 0$, $y_{1,2} = 1$, and so on, up to $y_{3,2} = 1$, with all other entries undefined.

Letting $N$ be the number of $y$ that are defined, here $N = 6$, the data and model can be formulated as follows.

```stan
data {
    ...
    int<lower=1> N;
    int<lower=1,upper=J> jj[N];
    int<lower=1,upper=K> kk[N];
    int<lower=0,upper=1> y[N];
    ...
}
model {
    for (n in 1:N)
        y[n] ~ bernoulli_logit(delta[kk[n]]
            * (alpha[jj[n]] - beta[kk[n]]));
    ...
}
```

In the situation where there are no missing values, the two model formulations produce exactly the same log posterior density.

### 8.2. Ragged data structures

Ragged arrays are arrays that are not rectangular, but have different sized entries. This kind of structure crops up when there are different numbers of observations per entry.

A general approach to dealing with ragged structure is to move to a full database-like data structure as discussed in the previous section. A more compact approach is possible with some indexing into a linear array.

For example, consider a data structure for three groups, each of which has a different number of observations.
On the left is the definition of a ragged data structure $y$ with three rows of different sizes ($y_1$ is size 3, $y_2$ size 2, and $y_3$ size 4). On the right is an example of how to code the data in Stan, using a single vector $y$ to hold all the values and a separate array of integers $s$ to hold the group row sizes. In this example, $y_1 = z_{1:3}$, $y_2 = z_{4:5}$, and $y_3 = z_{6:9}$.

Suppose the model is a simple varying intercept model, which, using vectorized notation, would yield a log-likelihood

$$
\sum_{n=1}^{3} \log \text{normal}(y_n \mid \mu_n, \sigma).
$$

There's no direct way to encode this in Stan.

A full database type structure could be used, as in the sparse example, but this is inefficient, wasting space for unnecessary indices and not allowing vector-based density operations. A better way to code this data is as a single list of values, with a separate data structure indicating the sizes of each subarray. This is indicated on the right of the example. This coding uses a single array for the values and a separate array for the sizes of each row.

The model can then be coded up using slicing operations as follows.

data {
  int<lower=0> N;  // # observations
  int<lower=0> K;  // # of groups
  vector[N] y;    // observations
  int s[K];       // group sizes
...}

model {
  int pos;
  pos = 1;
  for (k in 1:K) {
    segment(y, pos, s[k]) ~ normal(mu[k], sigma);
    pos = pos + s[k];
  }
}

This coding allows for efficient vectorization, which is worth the copy cost entailed by the `segment()` vector slicing operation.
9. Clustering Models

Unsupervised methods for organizing data into groups are collectively referred to as clustering. This chapter describes the implementation in Stan of two widely used statistical clustering models, soft $K$-means and latent Dirichlet allocation (LDA). In addition, this chapter includes naive Bayesian classification, which can be viewed as a form of clustering which may be supervised. These models are typically expressed using discrete parameters for cluster assignments. Nevertheless, they can be implemented in Stan like any other mixture model by marginalizing out the discrete parameters (see the mixture modeling chapter).

9.1. Relation to finite mixture models

As mentioned in the clustering section, clustering models and finite mixture models are really just two sides of the same coin. The “soft” $K$-means model described in the next section is a normal mixture model (with varying assumptions about covariance in higher dimensions leading to variants of $K$-means). Latent Dirichlet allocation is a mixed-membership multinomial mixture.

9.2. Soft $K$-means

$K$-means clustering is a method of clustering data represented as $D$-dimensional vectors. Specifically, there will be $N$ items to be clustered, each represented as a vector $y_n \in \mathbb{R}^D$. In the “soft” version of $K$-means, the assignments to clusters will be probabilistic.

Geometric hard $K$-means clustering

$K$-means clustering is typically described geometrically in terms of the following algorithm, which assumes the number of clusters $K$ and data vectors $y$ as input.

1. For each $n$ in $\{1, \ldots, N\}$, randomly assign vector $y_n$ to a cluster in $\{1, \ldots, K\}$;
2. Repeat
   1. For each cluster $k$ in $\{1, \ldots, K\}$, compute the cluster centroid $\mu_k$ by averaging the vectors assigned to that cluster;
   2. For each $n$ in $\{1, \ldots, N\}$, reassign $y_n$ to the cluster $k$ for which the (Euclidean) distance from $y_n$ to $\mu_k$ is smallest;
   3. If no vectors changed cluster, return the cluster assignments.

This algorithm is guaranteed to terminate.
Soft K-means clustering
Soft K-means clustering treats the cluster assignments as probability distributions over the clusters. Because of the connection between Euclidean distance and multivariate normal models with a fixed covariance, soft K-means can be expressed (and coded in Stan) as a multivariate normal mixture model.

In the full generative model, each data point \( n \in \{1, \ldots, N\} \) is assigned a cluster \( z_n \in \{1, \ldots, K\} \) with symmetric uniform probability,

\[
z_n \sim \text{categorical}(1/K),
\]

where \( 1 \) is the unit vector of \( K \) dimensions, so that \( 1/K \) is the symmetric \( K \)-simplex. Thus the model assumes that each data point is drawn from a hard decision about cluster membership. The softness arises only from the uncertainty about which cluster generated a data point.

The data points themselves are generated from a multivariate normal distribution whose parameters are determined by the cluster assignment \( z_n \),

\[
y_n \sim \text{normal}(\mu_{z[n]}, \Sigma_{z[n]})
\]

The sample implementation in this section assumes a fixed unit covariance matrix shared by all clusters \( k \),

\[
\Sigma_k = \text{diag\_matrix}(1),
\]

so that the log multivariate normal can be implemented directly up to a proportion by

\[
\text{normal}\left(y_n | \mu_k, \text{diag\_matrix}(1)\right) \propto \exp\left(-\frac{1}{2} \sum_{d=1}^{D} (\mu_{k,d} - y_{n,d})^2 \right).
\]

The spatial perspective on K-means arises by noting that the inner term is just half the negative Euclidean distance from the cluster mean \( \mu_k \) to the data point \( y_n \).

Stan implementation of soft K-means
Consider the following Stan program for implementing K-means clustering.

data {
    int<lower=0> N; // number of data points
    int<lower=1> D; // number of dimensions
    int<lower=1> K; // number of clusters
    vector[D] y[N]; // observations
}
transformed data {
real<upper=0> neg_log_K;
neg_log_K = -log(K);
}
parameters {
    vector[D] mu[K]; // cluster means
}
transformed parameters {
    real<upper=0> soft_z[N, K]; // log unnormalized clusters
    for (n in 1:N)
        for (k in 1:K)
            soft_z[n, k] = neg_log_K
            - 0.5 * dot_self(mu[k] - y[n]);
}
model {
    // prior
    for (k in 1:K)
        mu[k] ~ std_normal();

    // likelihood
    for (n in 1:N)
        target += log_sum_exp(soft_z[n]);
}

There is an independent standard normal prior on the centroid parameters; this prior could be swapped with other priors, or even a hierarchical model to fit an overall problem scale and location.

The only parameter is $\mu$, where $\mu[k]$ is the centroid for cluster $k$. The transformed parameters $soft_z[n]$ contain the log of the unnormalized cluster assignment probabilities. The vector $soft_z[n]$ can be converted back to a normalized simplex using the softmax function (see the functions reference manual), either externally or within the model's generated quantities block.

**Generalizing soft K-means**

The multivariate normal distribution with unit covariance matrix produces a log probability density proportional to Euclidean distance (i.e., $L_2$ distance). Other distributions relate to other geometries. For instance, replacing the normal distribution with the double exponential (Laplace) distribution produces a clustering model based on $L_1$ distance (i.e., Manhattan or taxicab distance).

Within the multivariate normal version of $K$-means, replacing the unit covariance
matrix with a shared covariance matrix amounts to working with distances defined in a space transformed by the inverse covariance matrix.

Although there is no global spatial analog, it is common to see soft $K$-means specified with a per-cluster covariance matrix. In this situation, a hierarchical prior may be used for the covariance matrices.

9.3. The difficulty of Bayesian inference for clustering

Two problems make it pretty much impossible to perform full Bayesian inference for clustering models, the lack of parameter identifiability and the extreme multimodality of the posteriors. There is additional discussion related to the non-identifiability due to label switching in the label switching section.

Non-identifiability
Cluster assignments are not identified—permuting the cluster mean vectors $\mu$ leads to a model with identical likelihoods. For instance, permuting the first two indexes in $\mu$ and the first two indexes in each $\text{soft}_z[n]$ leads to an identical likelihood (and prior).

The lack of identifiability means that the cluster parameters cannot be compared across multiple Markov chains. In fact, the only parameter in soft $K$-means is not identified, leading to problems in monitoring convergence. Clusters can even fail to be identified within a single chain, with indices swapping if the chain is long enough or the data are not cleanly separated.

Multimodality
The other problem with clustering models is that their posteriors are highly multimodal. One form of multimodality is the non-identifiability leading to index swapping. But even without the index problems the posteriors are highly multimodal.

Bayesian inference fails in cases of high multimodality because there is no way to visit all of the modes in the posterior in appropriate proportions and thus no way to evaluate integrals involved in posterior predictive inference.

In light of these two problems, the advice often given in fitting clustering models is to try many different initializations and select the sample with the highest overall probability. It is also popular to use optimization-based point estimators such as expectation maximization or variational Bayes, which can be much more efficient than sampling-based approaches.
9.4. Naive Bayes classification and clustering

Naive Bayes is a kind of mixture model that can be used for classification or for clustering (or a mix of both), depending on which labels for items are observed.\footnote{For clustering, the non-identifiability problems for all mixture models present a problem, whereas there is no such problem for classification. Despite the difficulties with full Bayesian inference for clustering, researchers continue to use it, often in an exploratory data analysis setting rather than for predictive modeling.}

Multinomial mixture models are referred to as “naive Bayes” because they are often applied to classification problems where the multinomial independence assumptions are clearly false.

Naive Bayes classification and clustering can be applied to any data with multinomial structure. A typical example of this is natural language text classification and clustering, which is used as an example in what follows.

The observed data consists of a sequence of $M$ documents made up of bags of words drawn from a vocabulary of $V$ distinct words. A document $m$ has $N_m$ words, which are indexed as $w_{m,1}, \ldots, w_{m,N_m}$ $\in \{1, \ldots, V\}$. Despite the ordered indexing of words in a document, this order is not part of the model, which is clearly defective for natural human language data. A number of topics (or categories) $K$ is fixed.

The multinomial mixture model generates a single category $z_m \in \{1, \ldots, K\}$ for each document $m \in \{1, \ldots, M\}$ according to a categorical distribution,

$$z_m \sim \text{categorical}(\theta).$$

The $K$-simplex parameter $\theta$ represents the prevalence of each category in the data.

Next, the words in each document are generated conditionally independently of each other and the words in other documents based on the category of the document, with word $n$ of document $m$ being generated as

$$w_{m,n} \sim \text{categorical}(\phi_{z[m]}).$$

The parameter $\phi_{z[m]}$ is a $V$-simplex representing the probability of each word in the vocabulary in documents of category $z_m$.

The parameters $\theta$ and $\phi$ are typically given symmetric Dirichlet priors. The prevalence $\theta$ is sometimes fixed to produce equal probabilities for each category $k \in \{1, \ldots, K\}$.

Coding ragged arrays

The specification for naive Bayes in the previous sections have used a ragged array notation for the words $w$. Because Stan does not support ragged arrays, the models...
are coded using an alternative strategy that provides an index for each word in a
global list of words. The data is organized as follows, with the word arrays laid out
in a column and each assigned to its document in a second column.

The relevant variables for the program are \( N \), the total number of words in all the
documents, the word array \( w \), and the document identity array \( \text{doc} \).

**Estimation with category-labeled training data**

A naive Bayes model for estimating the simplex parameters given training data with
documents of known categories can be coded in Stan as follows

```
data {
  // training data
  int<lower=1> K; // num topics
  int<lower=1> V; // num words
  int<lower=0> M; // num docs
  int<lower=0> N; // total word instances
  int<lower=1,upper=K> z[M]; // topic for doc m
  int<lower=1,upper=V> w[N]; // word n
  int<lower=1,upper=M> doc[N]; // doc ID for word n
  // hyperparameters
  vector<lower=0>[K] alpha; // topic prior
  vector<lower=0>[V] beta; // word prior
}
```
parameters {
  simplex[K] theta;             // topic prevalence
  simplex[V] phi[K];            // word dist for topic k
}
model {
  theta ~ dirichlet(alpha);
  for (k in 1:K)
    phi[k] ~ dirichlet(beta);
  for (m in 1:M)
    z[m] ~ categorical(theta);
  for (n in 1:N)
    w[n] ~ categorical(phi[z[doc[n]]]);
}

The topic identifiers \( z_m \) are declared as data and the latent category assignments are included as part of the likelihood function.

**Estimation without category-labeled training data**

Naive Bayes models can be used in an unsupervised fashion to cluster multinomial-structured data into a fixed number \( K \) of categories. The data declaration includes the same variables as the model in the previous section excluding the topic labels \( z \). Because \( z \) is discrete, it needs to be summed out of the model calculation. This is done for naive Bayes as for other mixture models. The parameters are the same up to the priors, but the likelihood is now computed as the marginal document probability

\[
\log p(w_{m,1}, \ldots, w_{m,N_m} \mid \theta, \phi) = \log \sum_{k=1}^{K} \left( \text{categorical}(k \mid \theta) \times \prod_{n=1}^{N_m} \text{categorical}(w_{m,n} \mid \phi_k) \right)
\]

\[
= \log \sum_{k=1}^{K} \exp \left( \log \text{categorical}(k \mid \theta) + \sum_{n=1}^{N_m} \log \text{categorical}(w_{m,n} \mid \phi_k) \right).
\]

The last step shows how the \( \log\_\text{sum}\_\text{exp} \) function can be used to stabilize the numerical calculation and return a result on the log scale.

model {
  real gamma[M, K];
  theta ~ dirichlet(alpha);
  for (k in 1:K)
    phi[k] ~ dirichlet(beta);
}
for (m in 1:M)
    for (k in 1:K)
        gamma[m, k] = categorical_lpmf(k | theta);
for (n in 1:N)
    for (k in 1:K)
        gamma[doc[n], k] = gamma[doc[n], k] + categorical_lpmf(w[n] | phi[k]);
for (m in 1:M)
    target += log_sum_exp(gamma[m]);
}

The local variable \( \gamma_{m,k} \) represents the value

\[
\gamma_{m,k} = \log \text{categorical}(k \mid \theta) + \sum_{n=1}^{N_m} \log \text{categorical}(w_{m,n} \mid \phi_k).
\]

Given \( \gamma \), the posterior probability that document \( m \) is assigned category \( k \) is

\[
\Pr[z_m = k \mid w, \alpha, \beta] = \exp \left( \gamma_{m,k} - \log \sum_{k=1}^{K} \exp(\gamma_{m,k}) \right).
\]

If the variable gamma were declared and defined in the transformed parameter block, its sampled values would be saved by Stan. The normalized posterior probabilities could also be defined as generated quantities.

**Full Bayesian inference for naive Bayes**

Full Bayesian posterior predictive inference for the naive Bayes model can be implemented in Stan by combining the models for labeled and unlabeled data. The estimands include both the model parameters and the posterior distribution over categories for the unlabeled data. The model is essentially a missing data model assuming the unknown category labels are missing completely at random; see Andrew Gelman et al. (2013) and Andrew Gelman and Hill (2007) for more information on missing data imputation. The model is also an instance of semisupervised learning because the unlabeled data contributes to the parameter estimations.

To specify a Stan model for performing full Bayesian inference, the model for labeled data is combined with the model for unlabeled data. A second document collection is declared as data, but without the category labels, leading to new variables \( M' \), \( N' \), \( w' \), and \( doc' \). The number of categories and number of words, as well as the hyperparameters are shared and only declared once. Similarly, there is only one set
of parameters. Then the model contains a single set of statements for the prior, a set of statements for the labeled data, and a set of statements for the unlabeled data.

**Prediction without model updates**
An alternative to full Bayesian inference involves estimating a model using labeled data, then applying it to unlabeled data without updating the parameter estimates based on the unlabeled data. This behavior can be implemented by moving the definition of $\gamma$ for the unlabeled documents to the generated quantities block. Because the variables no longer contribute to the log probability, they no longer jointly contribute to the estimation of the model parameters.

### 9.5. Latent Dirichlet allocation
Latent Dirichlet allocation (LDA) is a mixed-membership multinomial clustering model (Blei, Ng, and Jordan 2003) that generalizes naive Bayes. Using the topic and document terminology common in discussions of LDA, each document is modeled as having a mixture of topics, with each word drawn from a topic based on the mixing proportions.

#### The LDA Model
The basic model assumes each document is generated independently based on fixed hyperparameters. For document $m$, the first step is to draw a topic distribution simplex $\theta_m$ over the $K$ topics,

$$\theta_m \sim \text{Dirichlet}(\alpha).$$

The prior hyperparameter $\alpha$ is fixed to a $K$-vector of positive values. Each word in the document is generated independently conditional on the distribution $\theta_m$. First, a topic $z_{m,n} \in \{1, \ldots, K\}$ is drawn for the word based on the document-specific topic-distribution,

$$z_{m,n} \sim \text{categorical}(\theta_m).$$

Finally, the word $w_{m,n}$ is drawn according to the word distribution for topic $z_{m,n}$,

$$w_{m,n} \sim \text{categorical}(\phi_{z_{m,n}}).$$

The distributions $\phi_k$ over words for topic $k$ are also given a Dirichlet prior,

$$\phi_k \sim \text{Dirichlet}(\beta)$$

where $\beta$ is a fixed $V$-vector of positive values.
Summing out the discrete parameters
Although Stan does not (yet) support discrete sampling, it is possible to calculate the marginal distribution over the continuous parameters by summing out the discrete parameters as in other mixture models. The marginal posterior of the topic and word variables is

\[
p(\theta, \phi \mid w, \alpha, \beta) \propto p(\theta \mid \alpha) p(\phi \mid \beta) p(w \mid \theta, \phi)
\]

\[
= \prod_{m=1}^{M} p(\theta_m \mid \alpha) \times \prod_{k=1}^{K} p(\phi_k \mid \beta) \times \prod_{m=1}^{M} \prod_{n=1}^{M[n]} p(w_{m,n} \mid \theta_m, \phi).
\]

The inner word-probability term is defined by summing out the topic assignments,

\[
p(w_{m,n} \mid \theta_m, \phi) = \sum_{z=1}^{K} p(z, w_{m,n} \mid \theta_m, \phi)
\]

\[
= \sum_{z=1}^{K} p(z \mid \theta_m) p(w_{m,n} \mid \phi_z).
\]

Plugging the distributions in and converting to the log scale provides a formula that can be implemented directly in Stan,

\[
\log p(\theta, \phi \mid w, \alpha, \beta)
\]

\[
= \sum_{m=1}^{M} \log \text{Dirichlet}(\theta_m \mid \alpha) + \sum_{k=1}^{K} \log \text{Dirichlet}(\phi_k \mid \beta)
\]

\[
+ \sum_{m=1}^{M} \sum_{n=1}^{N[m]} \log \left( \sum_{z=1}^{K} \text{categorical}(z \mid \theta_m) \times \text{categorical}(w_{m,n} \mid \phi_z) \right)
\]

Implementation of LDA
Applying the marginal derived in the last section to the data structure described in this section leads to the following Stan program for LDA.

data {
  int<lower=2> K; // num topics
  int<lower=2> V; // num words
  int<lower=1> M; // num docs
  int<lower=1> N; // total word instances
  int<lower=1,upper=V> w[N]; // word n
int<lower=1,upper=M> doc[N]; // doc ID for word n  
vector<lower=0>[K] alpha; // topic prior  
vector<lower=0>[V] beta; // word prior  
}
parameters {  
simplex[K] theta[M]; // topic dist for doc m  
simplex[V] phi[K]; // word dist for topic k  
}
model {  
  for (m in 1:M)  
    theta[m] ~ dirichlet(alpha); // prior  
  for (k in 1:K)  
    phi[k] ~ dirichlet(beta); // prior  
  for (n in 1:N) {  
    real gamma[K];  
    for (k in 1:K)  
      gamma[k] = log(theta[doc[n], k]) + log(phi[k, w[n]]);  
    target += log_sum_exp(gamma); // likelihood;  
  }
}

As in the other mixture models, the log-sum-of-exponents function is used to stabilize the numerical arithmetic.

**Correlated topic model**

To account for correlations in the distribution of topics for documents, Blei and Lafferty (2007) introduced a variant of LDA in which the Dirichlet prior on the per-document topic distribution is replaced with a multivariate logistic normal distribution.

The authors treat the prior as a fixed hyperparameter. They use an $L_1$-regularized estimate of covariance, which is equivalent to the maximum a posteriori estimate given a double-exponential prior. Stan does not (yet) support maximum a posteriori estimation, so the mean and covariance of the multivariate logistic normal must be specified as data.

**Fixed hyperparameter correlated topic model**

The Stan model in the previous section can be modified to implement the correlated topic model by replacing the Dirichlet topic prior alpha in the data declaration with the mean and covariance of the multivariate logistic normal prior.
Rather than drawing the simplex parameter \( \theta \) from a Dirichlet, a parameter \( \eta \) is drawn from a multivariate normal distribution and then transformed using softmax into a simplex.

```
data {
    ... data as before without alpha ...
    vector[K] mu;       // topic mean
    cov_matrix[K] Sigma; // topic covariance
}
```

```
parameters {
    simplex[V] phi[K];     // word dist for topic k
    vector[K] eta[M];      // topic dist for doc m
}
```

```
transformed parameters {
    simplex[K] theta[M];
    for (m in 1:M)
        theta[m] = softmax(eta[m]);
}
```

```
model {
    for (m in 1:M)
        eta[m] ~ multi_normal(mu, Sigma);
    ... model as before w/o prior for theta ...
}
```

**Full Bayes correlated topic model**

By adding a prior for the mean and covariance, Stan supports full Bayesian inference for the correlated topic model. This requires moving the declarations of topic mean \( \mu \) and covariance \( \Sigma \) from the data block to the parameters block and providing them with priors in the model. A relatively efficient and interpretable prior for the covariance matrix \( \Sigma \) may be encoded as follows.

```
data block as before, but without alpha ...
parameters {
    vector[K] mu;       // topic mean
    corr_matrix[K] Omega; // correlation matrix
    vector<lower=0>[K] sigma; // scales
    vector[K] eta[M];    // logit topic dist for doc m
    simplex[V] phi[K];   // word dist for topic k
}
```
transformed parameters {
    ... eta as above ...

cov_matrix[K] Sigma; // covariance matrix
for (m in 1:K)
    Sigma[m, m] = sigma[m] * sigma[m] * Omega[m, m];
for (m in 1:(K-1)) {
    for (n in (m+1):K) {
        Sigma[m, n] = sigma[m] * sigma[n] * Omega[m, n];
        Sigma[n, m] = Sigma[m, n];
    }
}
}

model {
    mu ~ normal(0, 5); // vectorized, diffuse
    Omega ~ lkj_corr(2.0); // regularize to unit correlation
    sigma ~ cauchy(0, 5); // half-Cauchy due to constraint
    ... words sampled as above ...
}

The LKJCorr distribution with shape $\alpha > 0$ has support on correlation matrices (i.e., symmetric positive definite with unit diagonal). Its density is defined by

$$LkjCorr(\Omega | \alpha) \propto \det(\Omega)^{\alpha - 1}$$

With a scale of $\alpha = 2$, the weakly informative prior favors a unit correlation matrix. Thus the compound effect of this prior on the covariance matrix $\Sigma$ for the multivariate logistic normal is a slight concentration around diagonal covariance matrices with scales determined by the prior on $\sigma$. 
10. Gaussian Processes

Gaussian processes are continuous stochastic processes and thus may be interpreted as providing a probability distribution over functions. A probability distribution over continuous functions may be viewed, roughly, as an uncountably infinite collection of random variables, one for each valid input. The generality of the supported functions makes Gaussian priors popular choices for priors in general multivariate (non-linear) regression problems.

The defining feature of a Gaussian process is that the joint distribution of the function’s value at a finite number of input points is a multivariate normal distribution. This makes it tractable to both fit models from finite amounts of observed data and make predictions for finitely many new data points.

Unlike a simple multivariate normal distribution, which is parameterized by a mean vector and covariance matrix, a Gaussian process is parameterized by a mean function and covariance function. The mean and covariance functions apply to vectors of inputs and return a mean vector and covariance matrix which provide the mean and covariance of the outputs corresponding to those input points in the functions drawn from the process.

Gaussian processes can be encoded in Stan by implementing their mean and covariance functions and plugging the result into the Gaussian form of their sampling distribution, or by using the specialized covariance functions outlined below. This form of model is straightforward and may be used for simulation, model fitting, or posterior predictive inference. A more efficient Stan implementation for the GP with a normally distributed outcome marginalizes over the latent Gaussian process, and applies a Cholesky-factor reparameterization of the Gaussian to compute the likelihood and the posterior predictive distribution analytically.

After defining Gaussian processes, this chapter covers the basic implementations for simulation, hyperparameter estimation, and posterior predictive inference for univariate regressions, multivariate regressions, and multivariate logistic regressions. Gaussian processes are general, and by necessity this chapter only touches on some basic models. For more information, see Rasmussen and Williams (2006).

10.1. Gaussian process regression

The data for a multivariate Gaussian process regression consists of a series of $N$ inputs $x_1, \ldots, x_N \in \mathbb{R}^D$ paired with outputs $y_1, \ldots, y_N \in \mathbb{R}$. The defining feature of
Gaussian processes is that the probability of a finite number of outputs $y$ conditioned on their inputs $x$ is Gaussian:

$$ y \sim \text{multivariate normal}(m(x), K(x \mid \theta)),$$

where $m(x)$ is an $N$-vector and $K(x \mid \theta)$ is an $N \times N$ covariance matrix. The mean function $m : \mathbb{R}^{N \times D} \to \mathbb{R}^N$ can be anything, but the covariance function $K : \mathbb{R}^{N \times D} \to \mathbb{R}^{N \times N}$ must produce a positive-definite matrix for any input $x$.$^1$

A popular covariance function, which will be used in the implementations later in this chapter, is an exponentiated quadratic function,

$$ K(x \mid \alpha, \rho, \sigma)_{i,j} = \alpha^2 \exp\left(-\frac{1}{2\rho^2} \sum_{d=1}^{D} (x_{i,d} - x_{j,d})^2\right) + \delta_{i,j} \sigma^2,$$

where $\alpha$, $\rho$, and $\sigma$ are hyperparameters defining the covariance function and where $\delta_{i,j}$ is the Kronecker delta function with value 1 if $i = j$ and value 0 otherwise; this test is between the indexes $i$ and $j$, not between values $x_i$ and $x_j$. This kernel is obtained through a convolution of two independent Gaussian processes, $f_1$ and $f_2$, with kernels

$$ K_1(x \mid \alpha, \rho)_{i,j} = \alpha^2 \exp\left(-\frac{1}{2\rho^2} \sum_{d=1}^{D} (x_{i,d} - x_{j,d})^2\right) $$

and

$$ K_2(x \mid \sigma)_{i,j} = \delta_{i,j} \sigma^2,$$

The addition of $\sigma^2$ on the diagonal is important to ensure the positive definiteness of the resulting matrix in the case of two identical inputs $x_i = x_j$. In statistical terms, $\sigma$ is the scale of the noise term in the regression.

The hyperparameter $\rho$ is the length-scale, and corresponds to the frequency of the functions represented by the Gaussian process prior with respect to the domain. Values of $\rho$ closer to zero lead the GP to represent high-frequency functions, whereas larger values of $\rho$ lead to low-frequency functions. The hyperparameter $\alpha$ is the marginal standard deviation. It controls the magnitude of the range of the function represented by the GP. If you were to take the standard deviation of many draws from the GP $f_1$ prior at a single input $x$ conditional on one value of $\alpha$ one would recover $\alpha$.$^1$

$^1$Gaussian processes can be extended to covariance functions producing positive semi-definite matrices, but Stan does not support inference in the resulting models because the resulting distribution does not have unconstrained support.
The only term in the squared exponential covariance function involving the inputs $x_i$ and $x_j$ is their vector difference, $x_i - x_j$. This produces a process with stationary covariance in the sense that if an input vector $x$ is translated by a vector $\epsilon$ to $x + \epsilon$, the covariance at any pair of outputs is unchanged, because $K(x | \theta) = K(x + \epsilon | \theta)$.

The summation involved is just the squared Euclidean distance between $x_i$ and $x_j$ (i.e., the $L_2$ norm of their difference, $x_i - x_j$). This results in support for smooth functions in the process. The amount of variation in the function is controlled by the free hyperparameters $\alpha$, $\rho$, and $\sigma$.

Changing the notion of distance from Euclidean to taxicab distance (i.e., an $L_1$ norm) changes the support to functions which are continuous but not smooth.

**10.2. Simulating from a Gaussian process**

It is simplest to start with a Stan model that does nothing more than simulate draws of functions $f$ from a Gaussian process. In practical terms, the model will draw values $y_n = f(x_n)$ for finitely many input points $x_n$.

The Stan model defines the mean and covariance functions in a transformed data block and then samples outputs $y$ in the model using a multivariate normal distribution. To make the model concrete, the squared exponential covariance function described in the previous section will be used with hyperparameters set to $\alpha^2 = 1$, $\rho^2 = 1$, and $\sigma^2 = 0.1$, and the mean function $m$ is defined to always return the zero vector, $m(x) = 0$. Consider the following implementation of a Gaussian process simulator:

```stan
data {
  int<lower=1> N;
  real x[N];
}
transformed data {
  matrix[N, N] K;
  vector[N] mu = rep_vector(0, N);
  for (i in 1:(N - 1)) {
    K[i, i] = 1 + 0.1;
    for (j in (i + 1):N) {
      K[i, j] = exp(-0.5 * square(x[i] - x[j]));
      K[j, i] = K[i, j];
    }
  }
  K[N, N] = 1 + 0.1;
}
```
parameters {
  vector[N] y;
}

model {
  y ~ multi_normal(mu, K);
}

The above model can also be written more compactly using the specialized covariance function that implements the exponentiated quadratic kernel.

data {
  int<lower=1> N;
  real x[N];
}

transformed data {
  matrix[N, N] K = cov_exp_quad(x, 1.0, 1.0);
  vector[N] mu = rep_vector(0, N);
  for (n in 1:N)
    K[n, n] = K[n, n] + 0.1;
}

parameters {
  vector[N] y;
}

model {
  y ~ multi_normal(mu, K);
}

The input data are just the vector of inputs \( x \) and its size \( N \). Such a model can be used with values of \( x \) evenly spaced over some interval in order to plot sample draws of functions from a Gaussian process.

**Multivariate inputs**

Only the input data needs to change in moving from a univariate model to a multivariate model.

The only lines that change from the univariate model above are as follows.

data {
  int<lower=1> N;
  int<lower=1> D;
  vector[D] x[N];
}
transformed data {
...

The data are now declared as an array of vectors instead of an array of scalars; the dimensionality \( D \) is also declared.

In the remainder of the chapter, univariate models will be used for simplicity, but any of the models could be changed to multivariate in the same way as the simple sampling model. The only extra computational overhead from a multivariate model is in the distance calculation.

**Cholesky factored and transformed implementation**

A more efficient implementation of the simulation model can be coded in Stan by relocating, rescaling and rotating an isotropic standard normal variate. Suppose \( \eta \) is an isotropic standard normal variate

\[
\eta \sim \text{normal}(0, 1),
\]

where \( 0 \) is an \( N \)-vector of 0 values and \( 1 \) is the \( N \times N \) identity matrix. Let \( L \) be the Cholesky decomposition of \( K(x \mid \theta) \), i.e., the lower-triangular matrix \( L \) such that \( LL^\top = K(x \mid \theta) \). Then the transformed variable \( \mu + L\eta \) has the intended target distribution, \( \mu + L\eta \sim \text{multivariate normal}(\mu(x), K(x \mid \theta)) \).

This transform can be applied directly to Gaussian process simulation.

This model has the same data declarations for \( N \) and \( x \), and the same transformed data definitions of \( \mu_0 \) and \( K \) as the previous model, with the addition of a transformed data variable for the Cholesky decomposition. The parameters change to the raw parameters sampled from an isotropic standard normal, and the actual samples are defined as generated quantities.

...  
transformed data {
    matrix[N, N] L;
    ...
    L = cholesky_decompose(K);
}
parameters {
    vector[N] eta;
}
model {
    eta ~ std_normal();
}
The Cholesky decomposition is only computed once, after the data are loaded and the covariance matrix $K$ computed. The isotropic normal distribution for $\eta$ is specified as a vectorized univariate distribution for efficiency; this specifies that each $\eta[n]$ has an independent standard normal distribution. The sampled vector $y$ is then defined as a generated quantity using a direct encoding of the transform described above.

## 10.3. Fitting a Gaussian process

### GP with a normal outcome

The full generative model for a GP with a normal outcome, $y \in \mathbb{R}^N$, with inputs $x \in \mathbb{R}^N$, for a finite $N$:

\[
\begin{align*}
\rho &\sim \text{InvGamma}(5, 5) \\
\alpha &\sim \text{normal}(0, 1) \\
\sigma &\sim \text{normal}(0, 1) \\
f &\sim \text{multivariate normal} \left(0, K(x | \alpha, \rho)\right) \\
y_i &\sim \text{normal}(f_i, \sigma) \forall i \in \{1, \ldots, N\}
\end{align*}
\]

With a normal outcome, it is possible to integrate out the Gaussian process $f$, yielding the more parsimonious model:

\[
\begin{align*}
\rho &\sim \text{InvGamma}(5, 5) \\
\alpha &\sim \text{normal}(0, 1) \\
\sigma &\sim \text{normal}(0, 1) \\
y &\sim \text{multivariate normal} \left(0, K(x | \alpha, \rho) + I_N \sigma^2\right)
\end{align*}
\]

It can be more computationally efficient when dealing with a normal outcome to integrate out the Gaussian process, because this yields a lower-dimensional parameter space over which to do inference. We’ll fit both models in Stan. The former model will be referred to as the latent variable GP, while the latter will be called the marginal likelihood GP.
The hyperparameters controlling the covariance function of a Gaussian process can be fit by assigning them priors, like we have in the generative models above, and then computing the posterior distribution of the hyperparameters given observed data. The priors on the parameters should be defined based on prior knowledge of the scale of the output values ($\alpha$), the scale of the output noise ($\sigma$), and the scale at which distances are measured among inputs ($\rho$). See the Gaussian process priors section for more information about how to specify appropriate priors for the hyperparameters.

The Stan program implementing the marginal likelihood GP is shown below. The program is similar to the Stan programs that implement the simulation GPs above, but because we are doing inference on the hyperparameters, we need to calculate the covariance matrix $K$ in the model block, rather than the transformed data block.

data {
  int<lower=1> N;
  real x[N];
  vector[N] y;
}
transformed data {
  vector[N] mu = rep_vector(0, N);
}
parameters {
  real<lower=0> rho;
  real<lower=0> alpha;
  real<lower=0> sigma;
}
model {
  matrix[N, N] L_K;
  matrix[N, N] K = cov_exp_quad(x, alpha, rho);
  real sq_sigma = square(sigma);

  // diagonal elements
  for (n in 1:N)
    K[n, n] = K[n, n] + sq_sigma;

  L_K = cholesky_decompose(K);

  rho ~ inv_gamma(5, 5);
  alpha ~ std_normal();
  sigma ~ std_normal();
sigma ~ std_normal();

y ~ multi_normal_cholesky(mu, L_K);
}

The data block declares a vector $y$ of observed values $y[n]$ for inputs $x[n]$. The transformed data block now only defines the mean vector to be zero. The three hyperparameters are defined as parameters constrained to be non-negative. The computation of the covariance matrix $K$ is now in the model block because it involves unknown parameters and thus can’t simply be precomputed as transformed data. The rest of the model consists of the priors for the hyperparameters and the multivariate Cholesky-parameterized normal likelihood, only now the value $y$ is known and the covariance matrix $K$ is an unknown dependent on the hyperparameters, allowing us to learn the hyperparameters.

We have used the Cholesky parameterized multivariate normal rather than the standard parameterization because it allows us to the cholesky_decompose function which has been optimized for both small and large matrices. When working with small matrices the differences in computational speed between the two approaches will not be noticeable, but for larger matrices ($N \gtrsim 100$) the Cholesky decomposition version will be faster.

Hamiltonian Monte Carlo sampling is fast and effective for hyperparameter inference in this model (Neal 1997). If the posterior is well-concentrated for the hyperparameters the Stan implementation will fit hyperparameters in models with a few hundred data points in seconds.

*Latent variable GP*

We can also explicitly code the latent variable formulation of a GP in Stan. This will be useful for when the outcome is not normal. We’ll need to add a small positive term, $\delta$ to the diagonal of the covariance matrix in order to ensure that our covariance matrix remains positive definite.

data {
    int<lower=1> N;
    real x[N];
    vector[N] y;
}
transformed data {
    real delta = 1e-9;
}
parameters {
    real<lower=0> rho;
    real<lower=0> alpha;
    real<lower=0> sigma;
    vector[N] eta;
}
model {
    vector[N] f;
    { 
        matrix[N, N] L_K;
        matrix[N, N] K = cov_exp_quad(x, alpha, rho);

        // diagonal elements
        for (n in 1:N)
            K[n, n] = K[n, n] + delta;

        L_K = cholesky_decompose(K);
        f = L_K * eta;
    }

    rho ~ inv_gamma(5, 5);
    alpha ~ std_normal();
    sigma ~ std_normal();
    eta ~ std_normal();

    y ~ normal(f, sigma);
}

Two differences between the latent variable GP and the marginal likelihood GP are worth noting. The first is that we have augmented our parameter block with a new parameter vector of length $N$ called $\eta$. This is used in the model block to generate a multivariate normal vector called $f$, corresponding to the latent GP. We put a $\text{normal}(0, 1)$ prior on $\eta$ like we did in the Cholesky-parameterized GP in the simulation section. The second difference is that our likelihood is now univariate, though we could code $N$ likelihood terms as one $N$-dimensional multivariate normal with an identity covariance matrix multiplied by $\sigma^2$. However, it is more efficient to use the vectorized statement as shown above.
Discrete outcomes with Gaussian processes
Gaussian processes can be generalized the same way as standard linear models by introducing a link function. This allows them to be used as discrete data models.

**Poisson GP**

If we want to model count data, we can remove the $\sigma$ parameter, and use `poisson_log`, which implements a log link, for our likelihood rather than `normal`. We can also add an overall mean parameter, $a$, which will account for the marginal expected value for $y$. We do this because we cannot center count data like we would for normally distributed data.

data {  ...
  int<lower=0> y[N];  ...
}

parameters {
  ...
  real<lower=0> rho;
  real<lower=0> alpha;
  real a;
  vector[N] eta;
}

model {
  ...
  rho ~ inv_gamma(5, 5);
  alpha ~ std_normal();
  a ~ std_normal();
  eta ~ std_normal();

  y ~ poisson_log(a + f);
}

**Logistic Gaussian process regression**

For binary classification problems, the observed outputs $z_n \in \{0, 1\}$ are binary. These outputs are modeled using a Gaussian process with (unobserved) outputs $y_n$ through the logistic link,

$$z_n \sim \text{Bernoulli}(\logit^{-1}(y_n)),$$
or in other words,

$$\Pr[z_n = 1] = \text{logit}^{-1}(y_n).$$

We can extend our latent variable GP Stan program to deal with classification problems. Below a is the bias term, which can help account for imbalanced classes in the training data:

```stan
data {
  ...
  int<lower=0, upper=1> z[N];
  ...
}
...
model {
  ...
  y ~ bernoulli_logit(a + f);
}
```

**Automatic relevance determination**

If we have multivariate inputs $x \in \mathbb{R}^D$, the squared exponential covariance function can be further generalized by fitting a scale parameter $\rho_d$ for each dimension $d$,

$$k(x \mid \alpha, \bar{\rho}, \sigma)_{i,j} = \alpha^2 \exp \left( -\frac{1}{2} \sum_{d=1}^{D} \frac{1}{\rho_d^2} (x_{i,d} - x_{j,d})^2 \right) + \delta_{i,j} \sigma^2.$$ 

The estimation of $\rho$ was termed “automatic relevance determination” by Neal (1996a), but this is misleading, because the magnitude of the scale of the posterior for each $\rho_d$ is dependent on the scaling of the input data along dimension $d$. Moreover, the scale of the parameters $\rho_d$ measures non-linearity along the $d$-th dimension, rather than “relevance” (Piironen and Vehtari 2016).

A priori, the closer $\rho_d$ is to zero, the more nonlinear the conditional mean in dimension $d$ is. A posteriori, the actual dependencies between $x$ and $y$ play a role. With one covariate $x_1$ having a linear effect and another covariate $x_2$ having a nonlinear effect, it is possible that $\rho_1 > \rho_2$ even if the predictive relevance of $x_1$ is higher (Rasmussen and Williams 2006, 80). The collection of $\rho_d$ (or $1/\rho_d$) parameters can also be modeled hierarchically.

The implementation of automatic relevance determination in Stan is straightforward, though it currently requires the user to directly code the covariance matrix.
We'll write a function to generate the Cholesky of the covariance matrix called L_cov_exp_quad_ARD.

functions {
  matrix L_cov_exp_quad_ARD(vector[] x,
                   real alpha,
                   vector rho,
                   real delta) {

    int N = size(x);
    matrix[N, N] K;
    real sq_alpha = square(alpha);
    for (i in 1:(N-1)) {
      K[i, i] = sq_alpha + delta;
      for (j in (i + 1):N) {
        K[i, j] = sq_alpha
          * exp(-0.5 * dot_self((x[i] - x[j]) ./ rho));
        K[j, i] = K[i, j];
      }
    }
    K[N, N] = sq_alpha + delta;
    return cholesky_decompose(K);
  }
}

data {
  int<lower=1> N;
  int<lower=1> D;
  vector[D] x[N];
  vector[N] y;
}

transformed data {
  real delta = 1e-9;
}

parameters {
  vector<lower=0>[D] rho;
  real<lower=0> alpha;
  real<lower=0> sigma;
  vector[N] eta;
}

model {
  vector[N] f;
{ matrix[N, N] L_K = L_cov_exp_quad_ARD(x, alpha, rho, delta); 
f = L_K * eta;
}

rho ~ inv_gamma(5, 5);
alpha ~ std_normal();
sigma ~ std_normal();
eta ~ std_normal();

y ~ normal(f, sigma);

Priors for Gaussian process parameters

Formulating priors for GP hyperparameters requires the analyst to consider the inherent statistical properties of a GP, the GP’s purpose in the model, and the numerical issues that may arise in Stan when estimating a GP.

Perhaps most importantly, the parameters $\rho$ and $\alpha$ are weakly identified (Zhang 2004). The ratio of the two parameters is well-identified, but in practice we put independent priors on the two hyperparameters because these two quantities are more interpretable than their ratio.

Priors for length-scale

GPs are a flexible class of priors and, as such, can represent a wide spectrum of functions. For length scales below the minimum spacing of the covariates the GP likelihood plateaus. Unless regularized by a prior, this flat likelihood induces considerable posterior mass at small length scales where the observation variance drops to zero and the functions supported by the GP being to exactly interpolate between the input data. The resulting posterior not only significantly overfits to the input data, it also becomes hard to accurately sample using Euclidean HMC.

We may wish to put further soft constraints on the length-scale, but these are dependent on how the GP is used in our statistical model.

If our model consists of only the GP, i.e.:

\[
\begin{align*}
    f & \sim \text{multivariate normal} \left( 0, K(x \mid \alpha, \rho) \right) \\
    y_i & \sim \text{normal}(f_i, \sigma) \quad \forall i \in \{1, \ldots, N\} \\
    x & \in \mathbb{R}^{N \times D}, \quad f \in \mathbb{R}^N
\end{align*}
\]
we likely don’t need constraints beyond penalizing small length-scales. We’d like to allow the GP prior to represent both high-frequency and low-frequency functions, so our prior should put non-negligible mass on both sets of functions. In this case, an inverse gamma, \( \text{inv\_gamma\_lpdf} \) in Stan’s language, will work well as it has a sharp left tail that puts negligible mass on infinitesimal length-scales, but a generous right tail, allowing for large length-scales. Inverse gamma priors will avoid infinitesimal length-scales because the density is zero at zero, so the posterior for length-scale will be pushed away from zero. An inverse gamma distribution is one of many zero-avoiding or boundary-avoiding distributions.\(^2\).

If we’re using the GP as a component in a larger model that includes an overall mean and fixed effects for the same variables we’re using as the domain for the GP, i.e.:

\[
\begin{align*}
  f &\sim \text{multivariate normal}(0, K(x | \alpha, \rho)) \\
  y_i &\sim \text{normal}(\beta_0 + x_i \beta_{[1:D]} + f_i, \sigma) \quad \forall i \in \{1, \ldots, N\} \\
  x_i^T, \beta_{[1:D]} &\in \mathbb{R}^D, \quad x \in \mathbb{R}^{N \times D}, \quad f \in \mathbb{R}^N
\end{align*}
\]

we’ll likely want to constrain large length-scales as well. A length scale that is larger than the scale of the data yields a GP posterior that is practically linear (with respect to the particular covariate) and increasing the length scale has little impact on the likelihood. This will introduce nonidentifiability in our model, as both the fixed effects and the GP will explain similar variation. In order to limit the amount of overlap between the GP and the linear regression, we should use a prior with a sharper right tail to limit the GP to higher-frequency functions. We can use a generalized inverse Gaussian distribution:

\[
f(x | a, b, p) = \frac{(a/b)^{p/2}}{2K_p(\sqrt{ab})} x^{p-1} \exp\left( - \frac{(ax + b/x)}{2} \right)
\]

which has an inverse gamma left tail if \( p \leq 0 \) and an inverse Gaussian right tail. This has not yet been implemented in Stan’s math library, but it is possible to implement as a user defined function:

```stan
functions {
  real generalized_inverse_gaussian_lpdf(real x, int p, real a, real b) {
    // Implementation
  }
}
```

\(^2\)A boundary-avoiding prior is just one where the limit of the density is zero at the boundary, the result of which is estimates that are pushed away from the boundary.
If we have high-frequency covariates in our fixed effects, we may wish to further regularize the GP away from high-frequency functions, which means we’ll need to penalize smaller length-scales. Luckily, we have a useful way of thinking about how length-scale affects the frequency of the functions supported the GP. If we were to repeatedly draw from a zero-mean GP with a length-scale of \( \rho \) in a fixed-domain \([0, T]\), we would get a distribution for the number of times each draw of the GP crossed the zero axis. The expectation of this random variable, the number of zero crossings, is \( T/\pi \rho \). You can see that as \( \rho \) decreases, the expectation of the number of upcrossings increases as the GP is representing higher-frequency functions. Thus, this is a good statistic to keep in mind when setting a lower-bound for our prior on length-scale in the presence of high-frequency covariates. However, this statistic is only valid for one-dimensional inputs.

**Priors for marginal standard deviation**

The parameter \( \alpha \) corresponds to how much of the variation is explained by the regression function and has a similar role to the prior variance for linear model weights. This means the prior can be the same as used in linear models, such as a half-\( t \) prior on \( \alpha \).

A half-\( t \) or half-Gaussian prior on alpha also has the benefit of putting nontrivial prior mass around zero. This allows the GP support the zero functions and allows the possibility that the GP won’t contribute to the conditional mean of the total output.

**Predictive inference with a Gaussian process**

Suppose for a given sequence of inputs \( x \) that the corresponding outputs \( y \) are observed. Given a new sequence of inputs \( \tilde{x} \), the posterior predictive distribution of their labels is computed by sampling outputs \( \tilde{y} \) according to

\[
p ( \tilde{y} \mid \tilde{x}, x, y ) = \frac{p ( \tilde{y}, y \mid \tilde{x}, x )}{p(y \mid x)} \propto p ( \tilde{y}, y \mid \tilde{x}, x ) .
\]

A direct implementation in Stan defines a model in terms of the joint distribution of
the observed \( y \) and unobserved \( \tilde{y} \).

data {
  int<lower=1> N1;
  real x1[N1];
  vector[N1] y1;
  int<lower=1> N2;
  real x2[N2];
}
transformed data {
  real delta = 1e-9;
  int<lower=1> N = N1 + N2;
  real x[N];
  for (n1 in 1:N1) x[n1] = x1[n1];
  for (n2 in 1:N2) x[N1 + n2] = x2[n2];
}
parameters {
  real<lower=0> rho;
  real<lower=0> alpha;
  real<lower=0> sigma;
  vector[N] eta;
}
transformed parameters {
  vector[N] f;
  {
    matrix[N, N] L_K;
    matrix[N, N] K = cov_exp_quad(x, alpha, rho);
    // diagonal elements
    for (n in 1:N)
      K[n, n] = K[n, n] + delta;
    L_K = cholesky_decompose(K);
    f = L_K * eta;
  }
}
model {
  rho ~ inv_gamma(5, 5);
  alpha ~ std_normal();
  sigma ~ std_normal();
}
eta ~ std_normal();

y1 ~ normal(f[1:N1], sigma);
}
generated quantities {
  vector[N2] y2;
  for (n2 in 1:N2)
    y2[n2] = normal_rng(f[N1 + n2], sigma);
}

The input vectors $x_1$ and $x_2$ are declared as data, as is the observed output vector $y_1$. The unknown output vector $y_2$, which corresponds to input vector $x_2$, is declared in the generated quantities block and will be sampled when the model is executed.

A transformed data block is used to combine the input vectors $x_1$ and $x_2$ into a single vector $x$.

The model block declares and defines a local variable for the combined output vector $f$, which consists of the concatenation of the conditional mean for known outputs $y_1$ and unknown outputs $y_2$. Thus the combined output vector $f$ is aligned with the combined input vector $x$. All that is left is to define the univariate normal sampling statement for $y$.

The generated quantities block defines the quantity $y_2$. We generate $y_2$ by sampling $N_2$ univariate normals with each mean corresponding to the appropriate element in $f$.

**Predictive inference in non-Gaussian GPs**

We can do predictive inference in non-Gaussian GPs in much the same way as we do with Gaussian GPs.

Consider the following full model for prediction using logistic Gaussian process regression.

data {
  int<lower=1> N1;
  real x1[N1];
  int<lower=0, upper=1> z1[N1];
  int<lower=1> N2;
  real x2[N2];
}
transformed data {

real delta = 1e-9;
int<lower=1> N = N1 + N2;
real x[N];
for (n1 in 1:N1) x[n1] = x1[n1];
for (n2 in 1:N2) x[N1 + n2] = x2[n2];
}
parameters {
  real<lower=0> rho;
  real<lower=0> alpha;
  real a;
  vector[N] eta;
}
transformed parameters {
  vector[N] f;
  {
    matrix[N, N] L_K;
    matrix[N, N] K = cov_exp_quad(x, alpha, rho);

    // diagonal elements
    for (n in 1:N)
      K[n, n] = K[n, n] + delta;

    L_K = cholesky_decompose(K);
    f = L_K * eta;
  }
}
model {
  rho ~ inv_gamma(5, 5);
  alpha ~ std_normal();
  a ~ std_normal();
  eta ~ std_normal();

  z1 ~ bernoulli_logit(a + f[1:N1]);
}
generated quantities {
  int z2[N2];
  for (n2 in 1:N2)
    z2[n2] = bernoulli_logit_rng(a + f[N1 + n2]);
}
Analytical form of joint predictive inference

Bayesian predictive inference for Gaussian processes with Gaussian observations can be sped up by deriving the posterior analytically, then directly sampling from it. Jumping straight to the result,

\[
p (\tilde{y} \mid \tilde{x}, y, x) = \text{normal} \left( K^\top \Sigma^{-1} y, \Omega - K^\top \Sigma^{-1} K \right),
\]

where \( \Sigma = K(x \mid \alpha, \rho, \sigma) \) is the result of applying the covariance function to the inputs \( x \) with observed outputs \( y \), \( \Omega = K(\tilde{x} \mid \alpha, \rho) \) is the result of applying the covariance function to the inputs \( \tilde{x} \) for which predictions are to be inferred, and \( K \) is the matrix of covariances between inputs \( x \) and \( \tilde{x} \), which in the case of the exponentiated quadratic covariance function would be

\[
K(x \mid \alpha, \rho)_{i,j} = \eta^2 \exp \left( -\frac{1}{2\rho^2} \sum_{d=1}^{D} (x_{i,d} - \tilde{x}_{j,d})^2 \right).
\]

There is no noise term including \( \sigma^2 \) because the indexes of elements in \( x \) and \( \tilde{x} \) are never the same.

This Stan code below uses the analytic form of the posterior and provides sampling of the resulting multivariate normal through the Cholesky decomposition. The data declaration is the same as for the latent variable example, but we’ve defined a function called `gp_pred_rng` which will generate a draw from the posterior predictive mean conditioned on observed data \( y_1 \). The code uses a Cholesky decomposition in triangular solves in order to cut down on the number of matrix-matrix multiplications when computing the conditional mean and the conditional covariance of \( p(\tilde{y}) \).

```stan
functions {
  vector gp_pred_rng(real[] x2,
                      vector y1,
                      real[] x1,
                      real alpha,
                      real rho,
                      real sigma,
                      real delta) {
    int N1 = rows(y1);
    int N2 = size(x2);
    vector[N2] f2;
    {
```

matrix[N1, N1] L_K;
vector[N1] K_div_y1;
matrix[N1, N2] k_x1_x2;
matrix[N1, N2] v_pred;
vector[N2] f2_mu;
matrix[N2, N2] cov_f2;
matrix[N2, N2] diag_delta;
matrix[N1, N1] K;
K = cov_exp_quad(x1, alpha, rho);
for (n in 1:N1)
    K[n, n] = K[n,n] + square(sigma);
L_K = cholesky_decompose(K);
K_div_y1 = mdivide_left_tri_low(L_K, y1);
K_div_y1 = mdivide_right_tri_low(K_div_y1', L_K)';
k_x1_x2 = cov_exp_quad(x1, x2, alpha, rho);
f2_mu = (k_x1_x2' * K_div_y1);
v_pred = mdivide_left_tri_low(L_K, k_x1_x2);
cov_f2 = cov_exp_quad(x2, alpha, rho) - v_pred' * v_pred;
diag_delta = diag_matrix(rep_vector(delta, N2));

f2 = multi_normal_rng(f2_mu, cov_f2 + diag_delta);
}
return f2;
}

data {
    int<lower=1> N1;
    real x1[N1];
    vector[N1] y1;
    int<lower=1> N2;
    real x2[N2];
}
transformed data {
    vector[N1] mu = rep_vector(0, N1);
    real delta = 1e-9;
}
parameters {
    real<lower=0> rho;
    real<lower=0> alpha;
Multiple-output Gaussian processes

Suppose we have observations $y_i \in \mathbb{R}^M$ observed at $x_i \in \mathbb{R}^K$. One can model the data like so:

$$y_i \sim \text{multivariate normal} \left( f(x_i), I_M \sigma^2 \right)$$

$$f(x) \sim \text{GP} \left( m(x), K(x \mid \theta, \phi) \right)$$

$$K(x \mid \theta) \in \mathbb{R}^{M \times M}, \quad f(x), m(x) \in \mathbb{R}^M$$

where the $K(x, x' \mid \theta, \phi)_{m,m'}$ entry defines the covariance between $f_m(x)$ and $f_{m'}(x')(x)$. This construction of Gaussian processes allows us to learn the covariance...
between the output dimensions of $f(x)$. If we parameterize our kernel $K$:

$$K(x, x' \mid \theta, \phi)[m, m'] = k(x, x' \mid \theta) k(m, m' \mid \phi)$$

then our finite dimensional generative model for the above is:

$$f \sim \text{matrixnormal}(m(x), K(x \mid \alpha, \rho), C(\phi))$$

$$y_{i,m} \sim \text{normal}(f_{i,m}, \sigma)$$

$$f \in \mathbb{R}^{N \times M}$$

where $K(x \mid \alpha, \rho)$ is the exponentiated quadratic kernel we've used throughout this chapter, and $C(\phi)$ is a positive-definite matrix, parameterized by some vector $\phi$.

The matrix normal distribution has two covariance matrices: $K(x \mid \alpha, \rho)$ to encode column covariance, and $C(\phi)$ to define row covariance. The salient features of the matrix normal are that the rows of the matrix $f$ are distributed:

$$f_{[n,\cdot]} \sim \text{multivariate normal}(m(x)_{[n,\cdot]}, K(x \mid \alpha, \rho)_{[n,n]} C(\phi))$$

and that the columns of the matrix $f$ are distributed:

$$f_{[\cdot,m]} \sim \text{multivariate normal}(m(x)_{[\cdot,m]}, K(x \mid \alpha, \rho)_{[m,m]} C(\phi))$$

This also means that $\mathbb{E}[f^T f] = \text{trace}(K(x \mid \alpha, \rho)) \times C$, whereas $\mathbb{E}[ff^T] = \text{trace}(C) \times K(x \mid \alpha, \rho)$. We can derive this using properties of expectation and the matrix normal density.

We should set $\alpha$ to 1.0 because the parameter is not identified unless we constrain $\text{trace}(C) = 1$. Otherwise, we can multiply $\alpha$ by a scalar $d$ and $C$ by $1/d$ and our likelihood will not change.

We can generate a random variable $f$ from a matrix normal density in $\mathbb{R}^{N \times M}$ using the following algorithm:

$$\eta_{i,j} \sim \text{normal}(0, 1) \; \forall i, j$$

$$f = L_{K(x \mid 1.0, \rho)} \eta L_{C(\phi)}^T$$

$$f \sim \text{matrixnormal}(0, K(x \mid 1.0, \rho), C(\phi))$$

$$\eta \in \mathbb{R}^{N \times M}$$

$$L_{C(\phi)} = \text{cholesky_decompose}(C(\phi))$$

$$L_{K(x \mid 1.0, \rho)} = \text{cholesky_decompose}(K(x \mid 1.0, \rho))$$

This can be implemented in Stan using a latent-variable GP formulation. We’ve used LKJCorr for $C(\phi)$, but any positive-definite matrix will do.
data {
  int<lower=1> N;
  int<lower=1> D;
  real x[N];
  matrix[N, D] y;
}
transformed data {
  real delta = 1e-9;
}
parameters {
  real<lower=0> rho;
  vector<lower=0>[D] alpha;
  real<lower=0> sigma;
  cholesky_factor_corr[D] L_Omega;
  matrix[N, D] eta;
}
model {
  matrix[N, D] f;
  {
    matrix[N, N] K = cov_exp_quad(x, 1.0, rho);
    matrix[N, N] L_K;

    // diagonal elements
    for (n in 1:N)
      K[n, n] = K[n, n] + delta;

    L_K = cholesky_decompose(K);
    f = L_K * eta
      * diag_pre_multiply(alpha, L_Omega)';
  }

  rho ~ inv_gamma(5, 5);
  alpha ~ std_normal();
  sigma ~ std_normal();
  L_Omega ~ lkj_corr_cholesky(3);
  to_vector(eta) ~ std_normal();

  to_vector(y) ~ normal(to_vector(f), sigma);
}
generated quantities {
    matrix[D, D] Omega;
    Omega = L_Omega * L_Omega';
}
11. Directions, Rotations, and Hyperspheres

Directional statistics involve data and/or parameters that are constrained to be directions. The set of directions forms a sphere, the geometry of which is not smoothly mappable to that of a Euclidean space because you can move around a sphere and come back to where you started. This is why it is impossible to make a map of the globe on a flat piece of paper where all points that are close to each other on the globe are close to each other on the flat map. The fundamental problem is easy to visualize in two dimensions, because as you move around a circle, you wind up back where you started. In other words, 0 degrees and 360 degrees (equivalently, 0 and \(2\pi\) radians) pick out the same point, and the distance between 359 degrees and 2 degrees is the same as the distance between 137 and 140 degrees.

Stan supports directional statistics by providing a unit-vector data type, the values of which determine points on a hypersphere (circle in two dimensions, sphere in three dimensions).

11.1. Unit vectors

The length of a vector \(x \in \mathbb{R}^K\) is given by

\[
\|x\| = \sqrt{x^\top x} = \sqrt{x_1^2 + x_2^2 + \cdots + x_K^2}.
\]

Unit vectors are defined to be vectors of unit length (i.e., length one).

With a variable declaration such as

\[
\text{unit\_vector}[K] \ x;
\]

the value of \(x\) will be constrained to be a vector of size \(K\) with unit length; the reference manual chapter on constrained parameter transforms provides precise definitions.

Warning: An extra term gets added to the log density to ensure the distribution on unit vectors is proper. This is not a problem in practice, but it may lead to misunderstandings of the target log density output (\(\text{lp\_}...\)) in some interfaces. The underlying source of the problem is that a unit vector of size \(K\) has only \(K - 1\) degrees of freedom. But there is no way to map those \(K - 1\) degrees of freedom continuously to \(\mathbb{R}^N\)—for example, the circle can’t be mapped continuously to a line so the limits work out, nor can a sphere be mapped to a plane. A workaround is needed instead.
Stan’s unit vector transform uses $K$ unconstrained variables, then projects down to the unit hypersphere. Even though the hypersphere is compact, the result would be an improper distribution. To ensure the unit vector distribution is proper, each unconstrained variable is given a “Jacobian” adjustment equal to an independent standard normal distribution. Effectively, each dimension is drawn standard normal, then they are together projected down to the hypersphere to produce a unit vector. The result is a proper uniform distribution over the hypersphere.

11.2. Circles, spheres, and hyperspheres

An $n$-sphere, written $S^n$, is defined as the set of $(n + 1)$-dimensional unit vectors, 

$$S^n = \{ x \in \mathbb{R}^{n+1} : \|x\| = 1 \}.$$ 

Even though $S^n$ is made up of points in $(n+1)$ dimensions, it is only an $n$-dimensional manifold. For example, $S^2$ is defined as a set of points in $\mathbb{R}^3$, but each such point may be described uniquely by a latitude and longitude. Geometrically, the surface defined by $S^2$ in $\mathbb{R}^3$ behaves locally like a plane, i.e., $\mathbb{R}^2$. However, the overall shape of $S^2$ is not like a plane in that it is compact (i.e., there is a maximum distance between points). If you set off around the globe in a “straight line” (i.e., a geodesic), you wind up back where you started eventually; that is why the geodesics on the sphere ($S^2$) are called “great circles,” and why we need to use some clever representations to do circular or spherical statistics.

Even though $S^{n-1}$ behaves locally like $\mathbb{R}^{n-1}$, there is no way to smoothly map between them. For example, because latitude and longitude work on a modular basis (wrapping at $2\pi$ radians in natural units), they do not produce a smooth map.

Like a bounded interval $(a, b)$, in geometric terms, a sphere is compact in that the distance between any two points is bounded.

11.3. Transforming to unconstrained parameters

Stan (inverse) transforms arbitrary points in $\mathbb{R}^{K+1}$ to points in $S^K$ using the auxiliary variable approach of Marsaglia (1972). A point $y \in \mathbb{R}^K$ is transformed to a point $x \in S^{K-1}$ by 

$$x = \frac{y}{\sqrt{y^\top y}}.$$ 

The problem with this mapping is that it’s many to one; any point lying on a vector out of the origin is projected to the same point on the surface of the sphere. Marsaglia (1972) introduced an auxiliary variable interpretation of this mapping that provides the desired properties of uniformity; the reference manual contains the precise definitions used in the chapter on constrained parameter transforms.
Warning: undefined at zero!

The above mapping from $\mathbb{R}^n$ to $S^n$ is not defined at zero. While this point outcome has measure zero during sampling, and may thus be ignored, it is the default initialization point and thus unit vector parameters cannot be initialized at zero. A simple workaround is to initialize from a small interval around zero, which is an option built into all of the Stan interfaces.

### 11.4. Unit vectors and rotations

Unit vectors correspond directly to angles and thus to rotations. This is easy to see in two dimensions, where a point on a circle determines a compass direction, or equivalently, an angle $\theta$. Given an angle $\theta$, a matrix can be defined, the pre-multiplication by which rotates a point by an angle of $\theta$. For angle $\theta$ (in two dimensions), the $2 \times 2$ rotation matrix is defined by

$$R_\theta = \begin{bmatrix} \cos \theta & -\sin \theta \\ \sin \theta & \cos \theta \end{bmatrix}.$$ 

Given a two-dimensional vector $x$, $R_\theta x$ is the rotation of $x$ (around the origin) by $\theta$ degrees.

**Angles from unit vectors**

Angles can be calculated from unit vectors. For example, a random variable `theta` representing an angle in $(-\pi, \pi)$ radians can be declared as a two-dimensional unit vector then transformed to an angle.

```stan
parameters {
    unit_vector[2] xy;
}
transformed parameters {
    real<lower = -pi(), upper = pi()> theta = atan2(xy[2], xy[1]);
}
```

If the distribution of $(x, y)$ is uniform over a circle, then the distribution of $\arctan \frac{y}{x}$ is uniform over $(-\pi, \pi)$.

It might be tempting to try to just declare `theta` directly as a parameter with the lower and upper bound constraint as given above. The drawback to this approach is that the values $-\pi$ and $\pi$ are at $-\infty$ and $\infty$ on the unconstrained scale, which can produce multimodal posterior distributions when the true distribution on the circle is unimodal.

With a little additional work on the trigonometric front, the same conversion back to angles may be accomplished in more dimensions.
11.5. **Circular representations of days and years**

A 24-hour clock naturally represents the progression of time through the day, moving from midnight to noon and back again in one rotation. A point on a circle divided into 24 hours is thus a natural representation for the time of day. Similarly, years cycle through the seasons and return to the season from which they started.

In human affairs, temporal effects often arise by convention. These can be modeled directly with ad-hoc predictors for holidays and weekends, or with data normalization back to natural scales for daylight savings time.
12. Solving Algebraic Equations

Stan provides a built-in mechanism for specifying systems of algebraic equations. These systems can be solved either with the Newton method, as implemented in the Kinsol package (Hindmarsh et al. 2005), or with the Powell hybrid method (Powell 1970). The function signatures for Stan’s algebraic solvers are fully described in the algebraic solver section of the reference manual.

Solving any system of algebraic equations can be translated into a root-finding problem, that is, given a function $f$, we wish to find $y$ such that $f(y) = 0$.

12.1. Example: system of nonlinear algebraic equations

For systems of linear algebraic equations, we recommend solving the system using matrix division. The algebraic solver becomes handy when we want to solve nonlinear equations.

As an illustrative example, we consider the following nonlinear system of two equations with two unknowns:

$$z_1 = y_1 - \theta_1$$
$$z_2 = y_1 y_2 + \theta_2$$

Our goal is to simultaneously solve all equations for $y_1$ and $y_2$, such that the vector $z$ goes to 0.

12.2. Coding an algebraic system

A system of algebraic equations is coded directly in Stan as a function with a strictly specified signature. For example, the nonlinear system given above can be coded using the following function in Stan (see the user-defined functions section for more information on coding user-defined functions).

```stan
vector system(vector y, // unknowns
              vector theta, // parameters
              data real[] x_r, // data (real)
              int[] x_i) { // data (integer)
  vector[2] z;
  z[1] = y[1] - theta[1];
}
```
The function takes the unknowns we wish to solve for in \( y \) (a vector), the system parameters in \( \theta \) (a vector), the real data in \( x_r \) (a real array) and the integer data in \( x_i \) (an integer array). The system function returns the value of the function (a vector), for which we want to compute the roots. Our example does not use real or integer data. Nevertheless, these unused arguments must be included in the system function with exactly the signature above.

The body of the system function here could also be coded using a row vector constructor and transposition,

\[
\text{return } [ \ y[1] - \theta[1], \\
\quad y[1] \times y[2] - \theta[2] \ ];
\]

As systems get more complicated, naming the intermediate expressions goes a long way toward readability.

**Strict signature**

The function defining the system must have exactly these argument types and return type. This may require passing in zero-length arrays for data or a zero-length vector for parameters if the system does not involve data or parameters.

### 12.3. Calling the algebraic solver

Let’s suppose \( \theta = (3, 6) \). To call the algebraic solver, we need to provide an initial guess. This varies on a case-by-case basis, but in general a good guess will speed up the solver and, in pathological cases, even determine whether the solver converges or not. If the solver does not converge, the metropolis proposal gets rejected and a warning message, stating no acceptable solution was found, is issued.

The solver has three tuning parameters to determine convergence: the relative tolerance, the function tolerance, and the maximum number of steps. Their behavior is explained in the section about algebraic solvers with control parameters.

The following code returns the solution to our nonlinear algebraic system:

```cpp
transformed data {
  vector[2] y_guess = {1, 1};
  real x_r[0];
  int x_i[0];
}
```
transferred parameters {
    vector[2] theta = {3, 6};
    vector[2] y;

    y = algebra_solver_newton(system, y_guess, theta, x_r, x_i);
}

which returns $y = (3, -2)$.

**Data versus parameters**
The arguments for the real data $x_r$ and the integer data $x_i$ must be expressions that only involve data or transformed data variables. $\theta$, on the other hand, must only involve parameters. Note there are no restrictions on the initial guess, $y_{guess}$, which may be a data or a parameter vector.

**Length of the algebraic function and of the vector of unknowns**
The Jacobian of the solution with respect to the parameters is computed using the implicit function theorem, which imposes certain restrictions. In particular, the Jacobian of the algebraic function $f$ with respect to the unknowns $x$ must be invertible. This requires the Jacobian to be square, meaning $f(y)$ and $y$ have the same length or, in other words the number of equations in the system is the same as the number of unknowns.

**Pathological solutions**
Certain systems may be degenerate, meaning they have multiple solutions. The algebraic solver will not report these cases, as the algorithm stops once it has found an acceptable solution. The initial guess will often determine which solution gets found first. The degeneracy may be broken by putting additional constraints on the solution. For instance, it might make “physical sense” for a solution to be positive or negative.

On the other hand, a system may not have a solution (for a given point in the parameter space). In that case, the solver will not converge to a solution. When the solver fails to do so, the current metropolis proposal gets rejected.

**12.4. Control parameters for the algebraic solver**
The call to the algebraic solver shown previously uses the default control settings. The solver allows three additional parameters, all of which must be supplied if any of them is supplied.

$y = \text{algebra_solver_newton}(\text{system, } y_{\text{guess, theta, x_r, x_i,}})$

$\text{rel_tol, f_tol, max_steps}$;
The three control arguments are relative tolerance, function tolerance, and maximum number of steps. Both tolerances need to be satisfied. If one of them is not met, the metropolis proposal gets rejected with a warning message explaining which criterion was not satisfied. The default values for the control arguments are respectively $\text{rel\_tol} = 1e-10$ ($10^{-10}$), $\text{f\_tol} = 1e-6$ ($10^{-6}$), and $\text{max\_steps} = 1e3$ ($10^3$).

**Tolerance**

The relative and function tolerances control the accuracy of the solution generated by the solver. Relative tolerances are relative to the solution value. The function tolerance is the norm of the algebraic function, once we plug in the proposed solution. This norm should go to 0 (equivalently, all elements of the vector function are 0). It helps to think about this geometrically. Ideally the output of the algebraic function is at the origin; the norm measures deviations from this ideal. As the length of the return vector increases, a certain function tolerance becomes an increasingly difficult criterion to meet, given each individual element of the vector contribute to the norm. Smaller relative tolerances produce more accurate solutions but require more computational time.

**Sensitivity analysis**

The tolerances should be set low enough that setting them lower does not change the statistical properties of posterior samples generated by the Stan program.

**Maximum number of steps**

The maximum number of steps can be used to stop a runaway simulation. This can arise in MCMC when a bad jump is taken, particularly during warmup. If the limit is hit, the current metropolis proposal gets rejected. Users will see a warning message stating the maximum number of steps has been exceeded.
13. Ordinary Differential Equations

Stan provides a number of different methods for solving systems of ordinary differential equations (ODEs). All of these methods adaptively refine their solutions in order to satisfy given tolerances, but internally they handle calculations quite a bit differently.

Because Stan’s algorithms requires gradients of the log density, the ODE solvers must not only provide the solution to the ODE itself, but also the gradient of the ODE solution with respect to parameters (the sensitivities). Two fundamentally different approaches are available in Stan to solve this problem, each having very different computational cost depending on the number of ODE states $N$ and the number of parameters $M$ being used:

- A forward sensitivity solver expands the base ODE system with additional ODE equations for the gradients of the solution. For each parameter, an additional full set of $N$ sensitivity states are added meaning that the full ODE solved has

$$N + N \cdot M$$

states.

- An adjoint sensitivity solver starts by solving the base ODE system forward in time to get the ODE solution and then solves another ODE system (the adjoint) backward in time to get the gradients. The forward and reverse solves both have $N$ states each. There is additionally one quadrature problem solved for every parameter.

The adjoint sensitivity approach scales much better than the forward sensitivity approach. Whereas the computational cost of the forward approach scales multiplicatively in the number of ODE states $N$ and parameters $M$, the adjoint sensitivity approach scales linear in states $N$ and parameters $M$. However, the adjoint problem is harder to configure and the overhead for small problems actually makes it slower than solving the full forward sensitivity system. With that in mind, the rest of this introduction focuses on the forward sensitivity interfaces. For information on the adjoint sensitivity interface see the Adjoint ODE solver.

Two interfaces are provided for each forward sensitivity solver: one with default tolerances and default max number of steps, and one that allows these controls to
be modified. Choosing tolerances is important for making any of the solvers work well – the defaults will not work everywhere. The tolerances should be chosen primarily with consideration to the scales of the solutions, the accuracy needed for the solutions, and how the solutions are used in the model. For instance, if a solution component slowly varies between 3.0 and 5.0 and measurements of the ODE state are noisy, then perhaps the tolerances do not need to be as tight as for a situation where the solutions vary between 3.0 and 3.1 and very high precision measurements of the ODE state are available. It is also often useful to reduce the absolute tolerance when a component of the solution is expected to approach zero. For information on choosing tolerances, see the control parameters section.

The advantage of adaptive solvers is that as long as reasonable tolerances are provided and an ODE solver well-suited to the problem is chosen the technical details of solving the ODE can be abstracted away. The catch is that it is not always clear from the outset what reasonable tolerances are or which ODE solver is best suited to a problem. In addition, as changes are made to an ODE model, the optimal solver and tolerances may change.

With this in mind, the four forward solvers are rk45, bdf, adams, and ckrk. If no other information about the ODE is available, start with the rk45 solver. The list below has information on when each solver is useful.

If there is any uncertainty about which solver is the best, it can be useful to measure the performance of all the interesting solvers using profile statements. It is difficult to always know exactly what solver is the best in all situations, but a profile can provide a quick check.

- **rk45**: a fourth and fifth order Runge-Kutta method for non-stiff systems (Dormand and Prince 1980; Ahnert and Mulansky 2011). rk45 is the most generic solver and should be tried first.

- **bdf**: a variable-step, variable-order, backward-differentiation formula implementation for stiff systems (Cohen and Hindmarsh 1996; Serban and Hindmarsh 2005). bdf is often useful for ODEs modeling chemical reactions.

- **adams**: a variable-step, variable-order, Adams-Moulton formula implementation for non-stiff systems (Cohen and Hindmarsh 1996; Serban and Hindmarsh 2005). The method has order up to 12, hence is commonly used when high-accuracy is desired for a very smooth solution, such as in modeling celestial mechanics and orbital dynamics (Montenbruck and Gill 2000).

- **ckrk**: a fourth and fifth order explicit Runge-Kutta method for non-stiff and semi-stiff systems (Cash and Karp 1990; Mazzia, Cash, and Soetaert 2012).
The difference between \texttt{ckrk} and \texttt{rk45} is that \texttt{ckrk} should perform better for systems that exhibit rapidly varying solutions. Often in those situations the derivatives become large or even nearly discontinuous, and \texttt{ckrk} is designed to address such problems.

For a discussion of stiff ODE systems, see the stiff ODE section. For information on the adjoint sensitivity interface see the Adjoint ODE solver section. The function signatures for Stan’s ODE solvers can be found in the function reference manual section on ODE solvers.

\section*{13.1. Notation}

An ODE is defined by a set of differential equations, \( y(t, \theta)' = f(t, y, \theta) \), and initial conditions, \( y(t_0, \theta) = y_0 \). The function \( f(t, y, \theta) \) is called the system function. The \( \theta \) dependence is included in the notation for \( y(t, \theta) \) and \( f(t, y, \theta) \) as a reminder that the solution is a function of any parameters used in the computation.

\section*{13.2. Example: simple harmonic oscillator}

As an example of a system of ODEs, consider a harmonic oscillator. In a harmonic oscillator a particle disturbed from equilibrium is pulled back towards its equilibrium position by a force proportional to its displacement from equilibrium. The system here additionally has a friction force proportional to particle speed which points in the opposite direction of the particle velocity. The system state will be a pair \( y = (y_1, y_2) \) representing position and speed. The change in the system with respect to time is given by the following differential equations.\footnote{This example is drawn from the documentation for the Boost Numeric Odeint library (Ahnert and Mulansky 2011), which Stan uses to implement the \texttt{rk45} and \texttt{ckrk} solver.}

\[
\frac{d}{dt} y_1 = y_2 \\
\frac{d}{dt} y_2 = -y_1 - \theta y_2
\]

The state equations implicitly defines the state at future times as a function of an initial state and the system parameters.

\section*{13.3. Coding the ODE system function}

The first step in coding an ODE system in Stan is defining the ODE system function. The system functions require a specific signature so that the solvers know how to use them properly.

The first argument to the system function is time, passed as a \texttt{real}; the second argument to the system function is the system state, passed as a \texttt{vector}, and the return value from the system function are the current time derivatives of the state.
defined as a vector. Additional arguments can be included in the system function to pass other information into the solve (these will be passed through the function that starts the ODE integration). These argument can be parameters (in this case, the friction coefficient), data, or any quantities that are needed to define the differential equation.

The simple harmonic oscillator can be coded using the following function in Stan (see the user-defined functions chapter for more information on coding user-defined functions).

```stan
vector sho(real t, // time
            vector y, // state
            real theta) { // friction parameter
  vector[2] dydt;
  dydt[1] = y[2];
  return dydt;
}
```

The function takes in a time \( t \) (a real), the system state \( y \) (a vector), and the parameter \( \theta \) (a real). The function returns a vector of time derivatives of the system state at time \( t \), state \( y \), and parameter \( \theta \). The simple harmonic oscillator coded here does not have time-sensitive equations; that is, \( t \) does not show up in the definition of \( dy/dt \), however it is still required.

**Strict signature**
The types in the ODE system function are strict. The first argument is the time passed as a real, the second argument is the state passed as a vector, and the return type is a vector. A model that does not have this signature will fail to compile. The third argument onwards can be any type, granted all the argument types match the types of the respective arguments in the solver call.

All of these are possible ODE signatures:

- `vector myode1(real t, vector y, real a0);`
- `vector myode2(real t, vector y, int[] a0, vector a1);`
- `vector myode3(real t, vector y, matrix a0, real[] a1, row_vector a2);`

but these are not allowed:

- `vector myode1(real t, real[] y, real a0); // Second argument is not a vector`
- `real[] myode2(real t, vector y, real a0); // Return type is not a vector`
- `vector myode3(vector y, real a0); // First argument is not a real and second`
13.4. Measurement error models
Noisy observations of the ODE state can be used to estimate the parameters and/or the initial state of the system.

Simulating noisy measurements
As an example, suppose the simple harmonic oscillator has a parameter value of \( \theta = 0.15 \) and an initial state \( y(t = 0, \theta = 0.15) = (1, 0) \). Assume the system is measured at 10 time points, \( t = 1, 2, \cdots, 10 \), where each measurement of \( y(t, \theta) \) has independent normal(0, 0.1) error in both dimensions \( (y_1(t, \theta) \) and \( y_2(t, \theta) \)).

The following model can be used to generate data like this:

```c
functions {
  vector sho(real t, vector y, real theta) {
    vector[2] dydt;
    dydt[1] = y[2];
    return dydt;
  }
}
data {
  int<lower=1> T;
  vector[2] y0;
  real t0;
  real ts[T];
  real theta;
}
model {
}
generated quantities {
  vector[2] y_sim[T] = ode_rk45(sho, y0, t0, ts, theta);
  // add measurement error
  for (t in 1:T) {
    y_sim[t, 1] += normal_rng(0, 0.1);
    y_sim[t, 2] += normal_rng(0, 0.1);
  }
}
```

The system parameters \( \theta \) and initial state \( y_0 \) are read in as data along with the
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initial time \( t_0 \) and observation times \( t_s \). The ODE is solved for the specified times, and then random measurement errors are added to produce simulated observations \( y_{\text{sim}} \). Because the system is not stiff, the \texttt{ode_rk45} solver is used.

This program illustrates the way in which the ODE solver is called in a Stan program,

```
vector[2] y_{\text{sim}}[T] = ode_rk45(sho, y_0, t_0, t_s, theta);
```

this returns the solution of the ODE initial value problem defined by system function \( \text{sho} \), initial state \( y_0 \), initial time \( t_0 \), and parameter \( \text{theta} \) at the times \( t_s \). The call explicitly specifies the non-stiff RK45 solver.

The parameter \( \text{theta} \) is passed unmodified to the ODE system function. If there were additional arguments that must be passed, they could be appended to the end of the \texttt{ode} call here. For instance, if the system function took two parameters, \( \theta \) and \( \beta \), the system function definition would look like:

```
vector sho(real t, vector y, real theta, real beta) { ... }
```

and the appropriate ODE solver call would be:

```
ode_rk45(sho, y_0, t_0, t_s, theta, beta);
```

Any number of additional arguments can be added. They can be any Stan type (as long as the types match between the ODE system function and the solver call).

Because all none of the input arguments are a function of parameters, the ODE solver is called in the generated quantities block. The random measurement noise is added to each of the \( T \) outputs with \texttt{normal_rng}.

**Estimating system parameters and initial state**

These ten noisy observations of the state can be used to estimate the friction parameter, \( \theta \), the initial conditions, \( y(t_0, \theta) \), and the scale of the noise in the problem. The full Stan model is:
Figure 13.1: Typical realization of harmonic oscillator trajectory.

functions {
  vector sho(real t,
              vector y,
              real theta) {
    vector[2] dydt;
    dydt[1] = y[2];
    return dydt;
  }
}
data {
  int<lower=1> T;
  vector[2] y[T];
  real t0;
  real ts[T];
}
parameters {
    vector[2] y0;
    vector<lower=0>[2] sigma;
    real theta;
}

model {
    vector[2] mu[T] = ode_rk45(sho, y0, t0, ts, theta);
    sigma ~ normal(0, 2.5);
    theta ~ std_normal();
    y0 ~ std_normal();
    for (t in 1:T)
        y[t] ~ normal(mu[t], sigma);
}

Because the solves are now a function of model parameters, the ode_rk45 call is now made in the model block. There are half-normal priors on the measurement error scales sigma, and standard normal priors on theta and the initial state vector y0. The solutions to the ODE are assigned to mu, which is used as the location for the normal observation model.

As with other regression models, it’s easy to change the noise model to something with heavier tails (e.g., Student-t distributed), correlation in the state variables (e.g., with a multivariate normal distribution), or both heavy tails and correlation in the state variables (e.g., with a multivariate Student-t distribution).

13.5. Stiff ODEs

Stiffness is a numerical phenomena that causes some differential equation solvers difficulty, notably the Runge-Kutta RK45 solver used in the examples earlier. The phenomena is common in chemical reaction systems, which are often characterized by having multiple vastly different time-scales. The stiffness of a system can also vary between different parts of parameter space, and so a typically non-stiff system may exhibit stiffness occasionally. These sorts of difficulties can occur more frequently with loose priors or during warmup.

Stan provides a specialized solver for stiff ODEs (Cohen and Hindmarsh 1996; Serban and Hindmarsh 2005). An ODE system is specified exactly the same way with a function of exactly the same signature. The only difference is in the call to the solver the rk45 suffix is replaced with bdf, as in

ode_bdf(sho, y0, t0, ts, theta);
Using the stiff (bdf) solver on a system that is not stiff may be much slower than using the non-stiff (rk45) solver because each step of the stiff solver takes more time to compute. On the other hand, attempting to use the non-stiff solver for a stiff system will cause the timestep to become very small, leading the non-stiff solver taking more time overall even if each step is easier to compute than for the stiff solver.

If it is not known for sure that an ODE system is stiff, run the model with both the rk45 and bdf solvers and see which is faster. If the rk45 solver is faster, then the problem is probably non-stiff, and then it makes sense to try the adams solver as well. The adams solver uses higher order methods which can take larger timesteps than the rk45 solver, though similar to the bdf solver each of these steps is more expensive to compute.

13.6. Control parameters for ODE solving
For additional control of the solves, both the stiff and non-stiff forward ODE solvers have function signatures that makes it possible to specify the relative_tolerance, absolute_tolerance, and max_num_steps parameters. These are the same as the regular function names but with _tol appended to the end. All three control arguments must be supplied with this signature (there are no defaults).

```plaintext
vector[2] y_sim[T] = ode_bdf_tol(sho, y0, t0, ts,
relative_tolerance,
absolute_tolerance,
max_num_steps,
theta);
```

relative_tolerance and absolute_tolerance control accuracy the solver tries to achieve, and max_num_steps specifies the maximum number of steps the solver will take between output time points before throwing an error.

The control parameters must be data variables – they cannot be parameters or expressions that depend on parameters, including local variables in any block other than transformed data and generated quantities. User-defined function arguments may be qualified as only allowing data arguments using the data qualifier.

For the RK45 solver, the default values for relative and absolute tolerance are both $10^{-6}$ and the maximum number of steps between outputs is one million. For the BDF and Adams solvers, the relative and absolute tolerances are $10^{-10}$ and the maximum number of steps between outputs is one hundred million.
Discontinuous ODE system function

If there are discontinuities in the ODE system function, it is best to integrate the ODE between the discontinuities, stopping the solver at each one, and restarting it on the other side.

Nonetheless, the ODE solvers will attempt to integrate over discontinuities they encounters in the state function. The accuracy of the solution near the discontinuity may be problematic (requiring many small steps). An example of such a discontinuity is a lag in a pharmacokinetic model, where a concentration is zero for times $0 < t < t'$ and then positive for $t \geq t'$. In this example example, we would use code in the system such as

```plaintext
if (t < t_lag)
    return [0, 0];
else
    ... return non-zero vector...;
```

In general it is better to integrate up to $t_{lag}$ in one solve and then integrate from $t_{lag}$ onwards in another. Mathematically, the discontinuity can make the problem ill-defined and the numerical integrator may behave erratically around it.

If the location of the discontinuity cannot be controlled precisely, or there is some other rapidly change in ODE behavior, it can be useful to tell the ODE solver to produce output in the neighborhood. This can help the ODE solver avoid indiscriminately stepping over an important feature of the solution.

Tolerance

The relative tolerance RTOL and absolute tolerance ATOL control the accuracy of the numerical solution. Specifically, when solving an ODE with unknowns $y = (y_1, \ldots, y_n)^T$, at every step the solver controls estimated local error $e = (e_1, \ldots, e_n)^T$ through its weighted root-mean-square norm (Serban and Hindmarsh (2005), Hairer, Nørsett, and Wanner (1993))

$$
\sqrt{\sum_{i=1}^{n} \frac{1}{n (RTOL \times y_i + ATOL)^2} e_i^2} < 1
$$

by reducing the stepsize when the inequality is not satisfied.

To understand the roles of the two tolerances it helps to assume $y$ at opposite scales in the above expression: on one hand the absolute tolerance has little effect when $y_i \gg 1$, on the other the relative tolerance can not affect the norm when $y_i = 0$. Users are strongly encouraged to carefully choose tolerance values according
to the ODE and its application. One can follow Brenan, Campbell, and Petzold (1995) for a rule of thumb: let \( m \) be the number of significant digits required for \( y \), set \( RTOL = 10^{-(m+1)} \), and set \( ATOL \) at which \( y \) becomes insignificant. Note that the same weighted root-mean-square norm is used to control nonlinear solver convergence in \texttt{bdf} and \texttt{adams} solvers, and the same tolerances are used to control forward sensitivity calculation. See Serban and Hindmarsh (2005) for details.

**Maximum number of steps**

The maximum number of steps can be used to stop a runaway simulation. This can arise in when MCMC moves to a part of parameter space very far from where a differential equation would typically be solved. In particular this can happen during warmup. With the non-stiff solver, this may happen when the sampler moves to stiff regions of parameter space, which will requires small step sizes.

### 13.7. Adjoint ODE solver

The adjoint ODE solver method differs mathematically from the forward ODE solvers in the way gradients of the ODE solution are obtained. The forward ODE approach augments the original ODE system with \( N \) additional states for each parameter for which gradients are needed. If there are \( M \) parameters for which sensitivities are required, then the augmented ODE system has a total of \( N \cdot (M + 1) \) states. This can result in very large ODE systems through the multiplicative scaling of the computational effort needed.

In contrast, the adjoint ODE solver integrates forward in time a system of \( N \) equations to compute the ODE solution and then integrates backwards in time another system of \( N \) equations to get the sensitivities. Additionally, for \( M \) parameters there are \( M \) additional equations to integrate during the backwards solve. Because of this the adjoint sensitivity problem scales better in parameters than the forward sensitivity problem. The adjoint solver in Stan uses CVODES (the same as the \texttt{bdf} and \texttt{adams} forward sensitivity interfaces).

The solution computed in the forward integration is required during the backward integration. CVODES uses a checkpointing scheme that saves the forward solver state regularly. The number of steps between saving checkpoints is configurable in the interface. These checkpoints are then interpolated during the backward solve using one of two interpolation schemes.

The solver type (either \texttt{bdf} or \texttt{adams}) can be individually set for both the forward and backward solves.

The tolerances for each phase of the solve must be specified in the interface. Note that the absolute tolerance for the forward and backward ODE integration phase
need to be set for each ODE state separately. The harmonic oscillator example call from above becomes:

```cpp
vector[2] y_sim[T]
  = ode_adjoint_tol_ctl(sho, y0, t0, ts,
    relative_tolerance/9.0, # forward tolerance
    rep_vector(absolute_tolerance/9.0, 2), # forward tol.
    relative_tolerance/3.0,  # backward tolerance
    rep_vector(absolute_tolerance/3.0, 2), # backward tol.
    relative_tolerance,      # quadrature tolerance
    absolute_tolerance,      # quadrature tolerance
    max_num_steps,           # number of steps between checkpoints
    150,
    1, # interpolation polynomial: 1=Hermite, 2=polynomial
    2, # solver for forward phase: 1=Adams, 2=BDF
    2, # solver for backward phase: 1=Adams, 2=BDF
    theta);
```

For a detailed information on each argument please see the Stan function reference manual.

### 13.8. Solving a system of linear ODEs using a matrix exponential

Linear systems of ODEs can be solved using a matrix exponential. This can be considerably faster than using one of the ODE solvers.

The solution to \( \frac{d}{dt} y = Ay \) is \( y = y_0 e^{at} \), where the constant \( y_0 \) is determined by boundary conditions. We can extend this solution to the vector case:

\[
\frac{d}{dt} y = Ay
\]

where \( y \) is now a vector of length \( n \) and \( A \) is an \( n \) by \( n \) matrix. The solution is then given by:

\[
y = e^{tA} y_0
\]

where the matrix exponential is formally defined by the convergent power series:

\[
e^{tA} = \sum_{n=0}^{\infty} \frac{t^n A^n}{n!} = I + tA + \frac{t^2 A^2}{2!} + \cdots
\]

We can apply this technique to the simple harmonic oscillator example, by setting

\[
y = \begin{bmatrix} y_1 \\ y_2 \end{bmatrix} \quad A = \begin{bmatrix} 0 & 1 \\ -1 & -\theta \end{bmatrix}
\]
The Stan model to simulate noisy observations using a matrix exponential function is given below.

In general, computing a matrix exponential will be more efficient than using a numerical solver. We can however only apply this technique to systems of linear ODEs.

data {
  int<lower=1> T;
  vector[2] y0;
  real ts[T];
  real theta[1];
}

model {
}

generated quantities {
  vector[2] y_sim[T];
  matrix[2, 2] A = [[ 0, 1],
                   [-1, -theta[1]]]
  for (t in 1:T)
    y_sim[t] = matrix_exp((t - 1) * A) * y0;
  // add measurement error
  for (t in 1:T) {
    y_sim[t, 1] += normal_rng(0, 0.1);
    y_sim[t, 2] += normal_rng(0, 0.1);
  }
}

This Stan program simulates noisy measurements from a simple harmonic oscillator. The system of linear differential equations is coded as a matrix. The system parameters theta and initial state y0 are read in as data along observation times ts. The generated quantities block is used to solve the ODE for the specified times and then add random measurement error, producing observations y_sim. Because the ODEs are linear, we can use the matrix_exp function to solve the system.
14. Computing One Dimensional Integrals

Definite and indefinite one dimensional integrals can be performed in Stan using the integrate_1d function.

As an example, the normalizing constant of a left-truncated normal distribution is

\[
\int_{a}^{\infty} \frac{1}{\sqrt{2\pi \sigma^2}} e^{-\frac{1}{2} \left( \frac{x-\mu}{\sigma} \right)^2} \, dx.
\]

To compute this integral in Stan, the integrand must first be defined as a Stan function (see the Stan Reference Manual chapter on User-Defined Functions for more information on coding user-defined functions).

```stan
real normal_density(real x,            // Function argument
                     real xc,           // Complement of function argument
                     // on the domain (defined later)
                     real[] theta,      // parameters
                     real[] x_r,        // data (real)
                     int[] x_i) {       // data (integer)
  real mu = theta[1];
  real sigma = theta[2];

  return 1 / (sqrt(2 * pi()) * sigma) * exp(-0.5 * ((x - mu) / sigma)^2);
}
```

This function is expected to return the value of the integrand evaluated at point \( x \). The argument \( xc \) is used in definite integrals to avoid loss of precision near the limits of integration and is set to NaN when either limit is infinite (see the section on precision/loss in the chapter on Higher-Order Functions of the Stan Functions Reference for details on how to use this). The argument \( theta \) is used to pass in arguments of the integral that are a function of the parameters in our model. The arguments \( x_r \) and \( x_i \) are used to pass in real and integer arguments of the integral that are not a function of our parameters.

The function defining the integrand must have exactly the argument types and return type of `normal_density` above, though argument naming is not important. Even if \( x_r \) and \( x_i \) are unused in the integrand, they must be included in the function.
signature. This may require passing in zero-length arrays for data or a zero-length vector for parameters if the integral does not involve data or parameters.

14.1. Calling the integrator
Suppose that our model requires evaluating the lpdf of a left-truncated normal, but the truncation limit is to be estimated as a parameter. Because the truncation point is a parameter, we must include the normalization term of the truncated pdf when computing our model's log density. Note this is just an example of how to use the 1D integrator. The more efficient way to perform the correct normalization in Stan is described in the chapter on Truncated or Censored Data of this guide.

Such a model might look like (include the function defined at the beginning of this chapter to make this code compile):

data {
  int N;
  real y[N];
}

transformed data {
  real x_r[0];
  int x_i[0];
}

parameters {
  real mu;
  real<lower = 0.0> sigma;
  real left_limit;
}

model {
  mu ~ normal(0, 1);
  sigma ~ normal(0, 1);
  left_limit ~ normal(0, 1);
  target += normal_lpdf(y | mu, sigma);
  target += log(integrate_1d(normal_density,
               left_limit,
               positive_infinity(),
               { mu, sigma }, x_r, x_i));
}
Limits of integration
The limits of integration can be finite or infinite. The infinite limits are made available via the Stan calls negative_infinity() and positive_infinity().

If both limits are either negative_infinity() or positive_infinity(), the integral and its gradients are set to zero.

Data vs. parameters
The arguments for the real data $x_r$ and the integer data $x_i$ must be expressions that only involve data or transformed data variables. theta, on the other hand, can be a function of data, transformed data, parameters, or transformed parameters.

The endpoints of integration can be data or parameters (and internally the derivatives of the integral with respect to the endpoints are handled with the Leibniz integral rule).

14.2. Integrator convergence
The integral is performed with the iterative 1D quadrature methods implemented in the Boost library (Agrawal et al. 2017). If the $n$th estimate of the integral is denoted $I_n$ and the $n$th estimate of the norm of the integral is denoted $|I|_n$, the iteration is terminated when

$$\frac{|I_{n+1} - I_n|}{|I|_{n+1}} < \text{relative tolerance}.$$ 

The relative_tolerance parameter can be optionally specified as the last argument to integrate_1d. By default, integrate_1d follows the Boost library recommendation of setting relative_tolerance to the square root of the machine epsilon of double precision floating point numbers (about $1e-8$).

Zero-crossing integrals
Integrals on the (possibly infinite) interval $(a, b)$ that cross zero are split into two integrals, one from $(a, 0)$ and one from $(0, b)$. This is because the quadrature methods employed internally can have difficulty near zero.

In this case, each integral is separately integrated to the given relative_tolerance.

Avoiding precision loss near limits of integration in definite integrals
If care is not taken, the quadrature can suffer from numerical loss of precision near the endpoints of definite integrals.

For instance, in integrating the pdf of a beta distribution when the values of $\alpha$ and $\beta$ are small, most of the probability mass is lumped near zero and one.
The pdf of a beta distribution is proportional to

\[ p(x) \propto x^{\alpha-1}(1 - x)^{\beta-1} \]

Normalizing this distribution requires computing the integral of \( p(x) \) from zero to one. In Stan code, the integrand might look like:

```stan
real beta(real x, real xc, real[] theta, real[] x_r, int[] x_i) {
  real alpha = theta[1];
  real beta = theta[2];

  return x^(alpha - 1.0) * (1.0 - x)^(beta - 1.0);
}
```

The issue is that there will be numerical breakdown in the precision of \( 1.0 - x \) as \( x \) gets close to one. This is because of the limited precision of double precision floating numbers. This integral will fail to converge for values of \( \alpha \) and \( \beta \) much less than one.

This is where \( xc \) is useful. It is defined, for definite integrals, as a high precision version of the distance from \( x \) to the nearest endpoint. To make use of this for the beta integral, the integrand can be re-coded:

```stan
real beta(real x, real xc, real[] theta, real[] x_r, int[] x_i) {
  real alpha = theta[1];
  real beta = theta[2];
  real v;

  if(x > 0.5) {
    v = x^(alpha - 1.0) * xc^(beta - 1.0);
  } else {
    v = x^(alpha - 1.0) * (1.0 - x)^(beta - 1.0);
  }

  return v;
}
```

This version of the integrand will converge for much smaller values of \( \alpha \) and \( \beta \) than otherwise possible.

Note, \( xc \) is only used for definite integrals. If either the left endpoint is at negative infinity or the right endpoint is at positive infinity, \( xc \) will be NaN.
For zero-crossing definite integrals (see section Zero Crossing) the integrals are broken into two pieces \((a, 0)\) and \((0, b)\) for endpoints \(a < 0\) and \(b > 0\) and \(xc\) is a high precision version of the distance to the limits of each of the two integrals separately. This means \(xc\) will be the a high precision version of \(a - x\), \(x\), or \(b - x\), depending on the value of \(x\) and the endpoints.
Part 2. Programming Techniques

This part of the manual surveys general programming techniques in Stan that are useful across a range of different model types.
15. Floating Point Arithmetic

Computers approximate real values in \( \mathbb{R} \) using a fixed number of bits. This chapter explains how this is done and why it is important for writing robust Stan (and other numerical) programs. The subfield of computer science devoted to studying how real arithmetic works on computers is called numerical analysis.

15.1. Floating-point representations

Stan’s arithmetic is implemented using double-precision arithmetic. The behavior of most\(^1\) modern computers follows the floating-point arithmetic, *IEEE Standard for Floating-Point Arithmetic* (IEEE 754).

Finite values

The double-precision component of the IEEE 754 standard specifies the representation of real values using a fixed pattern of 64 bits (8 bytes). All values are represented in base two (i.e., binary). The representation is divided into two signed components:

- **significand** (53 bits): base value representing significant digits
- **exponent** (11 bits): power of two multiplied by the base

The value of a finite floating point number is

\[
v = (-1)^s \times c 2^q
\]

Normality

A *normal* floating-point value does not use any leading zeros in its significand; *subnormal* numbers may use leading zeros. Not all I/O systems support subnormal numbers.

Ranges and extreme values

There are some reserved exponent values so that legal exponent values range between \(-(2^{10}) + 2 = -1022\) and \(2^{10} - 1 = 1023\). Legal significand values are between \(-2^{52}\) and \(2^{52} - 1\). Floating point allows the representation of both really big and really small values. Some extreme values are

- **largest normal finite number**: \(\approx 1.8 \times 10^{308}\)

\(^{1}\)The notable exception is Intel's optimizing compilers under certain optimization settings.
• largest subnormal finite number: \( \approx 2.2 \times 10^{308} \)
• smallest positive normal number: \( \approx 2.2 \times 10^{-308} \)
• smallest positive subnormal number: \( \approx 4.9 \times 10^{-324} \)

Signed zero
Because of the sign bit, there are two ways to represent zero, often called “positive zero” and “negative zero.” This distinction is irrelevant in Stan (as it is in R), because the two values are equal (i.e., \( 0 == -0 \) evaluates to true).

Not-a-number values
A specially chosen bit pattern is used for the not-a-number value (often written as NaN in programming language output, including Stan’s).

Stan provides a value function `not_a_number()` that returns this special not-a-number value. It is meant to represent error conditions, not missing values. Usually when not-a-number is an argument to a function, the result will not-a-number if an exception (a rejection in Stan) is not raised.

Stan also provides a test function `is_nan(x)` that returns 1 if \( x \) is not-a-number and 0 otherwise.

Not-a-number values propagate under almost all mathematical operations. For example, all of the built-in binary arithmetic operations (addition, subtraction, multiplication, division, negation) return not-a-number if any of their arguments are not-a-number. The built-in functions such as \( \log \) and \( \exp \) have the same behavior, propagating not-a-number values.

Most of Stan’s built-in functions will throw exceptions (i.e., reject) when any of their arguments is not-a-number.

Comparisons with not-a-number always return false, up to and including comparison with itself. That is, `not_a_number() == not_a_number()` somewhat confusingly returns false. That is why there is a built-in `is_nan()` function in Stan (and in C++). The only exception is negation, which remains coherent. This means `not_a_number() != not_a_number()` returns true.

Undefined operations often return not-a-number values. For example, \( \sqrt{-1} \) will evaluate to not-a-number.

Positive and negative infinity
There are also two special values representing positive infinity (\( \infty \)) and negative infinity (\( -\infty \)). These are not as pathological as not-a-number, but are often used to represent error conditions such as overflow and underflow. For example, rather than
raising an error or returning not-a-number, \( \log(0) \) evaluates to negative infinity. Exponentiating negative infinity leads back to zero, so that \( 0 == \exp(\log(0)) \). Nevertheless, this should not be done in Stan because the chain rule used to calculate the derivatives will attempt illegal operations and return not-a-number.

There are value functions \texttt{positive_infinity()} and \texttt{negative_infinity()} as well as a test function \texttt{is_inf()}.

Positive and negative infinity have the expected comparison behavior, so that \texttt{negative_infinity()} < 0 evaluates to true (represented with 1 in Stan). Also, negating positive infinity leads to negative infinity and vice-versa.

Positive infinity added to either itself or a finite value produces positive infinity. Negative infinity behaves the same way. However, attempts to subtract positive infinity from itself produce not-a-number, not zero. Similarly, attempts to divide infinite values results in a not-a-number value.

**15.2. Literals: decimal and scientific notation**

In programming languages such as Stan, numbers may be represented in standard decimal (base 10) notation. For example, 2.39 or -1567846.276452. Remember there is no point in writing more than 16 significant digits as they cannot be represented. A number may be coded in Stan using scientific notation, which consists of a signed decimal representation of a base and a signed integer decimal exponent. For example, \( 36.29 \times 10^{-3} \), which is the same number as is represented by \( 0.03629 \).

**15.3. Arithmetic precision**

The choice of significand provides \( \log_{10} 2^{53} \approx 15.95 \) decimal (base 10) digits of arithmetic precision. This is just the precision of the floating-point representation. After several operations are chained together, the realized arithmetic precision is often much lower.

**Rounding and probabilities**

In practice, the finite amount of arithmetic precision leads to rounding, whereby a number is represented by the closest floating-point number. For example, with only 16 decimal digits of accuracy,

\[
1 + 1e-20 == 1
\]

The closest floating point number to \( 1 + 10^{-20} \) turns out to be 1 itself. By contrast,

\[
0 + 1e-20 == 1e-20
\]

This highlights the fact that precision depends on scale. Even though \( 1 + 1e-20 == 

\]
1e-20 + 1e-20 == 2e-20, as expected.

Rounding also manifests itself in a lack of *transitivity*. In particular, it does not usually hold for three floating point numbers \(a, b, c\) that \((a + b) + c = a + (b = c)\).

In statistical applications, problems often manifest in situations where users expect the usual rules of real-valued arithmetic to hold. Suppose we have a lower triangular matrix \(L\) with strictly positive diagonal, so that it is the Cholesky factor of a positive-definite matrix \(LL^\top\). In practice, rounding and loss of precision may render the result \(LL^\top\) neither symmetric nor positive definite.

In practice, care must be taken to defend against rounding. For example, symmetry may be produced by adding \(LL^\top\) with its transpose and dividing by two, or by copying the lower triangular portion into the upper portion. Positive definiteness may be maintained by adding a small quantity to the diagonal.

**Machine precision and the asymmetry of 0 and 1**
The smallest number greater than zero is roughly \(0 + 10^{-323}\). The largest number less than zero is roughly \(1 - 10^{-15.95}\). The asymmetry is apparent when considering the representation of that largest number smaller than one—the exponent is of no help, and the number is represented as the binary equivalent of \(0.9999999999999999\).

For this reason, the *machine precision* is said to be roughly \(10^{15.95}\). This constant is available as `machine_precision()` in Stan.

**Complementary and epsilon functions**
Special operations are available to mitigate this problem with numbers rounding when they get close to one. For example, consider the operation \(\log(1 + x)\) for positive \(x\). When \(x\) is small (less than \(10^{-16}\) for double-precision floating point), the sum in the argument will round to 1 and the result will round to zero. To allow more granularity, programming languages provide a library function directly implementing \(f(x) = \log(1 + x)\). In Stan (as in C++), this operation is written as `log1p(x)`. Because \(x\) itself may be close to zero, the function `log1p(x)` can take the logarithm of values very close to one, the results of which are close to zero.

Similarly, the complementary cumulative distribution functions (CCDF), defined by \(F_Y^\complement(y) = 1 - F_Y(y)\), where \(F_Y\) is the cumulative distribution function (CDF) for the random variable \(Y\). This allows values very close to one to be represented in complementary form.

**Catastrophic cancellation**
Another downside to floating point representations is that subtraction of two numbers close to each other results in a loss of precision that depends on how close they are.
This is easy to see in practice. Consider
\[
1.23456789012345 \\
-1.23456789012344 \\
= 0.00000000000001
\]

We start with fifteen decimal places of accuracy in the arguments and are left with a single decimal place of accuracy in the result.

Catastrophic cancellation arises in statistical computations whenever we calculate variance for a distribution with small standard deviations relative to its location. When calculating summary statistics, Stan uses Welford’s algorithm for computing variances. This avoids catastrophic cancellation and may also be carried out in a single pass.

**Overflow**

Even though \(1 \times 10^{200}\) may be represented as a double precision floating point value, there is no finite value large enough to represent \(1 \times 10^{200} \times 1 \times 10^{200}\). The result of \(1 \times 10^{200} \times 1 \times 10^{200}\) is said to **overflow**. The IEEE 754 standard requires the result to be positive infinity.

Overflow is rarely a problem in statistical computations. If it is, it’s possible to work on the log scale, just as for underflow as described below.

**Underflow and the log scale**

When there is no number small enough to represent a result, it is said to **underflow**. For instance, \(1 \times 10^{-200}\) may be represented, but \(1 \times 10^{-200} \times 1 \times 10^{-200}\) underflows so that the result is zero.

Underflow is a ubiquitous problem in likelihood calculations. For example, if \(p(y_n \mid \theta) < 0.1\), then

\[
p(y \mid \theta) = \prod_{n=1}^{N} p(y_n \mid \theta)
\]

will underflow as soon as \(N > 350\) or so.

To deal with underflow, work on the log scale. Even though \(p(y \mid \theta)\) can’t be represented, there is no problem representing

\[
\log p(y \mid \theta) = \log \prod_{n=1}^{N} p(y_n \mid \theta) = \sum_{n=1}^{N} \log p(y_n \mid \theta)
\]

This is why all of Stan’s probability functions operate on the log scale.
15.4. Log sum of exponentials

Working on the log scale, multiplication is converted to addition,

\[ \log(a \cdot b) = \log a + \log b. \]

Thus sequences of multiplication operations can remain on the log scale. But what about addition? Given \( \log a \) and \( \log b \), how do we get \( \log(a + b) \)? Working out the algebra,

\[ \log(a + b) = \log(\exp(\log a) + \exp(\log b)). \]

**Log-sum-exp function**

The nested log of sum of exponentials is so common, it has its own name, “log-sum-exp,”

\[ \text{log-sum-exp}(u, v) = \log(\exp(u) + \exp(v)). \]

so that

\[ \log(a + b) = \text{log-sum-exp}(\log a, \log b). \]

Although it appears this might overflow as soon as exponentiation is introduced, evaluation does not proceed by evaluating the terms as written. Instead, with a little algebra, the terms are rearranged into a stable form,

\[ \text{log-sum-exp}(u, v) = \max(u, v) + \log \left( \exp(u - \max(u, v)) + \exp(v - \max(u, v)) \right). \]

Because the terms inside the exponentiations are \( u - \max(u, v) \) and \( v - \max(u, v) \), one will be zero and the other will be negative. Because the operation is symmetric, it may be assumed without loss of generality that \( u \geq v \), so that

\[ \text{log-sum-exp}(u, v) = u + \log \left( 1 + \exp(v - u) \right). \]

Although the inner term may itself be evaluated using the built-in function \( \log1p \), there is only limited gain because \( \exp(v - u) \) is only near zero when \( u \) is much larger than \( v \), meaning the final result is likely to round to \( u \) anyway.

To conclude, when evaluating \( \log(a + b) \) given \( \log a \) and \( \log b \), and assuming \( \log a > \log b \), return

\[ \log(a + b) = \log a + \log1p(\exp(\log b - \log a)). \]
CHAPTER 15. FLOATING POINT ARITHMETIC

Applying log-sum-exp to a sequence
The log sum of exponentials function may be generalized to sequences in the obvious way, so that if \( v = v_1, \ldots, v_N \), then

\[
\log\text{-sum}\text{-exp}(v) = \log \sum_{n=1}^{N} \exp(v_n) = \max(v) + \log \sum_{n=1}^{N} \exp(v_n - \max(v)).
\]

The exponent cannot overflow because its argument is either zero or negative. This form makes it easy to calculate \( \log(u_1 + \cdots + u_N) \) given only \( \log u_n \).

Calculating means with log-sum-exp
An immediate application is to computing the mean of a vector \( u \) entirely on the log scale. That is, given \( \log u \) and returning \( \log \text{mean}(u) \).

\[
\log \left( \frac{1}{N} \sum_{n=1}^{N} u_n \right) = \log \frac{1}{N} + \log \sum_{n=1}^{N} \exp(\log u_n) = -\log N + \log\text{-sum}\text{-exp}(\log u).
\]

where \( \log u = (\log u_1, \ldots, \log u_N) \) is understood elementwise.

15.5. Comparing floating-point numbers
Because floating-point representations are inexact, it is rarely a good idea to test exact inequality. The general recommendation is that rather than testing \( x == y \), an approximate test may be used given an absolute or relative tolerance.

Given a positive absolute tolerance of \( \epsilon \), \( x \) can be compared to \( y \) using the conditional

\[ \text{abs}(x - y) \leq \epsilon. \]

Absolute tolerances work when the scale of \( x \) and \( y \) and the relevant comparison is known.

Given a positive relative tolerance of \( \epsilon \), a typical comparison is

\[ 2 \times \text{abs}(x - y) / (\text{abs}(x) + \text{abs}(y)) \leq \epsilon. \]
16. Matrices, Vectors, and Arrays

This chapter provides pointers as to how to choose among the various matrix, vector, and array data structures provided by Stan.

16.1. Basic motivation
Stan provides two basic scalar types, `int` and `real`, and three basic linear algebra types, `vector`, `row_vector`, and `matrix`. Then Stan allows arrays to be of any dimension and contain any type of element (though that type must be declared and must be the same for all elements).

This leaves us in the awkward situation of having three one-dimensional containers, as exemplified by the following declarations.

```plaintext
real a[N];
vector[N] a;
row_vector[N] a;
```

These distinctions matter. Matrix types, like vector and row vector, are required for linear algebra operations. There is no automatic promotion of arrays to vectors because the target, row vector or column vector, is ambiguous. Similarly, row vectors are separated from column vectors because multiplying a row vector by a column vector produces a scalar, whereas multiplying in the opposite order produces a matrix.

The following code fragment shows all four ways to declare a two-dimensional container of size $M \times N$.

```plaintext
real b[M, N]; // b[m] : real[] (efficient)
vector[N] b[M]; // b[m] : vector (efficient)
row_vector[N] b[M]; // b[m] : row_vector (efficient)
matrix[M, N] b; // b[m] : row_vector (inefficient)
```

The main differences among these choices involve efficiency for various purposes and the type of $b[m]$, which is shown in comments to the right of the declarations. Thus the only way to efficiently iterate over row vectors is to use the third declaration, but if you need linear algebra on matrices, but the only way to use matrix operations is to use the fourth declaration.

The inefficiencies due to any manual reshaping of containers is usually slight com-
pared to what else is going on in a Stan program (typically a lot of gradient calculations).

### 16.2. Fixed sizes and indexing out of bounds

Stan’s matrices, vectors, and array variables are sized when they are declared and may not be dynamically resized. Function arguments do not have sizes, but these sizes are fixed when the function is called and the container is instantiated. Also, declarations may be inside loops and thus may change over the course of running a program, but each time a declaration is visited, it declares a fixed size object.

When an index is provided that is out of bounds, Stan throws a rejection error and computation on the current log density and gradient evaluation is halted and the algorithm is left to clean up the error. All of Stan’s containers check the sizes of all indexes.

### 16.3. Data type and indexing efficiency

The underlying matrix and linear algebra operations are implemented in terms of data types from the Eigen C++ library. By having vectors and matrices as basic types, no conversion is necessary when invoking matrix operations or calling linear algebra functions.

Arrays, on the other hand, are implemented as instances of the C++ `std::vector` class (not to be confused with Eigen’s `Eigen::Vector` class or Stan vectors). By implementing arrays this way, indexing is efficient because values can be returned by reference rather than copied by value.

#### Matrices vs. two-dimensional arrays

In Stan models, there are a few minor efficiency considerations in deciding between a two-dimensional array and a matrix, which may seem interchangeable at first glance.

First, matrices use a bit less memory than two-dimensional arrays. This is because they don’t store a sequence of arrays, but just the data and the two dimensions.

Second, matrices store their data in column-major order. Furthermore, all of the data in a matrix is guaranteed to be contiguous in memory. This is an important consideration for optimized code because bringing in data from memory to cache is much more expensive than performing arithmetic operations with contemporary CPUs. Arrays, on the other hand, only guarantee that the values of primitive types are contiguous in memory; otherwise, they hold copies of their values (which are returned by reference wherever possible).

Third, both data structures are best traversed in the order in which they are stored.
This also helps with memory locality. This is column-major for matrices, so the following order is appropriate.

```
matrix[M, N] a;
//...
for (n in 1:N)
  for (m in 1:M)
    // ... do something with a[m, n] ...
```

Arrays, on the other hand, should be traversed in row-major order (i.e., last index fastest), as in the following example.

```
real a[M, N];
// ...
for (m in 1:M)
  for (n in 1:N)
    // ... do something with a[m, n] ...
```

The first use of $a[m,n]$ should bring $a[m]$ into memory. Overall, traversing matrices is more efficient than traversing arrays.

This is true even for arrays of matrices. For example, the ideal order in which to traverse a two-dimensional array of matrices is

```
matrix[M, N] b[I, J];
//...
for (i in 1:I)
  for (j in 1:J)
    for (n in 1:N)
      for (m in 1:M)
        ... do something with $b[i, j, m, n]$ ...
```

If $a$ is a matrix, the notation $a[m]$ picks out row $m$ of that matrix. This is a rather inefficient operation for matrices. If indexing of vectors is needed, it is much better to declare an array of vectors. That is, this

```
row_vector[N] b[M];
//...
for (m in 1:M)
  ... do something with row vector $b[m]$ ...
```

is much more efficient than the pure matrix version

```
matrix b[M, N];
```
// ...
for (m in 1:M)
    // ... do something with row vector b[m] ...

Similarly, indexing an array of column vectors is more efficient than using the col
function to pick out a column of a matrix.

In contrast, whatever can be done as pure matrix algebra will be the fastest. So if I
want to create a row of predictor-coefficient dot-products, it’s more efficient to do
this

```
matrix[N, k] x;       // predictors (aka covariates)
// ...
vector[K] beta;      // coeffs
// ...
vector[N] y_hat;     // linear prediction
// ...
y_hat = x * beta;
```

than it is to do this

```
row_vector[K] x[N];   // predictors (aka covariates)
// ...
vector[K] beta;      // coeffs
...
vector[N] y_hat;     // linear prediction
...
for (n in 1:N)
    y_hat[n] = x[n] * beta;
```

(Row) vectors vs. one-dimensional arrays
For use purely as a container, there is really nothing to decide among vectors, row
vectors and one-dimensional arrays. The Eigen::Vector template specialization
and the std::vector template class are implemented similarly as containers of
double values (the type real in Stan). Only arrays in Stan are allowed to store
integer values.

16.4. Memory locality
The key to understanding efficiency of matrix and vector representations is memory
locality and reference passing versus copying.
Memory locality
CPUs on computers bring in memory in blocks through layers of caches. Fetching from memory is much slower than performing arithmetic operations. The only way to make container operations fast is to respect memory locality and access elements that are close together in memory sequentially in the program.

Matrices
Matrices are stored internally in column-major order. That is, an $M \times N$ matrix stores its elements in the order

$$(1,1), (2,1), \ldots, (M,1), (1,2), \ldots, (M,2), \ldots, (1,N), \ldots, (M,N).$$

This means that it's much more efficient to write loops over matrices column by column, as in the following example.

```plaintext
matrix[M, N] a;
...
for (n in 1:N)
    for (m in 1:M)
        ... do something with a[m, n] ...
```

It also follows that pulling a row out of a matrix is not memory local, as it has to stride over the whole sequence of values. It also requires a copy operation into a new data structure as it is not stored internally as a unit in a matrix. For sequential access to row vectors in a matrix, it is much better to use an array of row vectors, as in the following example.

```plaintext
row_vector[N] a[M];
...
for (m in 1:M)
    ... do something with row vector a[m] ...
```

Even if what is done involves a function call, the row vector $a[m]$ will not have to be copied.

Arrays
Arrays are stored internally following their data structure. That means a two dimensional array is stored in row-major order. Thus it is efficient to pull out a “row” of a two-dimensional array.

```plaintext
real a[M, N];
...
for (m in 1:M)
```
A difference with matrices is that the entries $a[m]$ in the two dimensional array are not necessarily adjacent in memory, so there are no guarantees on iterating over all the elements in a two-dimensional array will provide memory locality across the “rows.”

16.5. Converting among matrix, vector, and array types
There is no automatic conversion among matrices, vectors, and arrays in Stan. But there are a wide range of conversion functions to convert a matrix into a vector, or a multi-dimensional array into a one-dimensional array, or convert a vector to an array. See the section on mixed matrix and array operations in the functions reference manual for a complete list of conversion operators and the multi-indexing chapter for some reshaping operations involving multiple indexing and range indexing.

16.6. Aliasing in Stan containers
Stan expressions are all evaluated before assignment happens, so there is no danger of so-called aliasing in array, vector, or matrix operations. Contrast the behavior of the assignments to $u$ and $x$, which start with the same values.

The loop assigning to $u$ and the compound slicing assigning to $x$.

The following trivial Stan program.

```stan
transformed data {
  vector[4] x = [ 1, 2, 3, 4 ];
  vector[4] u = [ 1, 2, 3, 4 ];

  for (t in 2:4)
    u[t] = u[t - 1] * 3;


  print("u = ", u);
  print("x = ", x);
}
```

The output it produces is,

$u = [1, 3, 9, 27]$

$x = [1, 3, 6, 9]$

In the loop version assigning to $u$, the values are updated before being used to define
subsequent values; in the sliced expression assigning to \( x \), the entire right-hand side is evaluated before assigning to the left-hand side.
17. Multiple Indexing and Range Indexing

Stan allows multiple indexes to be provided for containers (i.e., arrays, vectors, and matrices) in a single position, using either an array of integer indexes or range bounds. This allows many models to be vectorized. For instance, consider the likelihood for a varying-slope, varying-intercept hierarchical linear regression, which could be coded as

```plaintext
for (n in 1:N)
  y[n] ~ normal(alpha[ii[n]] + beta[ii[n]] * x[n], sigma);
```

With multiple indexing, this can be coded in one line, leading to more efficient vectorized code.

```plaintext
y ~ normal(alpha[ii] + rows_dot_product(beta[ii], x), sigma);
```

This latter version is equivalent in speed to the clunky assignment to a local variable.

```plaintext
{
  vector[N] mu;
  for (n in 1:N)
    mu[n] = alpha[ii[n]] + beta[ii[n]] * x[n];
  y ~ normal(mu, sigma);
}
```

17.1. Multiple indexing

The following is the simplest concrete example of multiple indexing with an array of integers; the ellipses stand for code defining the variables as indicated in the comments.

```plaintext
int c[3];
... // define: c == (5, 9, 7)
int idxs[4];
... // define: idxs == (3, 3, 1, 2)
int d[4];
d = c[idxs]; // result: d == (7, 7, 5, 9)
```

In general, the multiple indexed expression `c[idxs]` is defined as follows, assuming `idxs` is of size `K`.

```plaintext
c[idxs] = ( c[idxs[1]], c[idxs[2]], ..., c[idxs[K]] )
```
Thus \( c[\text{idxs}] \) is of the same size as \( \text{idxs} \), which is \( K \) in this example.

Multiple indexing can also be used with multi-dimensional arrays. For example, consider the following.

```c
int c[2, 3];
... // define: c = ((1, 3, 5), (7, 11, 13))
int idxs[4];
... // define: idxs = (2, 2, 1, 2)
int d[4, 3]
d = c[idxs]; // result: d = ((7, 11, 13), (7, 11, 13),
               // (1, 3, 5), (7, 11, 13))
```

That is, putting an index in the first position acts exactly the same way as defined above. The fact that the values are themselves arrays makes no difference—the result is still defined by \( c[\text{idxs}][j] == c[\text{idxs}[j]] \).

Multiple indexing may also be used in the second position of a multi-dimensional array. Continuing the above example, consider a single index in the first position and a multiple index in the second.

```c
int e[4];
e = c[2, idxs]; // result: c[2] = (7, 11, 13)
               // result: e = (11, 11, 7, 11)
```

The single index is applied, the one-dimensional result is determined, then the multiple index is applied to the result. That is, \( c[2, \text{idxs}] \) evaluates to the same value as \( c[2][\text{idxs}] \).

Multiple indexing can apply to more than one position of a multi-dimensional array. For instance, consider the following

```c
int c[2, 3];
... // define: c = ((1, 3, 5), (7, 11, 13))
int idxs1[3];
... // define: idxs1 = (2, 2, 1)
int idxs2[2];
... // define: idxs2 = (1, 3)
int d[3, 2];
d = c[idxs1, idxs2]; // result: d = ((7, 13), (7, 13), (1, 5))
```

With multiple indexes, we no longer have \( c[\text{idxs1}, \text{idxs2}] \) being the same as \( c[\text{idxs1}][\text{idxs2}] \). Rather, the entry \( d[i, j] \) after executing the above is given by
d[i, j] == c[idxs1, idxs2][i, j] = c[idxs1[i], idxs2[j]]

This example illustrates the operation of multiple indexing in the general case: a multiple index like idxs1 converts an index i used on the result (here, c[idxs1, idxs2]) to index idxs1[i] in the variable being indexed (here, c). In contrast, a single index just returns the value at that index, thus reducing dimensionality by one in the result.

17.2. Slicing with range indexes
Slicing returns a contiguous slice of a one-dimensional array, a contiguous sub-block of a two-dimensional array, and so on. Semantically, it is just a special form of multiple indexing.

Lower and upper bound indexes
For instance, consider supplying an upper and lower bound for an index.

```c
int c[7];
...
int d[4];
d = c[3:6]; // result: d == (c[3], c[4], c[5], c[6])
```

The range index 3:6 behaves semantically just like the multiple index (3, 4, 5, 6). In terms of implementation, the sliced upper and/or lower bounded indices are faster and use less memory because they do not explicitly create a multiple index, but rather use a direct loop. They are also easier to read, so should be preferred over multiple indexes where applicable.

Lower or upper bound indexes
It is also possible to supply just a lower bound, or just an upper bound. Writing c[3:] is just shorthand for c[3:size(c)]. Writing c[:5] is just shorthand for c[1:5].

Full range indexes
Finally, it is possible to write a range index that covers the entire range of an array, either by including just the range symbol (:) as the index or leaving the index position empty. In both cases, c[] and c[:] are equal to c[1:size(c)], which in turn is just equal to c.

17.3. Multiple indexing on the left of assignments
Multiple expressions may be used on the left-hand side of an assignment statement, where they work exactly the same way as on the right-hand side in terms of picking out entries of a container. For example, consider the following.

```c
int a[3];
```
CHAPTER 17. MULTIPLE INDEXING AND RANGE INDEXING

int c[2];
int idxs[2];
...
    // define: a == (1, 2, 3);  c == (5, 9)
    //       idxs = (3,2)
a[idxs] = c;  // result: a == (1, 9, 5)

The result above can be worked out by noting that the assignment sets \(a[idxs[1]]\) (\(a[3]\)) to \(c[1]\) (5) and \(a[idxs[2]]\) (\(a[2]\)) to \(c[2]\) (9).

The same principle applies when there are many multiple indexes, as in the following example.

int a[5, 7];
int c[2, 2];
...

As in the one-dimensional case, the right-hand side is written into the slice, block, or general chunk picked out by the left-hand side.

Usage on the left-hand side allows the full generality of multiple indexing, with single indexes reducing dimensionality and multiple indexes maintaining dimensionality while rearranging, slicing, or blocking. For example, it is valid to assign to a segment of a row of an array as follows.

int a[10, 13];
int c[2];
...

**Assign-by-value and aliasing**

Aliasing issues arise when there are references to the same data structure on the right-hand and left-hand side of an assignment. For example, consider the array \(a\) in the following code fragment.

int a[3];
...
    // define: a == (5, 6, 7)
a[2:3] = a[1:2];
...
    // result: a == (5, 5, 6)

The reason the value of \(a\) after the assignment is (5, 5, 6) rather than (5, 5, 5) is that Stan behaves as if the right-hand side expression is evaluated to a fresh copy. As another example, consider the following.
int a[3];
int idxs[3];
...
// define idxs = (2, 1, 3)
a[idxs] = a;

In this case, it is evident why the right-hand side needs to be copied before the assignment.

It is tempting (but wrong) to think of the assignment $a[2:3] = a[1:2]$ as executing the following assignments.

...
// define: a = (5, 6, 7)
a[2] = a[1];  // result: a = (5, 5, 7)
a[3] = a[2];  // result: a = (5, 5, 5)!

This produces a different result than executing the assignment because $a[2]$’s value changes before it is used.

17.4. Multiple indexes with vectors and matrices

Multiple indexes can be supplied to vectors and matrices as well as arrays of vectors and matrices.

**Vectors**


The only subtlety with vectors is in inferring the return type when there are multiple indexes. For example, consider the following minimal example.

```plaintext
vector[5] v[3];
int idxs[7];
...
vector[7] u;
u = v[2, idxs];

real w[7];
w = v[idxs, 2];
```

The key is understanding that a single index always reduces dimensionality, whereas a multiple index never does. The dimensions with multiple indexes (and unindexed dimensions) determine the indexed expression’s type. In the example above, because $v$ is an array of vectors, $v[2, idxs]$ reduces the array dimension but doesn’t reduce the vector dimension, so the result is a vector. In contrast, $v[idxs, 2]$ does not
reduce the array dimension, but does reduce the vector dimension (to a scalar), so
the result type for \( w \) is an array of reals. In both cases, the size of the multiple index
(here, 7) determines the size of the result.

**Matrices**

Matrices are a bit trickier because they have two dimensions, but the underlying
principle of type inference is the same—multiple indexes leave dimensions in place,
whereas single indexes reduce them. The following code shows how this works for
multiple indexing of matrices.

```cpp
matrix[5,7] m;
...
row_vector[3] rv;
rv = m[4, 3:5]; // result is 1 x 3
...
vector[4] v;
v = m[2:5, 3]; // result is 3 x 1
...
matrix[3, 4] m2;
m2 = m[1:3, 2:5]; // result is 3 x 4
```

The key is realizing that any position with a multiple index or bounded index remains
in play in the result, whereas any dimension with a single index is replaced with 1 in
the resulting dimensions. Then the type of the result can be read off of the resulting
dimensionality as indicated in the comments above.

**Matrices with one multiple index**

If matrices receive a single multiple index, the result is a matrix. So if \( m \) is a matrix,
so is \( m[2:4] \). In contrast, supplying a single index, \( m[3] \), produces a row vector
result. That is, \( m[3] \) produces the same result as \( m[3, ] \) or \( m[3, 1:cols(m)] \).

**Arrays of vectors or matrices**

With arrays of matrices, vectors, and row vectors, the basic access rules remain
exactly the same: single indexes reduce dimensionality and multiple indexes redirect
indexes. For example, consider the following example.

```cpp
matrix[3, 4] m[5, 7];
...
matrix[3, 4] a[2];
a = m[1, 2:3]; // knock off first array dimension
a = m[3:4, 5]; // knock off second array dimension
```

In both assignments, the multiple index knocks off an array dimension, but it's
different in both cases. In the first case, \( a[i] == m[1, i + 1] \), whereas in the second case, \( a[i] == m[i + 2, 5] \).

Continuing the previous example, consider the following.

```plaintext
... vector[2] b;
    b = a[1, 3, 2:3, 2];
```

Here, the two array dimensions are reduced as is the column dimension of the matrix, leaving only a row dimension index, hence the result is a vector. In this case, \( b[j] == a[1, 3, 1 + j, 2] \).

This last example illustrates an important point: if there is a lower-bounded index, such as 2:3, with lower bound 2, then the lower bound minus one is added to the index, as seen in the \( 1 + j \) expression above.

Continuing further, consider continuing with the following.

```plaintext
... row_vector[3] c[2];
    c = a[4:5, 3, 1, 2: ];
```

Here, the first array dimension is reduced, leaving a single array dimension, and the row index of the matrix is reduced, leaving a row vector. For indexing, the values are given by \( c[i, j] == a[i + 3, 3, 1, j + 1] \)

### 17.5. Matrices with parameters and constants

Suppose you have a 3x3 matrix and know that two entries are zero but the others are parameters. Such a situation arises in missing data situations and in problems with fixed structural parameters.

Suppose a 3 × 3 matrix is known to be zero at indexes \([1, 2]\) and \([1, 3]\). The indexes for parameters are included in a “melted” data-frame or database format.

```plaintext
transformed data {
    int<lower=1, upper=3> idxs[7, 2]
    = { {1, 1},
        {2, 1}, {2, 2}, {2, 3},
        {3, 1}, {3, 2}, {3, 3} };
...
```

The seven remaining parameters are declared as a vector.

```plaintext
parameters {
```
...

Then the full matrix $A$ is constructed in the model block as a local variable.

```
model {
    matrix[3, 3] A;
    for (i in 1:7)
        A[idxs[i, 1], idxs[i, 2]] = A_raw[i];
    A[1, 2] = 0;
    A[1, 3] = 0;
    ...
```

This may seem like overkill in this setting, but in more general settings, the matrix size, vector size, and the `idxs` array will be too large to code directly. Similar techniques can be used to build up matrices with ad-hoc constraints, such as a handful of entries known to be positive.
18. User-Defined Functions

This chapter explains functions from a user perspective with examples; see the language reference for a full specification. User-defined functions allow computations to be encapsulated into a single named unit and invoked elsewhere by name. Similarly, functions allow complex procedures to be broken down into more understandable components. Writing modular code using descriptively named functions is easier to understand than a monolithic program, even if the latter is heavily commented.\(^1\)

18.1. Basic functions

Here’s an example of a skeletal Stan program with a user-defined relative difference function employed in the generated quantities block to compute a relative differences between two parameters.

```stan
functions {
  real relative_diff(real x, real y) {
    real abs_diff;
    real avg_scale;
    abs_diff = fabs(x - y);
    avg_scale = (fabs(x) + fabs(y)) / 2;
    return abs_diff / avg_scale;
  }
}
...

generated quantities {
  real rdiff;
  rdiff = relative_diff(alpha, beta);
}
```

The function is named `relative_diff`, and is declared to have two real-valued arguments and return a real-valued result. It is used the same way a built-in function would be used in the generated quantities block.

---

\(^1\)The main problem with comments is that they can be misleading, either due to misunderstandings on the programmer’s part or because the program’s behavior is modified after the comment is written. The program always behaves the way the code is written, which is why refactoring complex code into understandable units is preferable to simply adding comments.
User-defined functions block

All functions are defined in their own block, which is labeled `functions` and must appear before all other program blocks. The user-defined functions block is optional.

Function bodies

The body (the part between the curly braces) contains ordinary Stan code, including local variables. The new function is used in the generated quantities block just as any of Stan’s built-in functions would be used.

Return statements

Return statements, such as the one on the last line of the definition of `relative_diff` above, are only allowed in the bodies of function definitions. Return statements may appear anywhere in a function, but functions with non-void return types must end in a return statement.

Reject statements

The Stan reject statement provides a mechanism to report errors or problematic values encountered during program execution. It accepts any number of quoted string literals or Stan expressions as arguments. This statement is typically embedded in a conditional statement in order to detect bad or illegal outcomes of some processing step.

Catching errors

Rejection is used to flag errors that arise in inputs or in program state. It is far better to fail early with a localized informative error message than to run into problems much further downstream (as in rejecting a state or failing to compute a derivative).

The most common errors that are coded is to test that all of the arguments to a function are legal. The following function takes a square root of its input, so requires non-negative inputs; it is coded to guard against illegal inputs.

```stan
real dbl_sqrt(real x) {
  if (!(x >= 0))
    reject("dblsqrt(x): x must be positive; found x = ", x);
  return 2 * sqrt(x);
}
```

The negation of the positive test is important, because it also catches the case where `x` is a not-a-number value. If the condition had been coded as `(x < 0)` it would not catch the not-a-number case, though it could be written as `(x < 0 || is_nan(x))`. The positive infinite case is allowed through, but could also be checked with the `is_inf(x)` function. The square root function does not itself reject, but
some downstream consumer of \texttt{dbl\_sqrt(-2)} would be likely to raise an error, at which point the origin of the illegal input requires detective work. Or even worse, as Matt Simpson pointed out in the GitHub comments, the function could go into an infinite loop if it starts with an infinite value and tries to reduce it by arithmetic, likely consuming all available memory and crashing an interface. Much better to catch errors early and report on their origin.

The effect of rejection depends on the program block in which the rejection is executed. In transformed data, rejections cause the program to fail to load. In transformed parameters or in the model block, rejections cause the current state to be rejected in the Metropolis sense.\footnote{Just because this makes it possible to code a rejection sampler does not make it a good idea. Rejections break differentiability and the smooth exploration of the posterior. In Hamiltonian Monte Carlo, it can cause the sampler to be reduced to a diffusive random walk.}

In generated quantities, rejections cause execution to halt because there is no way to recover and generate the remaining parameters, so extra care should be taken in calling functions in the generated quantities block.

**Type declarations for functions**

<table>
<thead>
<tr>
<th>Function</th>
<th>Argument</th>
<th>Local</th>
<th>Block</th>
</tr>
</thead>
<tbody>
<tr>
<td>int</td>
<td>(unsized)</td>
<td>(unconstrained)</td>
<td>(constrained)</td>
</tr>
<tr>
<td>real</td>
<td>(unsized)</td>
<td>(unconstrained)</td>
<td>(constrained)</td>
</tr>
<tr>
<td>vector</td>
<td>(unsized)</td>
<td>(unconstrained)</td>
<td>(constrained)</td>
</tr>
<tr>
<td>row_vector</td>
<td>(unsized)</td>
<td>(unconstrained)</td>
<td>(constrained)</td>
</tr>
</tbody>
</table>
The leftmost column is a list of the unconstrained and undimensioned basic types; these are used as function return types and argument types. The middle column is of unconstrained types with dimensions; these are used as local variable types. The variables $M$ and $N$ indicate number of columns and rows, respectively. The variable $K$ is used for square matrices, i.e., $K$ denotes both the number of rows and columns. The rightmost column lists the corresponding constrained types. An expression of any right-hand column type may be assigned to its corresponding left-hand column basic type. At runtime, dimensions are checked for consistency for all variables; containers of any sizes may be assigned to function arguments. The constrained matrix types $\text{cov}_\text{matrix}[K]$, $\text{corr}_\text{matrix}[K]$, $\text{cholesky}_\text{factor}_\text{cov}[K]$, and $\text{cholesky}_\text{factor}_\text{corr}[K]$ are only assignable to matrices of dimensions $\text{matrix}[K, K]$ types. Stan also allows arrays of any of these types, with slightly different declarations for function arguments and return types and variables.

Function argument and return types for vector and matrix types are not declared with their sizes, unlike type declarations for variables. Function argument type declarations may not be declared with constraints, either lower or upper bounds or structured constraints like forming a simplex or correlation matrix, (as is also the case for local variables); see the table of constrained types for full details.

For example, here’s a function to compute the entropy of a categorical distribution with simplex parameter $\theta$.

```c++
real entropy(vector theta) {
  return sum(theta .* log(theta));
}
```
Although $\theta$ must be a simplex, only the type vector is used.\footnote{A range of built-in validation routines is coming to Stan soon! Alternatively, the reject statement can be used to check constraints on the simplex.}

Upper or lower bounds on values or constrained types are not allowed as return types or argument types in function declarations.

**Array types for function declarations**

Array arguments have their own syntax, which follows that used in this manual for function signatures. For example, a function that operates on a two-dimensional array to produce a one-dimensional array might be declared as follows.

```c
real[] baz(real[,] x);
```

The notation `[ ]` is used for one-dimensional arrays (as in the return above), `[ , ]` for two-dimensional arrays, `[ , , ]` for three-dimensional arrays, and so on.

Functions support arrays of any type, including matrix and vector types. As with other types, no constraints are allowed.

**Data-only function arguments**

A function argument which is a real-valued type or a container of a real-valued type, i.e., not an integer type or integer array type, can be qualified using the prefix qualifier data. The following is an example of a data-only function argument.

```c
real foo(real y, data real mu) {
  return -0.5 * (y - mu)^2;
}
```

This qualifier restricts this argument to being invoked with expressions which consist only of data variables, transformed data variables, literals, and function calls. A data-only function argument cannot involve real variables declared in the parameters, transformed parameters, or model block. Attempts to invoke a function using an expression which contains parameter, transformed parameters, or model block variables as a data-only argument will result in an error message from the parser.

Use of the data qualifier must be consistent between the forward declaration and the definition of a function.

This qualifier should be used when writing functions that call the built-in ordinary differential equation (ODE) solvers, algebraic solvers, or map functions. These higher-order functions have strictly specified signatures where some arguments of are data only expressions. (See the ODE solver chapter for more usage details and the functions reference manual for full definitions.)
which calls the ODE or algebraic solver, arguments to that function which are passed into the call to the solver, either directly or indirectly, should have the data prefix qualifier. This allows for compile-time type checking and increases overall program understandability.

18.2. Functions as statements
In some cases, it makes sense to have functions that do not return a value. For example, a routine to print the lower-triangular portion of a matrix can be defined as follows.

functions {
    void pretty_print_tri_lower(matrix x) {
        if (rows(x) == 0) {
            print("empty matrix");
            return;
        }
        print("rows=", rows(x), " cols=", cols(x));
        for (m in 1:rows(x))
            for (n in 1:m)
                print("[", m, ",", n, "]="", x[m, n]);
    }
}

The special symbol void is used as the return type. This is not a type itself in that there are no values of type void; it merely indicates the lack of a value. As such, return statements for void functions are not allowed to have arguments, as in the return statement in the body of the previous example.

Void functions applied to appropriately typed arguments may be used on their own as statements. For example, the pretty-print function defined above may be applied to a covariance matrix being defined in the transformed parameters block.

transformed parameters {
    cov_matrix[K] Sigma;
    ... code to set Sigma ...
    pretty_print_tri_lower(Sigma);
    ...
}

18.3. Functions accessing the log probability accumulator
Functions whose names end in _lp are allowed to use sampling statements and target += statements; other functions are not. Because of this access, their use is restricted to the transformed parameters and model blocks.
Here is an example of a function to assign standard normal priors to a vector of coefficients, along with a center and scale, and return the translated and scaled coefficients; see the reparameterization section for more information on efficient non-centered parameterizations

functions {
  vector center_lp(vector beta_raw, real mu, real sigma) {
    beta_raw ~ std_normal();
    sigma ~ cauchy(0, 5);
    mu ~ cauchy(0, 2.5);
    return sigma * beta_raw + mu;
  }
  ...
}

parameters {
  vector[K] beta_raw;
  real mu_beta;
  real<lower=0> sigma_beta;
  ...
  transformed parameters {
    vector[K] beta;
    ...
    beta = center_lp(beta_raw, mu_beta, sigma_beta);
    ...
}

18.4. Functions acting as random number generators
A user-specified function can be declared to act as a (pseudo) random number generator (PRNG) by giving it a name that ends in _rng. Giving a function a name that ends in _rng allows it to access built-in functions and user-defined functions that end in _rng, which includes all the built-in PRNG functions. Only functions ending in _rng are able access the built-in PRNG functions. The use of functions ending in _rng must therefore be restricted to transformed data and generated quantities blocks like other PRNG functions; they may also be used in the bodies of other user-defined functions ending in _rng.

For example, the following function generates an \( N \times K \) data matrix, the first column of which is filled with 1 values for the intercept and the remaining entries of which have values drawn from a standard normal PRNG.

matrix predictors_rng(int N, int K) {
  matrix[N, K] x;
for (n in 1:N) {
  x[n, 1] = 1.0; // intercept
  for (k in 2:K)
    x[n, k] = normal_rng(0, 1);
}
return x;
}
The following function defines a simulator for regression outcomes based on a data matrix $x$, coefficients $\beta$, and noise scale $\sigma$.

```c
vector regression_rng(vector beta, matrix x, real sigma) {
  vector[rows(x)] y;
  vector[rows(x)] mu;
  mu = x * beta;
  for (n in 1:rows(x))
    y[n] = normal_rng(mu[n], sigma);
  return y;
}
```
These might be used in a generated quantity block to simulate some fake data from a fitted regression model as follows.

```c
parameters {
  vector[K] beta;
  real<lower=0> sigma;
  ...
}
generated quantities {
  matrix[N_sim, K] x_sim;
  vector[N_sim] y_sim;
  x_sim = predictors_rng(N_sim, K);
  y_sim = regression_rng(beta, x_sim, sigma);
}
```
A more sophisticated simulation might fit a multivariate normal to the predictors $x$ and use the resulting parameters to generate multivariate normal draws for $x_{\text{sim}}$.

### 18.5. User-defined probability functions

Probability functions are distinguished in Stan by names ending in `_lpdf` for density functions and `_lpmf` for mass functions; in both cases, they must have `real` return types.

Suppose a model uses several standard normal distributions, for which there is not
a specific overloaded density nor defaults in Stan. So rather than writing out the
location of 0 and scale of 1 for all of them, a new density function may be defined
and reused.

functions {
    real unit_normal_lpdf(real y) {
        return normal_lpdf(y | 0, 1);
    }
}

... model {
    alpha ~ unit_normal();
    beta ~ unit_normal();
    ...
}

The ability to use the unit_normal function as a density is keyed off its name ending
in _lpdf (names ending in _lpmf for probability mass functions work the same
way).

In general, if foo_lpdf is defined to consume \( N + 1 \) arguments, then

\[ y \sim \text{foo}(\text{theta1, \ldots, thetaN}); \]

can be used as shorthand for

\[ \text{target} += \text{foo_lpdf}(y | \text{theta1, \ldots, thetaN}); \]

As with the built-in functions, the suffix _lpdf is dropped and the first argument
moves to the left of the sampling symbol (\( \sim \)) in the sampling statement.

Functions ending in _lpmf (for probability mass functions), behave exactly the same
way. The difference is that the first argument of a density function (_lpdf) must be
continuous (not an integer or integer array), whereas the first argument of a mass
function (_lpmf) must be discrete (integer or integer array).

**18.6. Overloading functions**

Stan does not permit overloading user-defined functions. This means that it is not
possible to define two different functions with the same name, even if they have
different signatures.

**18.7. Documenting functions**

Functions will ideally be documented at their interface level. The Stan style guide
for function documentation follows the same format as used by the Doxygen (C++)
and Javadoc (Java) automatic documentation systems. Such specifications indicate
the variables and their types and the return value, prefaced with some descriptive
text.

For example, here’s some documentation for the prediction matrix generator.

```java
/**
 * Return a data matrix of specified size with rows
 * corresponding to items and the first column filled
 * with the value 1 to represent the intercept and the
 * remaining columns randomly filled with unit-normal draws.
 *
 * @param N Number of rows corresponding to data items
 * @param K Number of predictors, counting the intercept, per
 *           item.
 * @return Simulated predictor matrix.
 */
matrix predictors_rng(int N, int K) {
  ...}
```

The comment begins with /*/, ends with *//, and has an asterisk (*) on each line. It
uses @param followed by the argument’s identifier to document a function argument.
The tag @return is used to indicate the return value. Stan does not (yet) have an
automatic documentation generator like Javadoc or Doxygen, so this just looks like
a big comment starting with /* and ending with */ to the Stan parser.

For functions that raise exceptions, exceptions can be documented using @throws.4

For example,

```java
...
 * @param theta
 * @throws If any of the entries of theta is negative.
 */
real entropy(vector theta) {
  ...

```

Usually an exception type would be provided, but these are not exposed as part of
the Stan language, so there is no need to document them.

---

4As of Stan 2.9.0, the only way a user-defined producer will raise an exception is if a function it calls
(including sampling statements) raises an exception via the reject statement.
18.8. Summary of function types

Functions may have a void or non-void return type and they may or may not have one of the special suffixes, \_lpdf, \_lpmf, \_lp, or \_rng.

Void vs. non-void return

Only functions declared to return void may be used as statements. These are also the only functions that use return statements with no arguments.

Only functions declared to return non-void values may be used as expressions. These functions require return statements with arguments of a type that matches the declared return type.

Suffixed or non-suffixed

Only functions ending in \_lpmf or \_lpdf and with return type real may be used as probability functions in sampling statements.

Only functions ending in \_lp may access the log probability accumulator through sampling statements or target += statements. Such functions may only be used in the transformed parameters or model blocks.

Only functions ending in \_rng may access the built-in pseudo-random number generators. Such functions may only be used in the generated quantities block or transformed data block, or in the bodies of other user-defined functions ending in \_rng.

18.9. Recursive functions

Stan supports recursive function definitions, which can be useful for some applications. For instance, consider the matrix power operation, $A^n$, which is defined for a square matrix $A$ and positive integer $n$ by

$$A^n = \begin{cases} 
I & \text{if } n = 0, \text{ and} \\
A A^{n-1} & \text{if } n > 0.
\end{cases}$$

where $I$ is the identity matrix. This definition can be directly translated to a recursive function definition.

```cpp
matrix matrix_pow(matrix a, int n);

matrix matrix_pow(matrix a, int n) {
  if (n == 0)
    return diag_matrix(rep_vector(1, rows(a)));
  else
    return a * matrix_pow(a, n - 1);
}
```
The forward declaration of the function signature before it is defined is necessary so that the embedded use of `matrix_pow` is well-defined when it is encountered. It would be more efficient to not allow the recursion to go all the way to the base case, adding the following conditional clause.

```c
else if (n == 1)
    return a;
```

### 18.10. Truncated random number generation

#### Generation with inverse CDFs

To generate random numbers, it is often sufficient to invert their cumulative distribution functions. This is built into many of the random number generators. For example, to generate a standard logistic variate, first generate a uniform variate \( u \sim \text{uniform}(0, 1) \), then run through the inverse cumulative distribution function, \( y = \logit(u) \). If this were not already built in as `logistic_rng(0, 1)` it could be coded in Stan directly as

```c
real standard_logistic_rng() {
    real u = uniform_rng(0, 1);
    real y = logit(u);
    return y;
}
```

Following the same pattern, a standard normal RNG could be coded as

```c
real standard_normal_rng() {
    real u = uniform_rng(0, 1);
    real y = inv_Phi(u);
    return y;
}
```

that is, \( y = \Phi^{-1}(u) \), where \( \Phi^{-1} \) is the inverse cumulative distribution function for the standard normal distribution, implemented in the Stan function `inv_Phi`.

In order to generate non-standard variates of the location-scale variety, the variate is scaled by the scale parameter and shifted by the location parameter. For example, to generate \( \text{normal}(\mu, \sigma) \) variates, it is enough to generate a uniform variate \( u \sim \text{uniform}(0, 1) \), then convert it to a standard normal variate, \( z = \Phi(u) \), where \( \Phi \) is the inverse cumulative distribution function for the standard normal, and then, finally, scale and translate it, \( y = \mu + \sigma \times z \). In code,
real my_normal_rng(real mu, real sigma) {
    real u = uniform_rng(0, 1);
    real z = inv_Phi(u);
    real y = mu + sigma * z;
    return y;
}

A robust version of this function would test that the arguments are finite and that
sigma is non-negative, e.g.,

if (is_nan(mu) || is_inf(mu))
    reject("my_normal_rng: mu must be finite; ",
           "found mu = ", mu);
if (is_nan(sigma) || is_inf(sigma) || sigma < 0)
    reject("my_normal_rng: sigma must be finite and non-negative; ",
           "found sigma = ", sigma);

Truncated variate generation

Often truncated uniform variates are needed, as in survival analysis when a time
of death is censored beyond the end of the observations. To generate a truncated
random variate, the cumulative distribution is used to find the truncation point in
the inverse CDF, a uniform variate is generated in range, and then the inverse CDF
translates it back.

Truncating below

For example, the following code generates a \textit{Weibull}(\alpha, \sigma) variate truncated below
at a time \( t \),\(^5\)

real weibull_lb_rng(real alpha, real sigma, real t) {
    real p = weibull_cdf(lt, alpha, sigma); // cdf for lb
    real u = uniform_rng(p, 1); // unif in bounds
    real y = sigma * (-log1m(u))^inv(alpha); // inverse cdf
    return y;
}

\(^5\)The original code and impetus for including this in the manual came from the Stan forums post http://discourse.mc-stan.org/t/rng-for-truncated-distributions/3122/7; by user lcomm, who also explained truncation above and below.
Truncating above and below

If there is a lower bound and upper bound, then the CDF trick is used twice to find a lower and upper bound. For example, to generate a \( \text{normal}(\mu, \sigma) \) truncated to a region \((a, b)\), the following code suffices,

```plaintext
real normal_lub_rng(real mu, real sigma, real lb, real ub) {
    real p_lb = normal_cdf(lb, mu, sigma);
    real p_ub = normal_cdf(ub, mu, sigma);
    real u = uniform_rng(p_lb, p_ub);
    real y = mu + sigma * inv_Phi(u);
    return y;
}
```

To make this more robust, all variables should be tested for finiteness, \( \sigma \) should be tested for positiveness, and \( lb \) and \( ub \) should be tested to ensure the upper bound is greater than the lower bound. While it may be tempting to compress lines, the variable names serve as a kind of chunking of operations and naming for readability; compare the multiple statement version above with the single statement

```plaintext
return mu + sigma * inv_Phi(uniform_rng(normal_cdf(lb, mu, sigma),
                                     normal_cdf(ub, mu, sigma)));
```

for readability. The names like \( p \) indicate probabilities, and \( p_{lb} \) and \( p_{ub} \) indicate the probabilities of the bounds. The variable \( u \) is clearly named as a uniform variate, and \( y \) is used to denote the variate being generated itself.
Custom distributions may also be implemented directly within Stan’s programming language. The only thing that is needed is to increment the total log probability. The rest of the chapter provides examples.

19.1. Examples

Triangle distribution
A simple example is the triangle distribution, whose density is shaped like an isosceles triangle with corners at specified bounds and height determined by the constraint that a density integrate to 1. If $\alpha \in \mathbb{R}$ and $\beta \in \mathbb{R}$ are the bounds, with $\alpha < \beta$, then $y \in (\alpha, \beta)$ has a density defined as follows.

$$
\text{triangle}(y \mid \alpha, \beta) = \frac{2}{\beta - \alpha} \left(1 - \left| y - \frac{\alpha + \beta}{\beta - \alpha} \right| \right)
$$

If $\alpha = -1$, $\beta = 1$, and $y \in (-1, 1)$, this reduces to

$$
\text{triangle}(y \mid -1, 1) = 1 - |y|.
$$

Consider the following Stan implementation of $\text{triangle}(-1, 1)$ for sampling.

```stan
parameters {
  real<lower=-1,upper=1> y;
}
model {
  target += log1m(fabs(y));
}
```

The single scalar parameter $y$ is declared as lying in the interval $(-1,1)$. The total log probability is incremented with the joint log probability of all parameters, i.e., $\log \text{Triangle}(y \mid -1, 1)$. This value is coded in Stan as $\log1m(fabs(y))$. The function $\log1m$ is defined so that $\log1m(x)$ has the same value as $\log(1 - x)$, but the computation is faster, more accurate, and more stable.

The constrained type `real<lower=-1,upper=1>` declared for $y$ is critical for correct sampling behavior. If the constraint on $y$ is removed from the program, say by declaring $y$ as having the unconstrained scalar type `real`, the program would compile,
but it would produce arithmetic exceptions at run time when the sampler explored values of \( y \) outside of \((-1, 1)\).

Now suppose the log probability function were extended to all of \( \mathbb{R} \) as follows by defining the probability to be \( \log(0.0) \), i.e., \(-\infty\), for values outside of \((-1, 1)\).

```c++
target += \log(fmax(0.0,1 - fabs(y)))
```

With the constraint on \( y \) in place, this is just a less efficient, slower, and less arithmetically stable version of the original program. But if the constraint on \( y \) is removed, the model will compile and run without arithmetic errors, but will not sample properly.\(^1\)

**Exponential distribution**

If Stan didn’t happen to include the exponential distribution, it could be coded directly using the following assignment statement, where \( \lambda \) is the inverse scale and \( y \) the sampled variate.

```c++
target += \log(\lambda) - y * \lambda;
```

This encoding will work for any \( \lambda \) and \( y \); they can be parameters, data, or one of each, or even local variables.

The assignment statement in the previous paragraph generates C++ code that is similar to that generated by the following sampling statement.

```c++
y ~ exponential(\lambda);
```

There are two notable differences. First, the sampling statement will check the inputs to make sure both \( \lambda \) is positive and \( y \) is non-negative (which includes checking that neither is the special not-a-number value).

The second difference is that if \( \lambda \) is not a parameter, transformed parameter, or local model variable, the sampling statement is clever enough to drop the \( \log(\lambda) \) term. This results in the same posterior because Stan only needs the log probability up to an additive constant. If \( \lambda \) and \( y \) are both constants, the sampling statement will drop both terms (but still check for out-of-domain errors on the inputs).

\(^1\)The problem is the (extremely!) light tails of the triangle distribution. The standard HMC and NUTS samplers can’t get into the corners of the triangle properly. Because the Stan code declares \( y \) to be of type `real<lower = -1, upper = 1>`, the inverse logit transform is applied to the unconstrained variable and its log absolute derivative added to the log probability. The resulting distribution on the logit-transformed \( y \) is well behaved.
Bivariate normal cumulative distribution function

For another example of user-defined functions, consider the following definition of the bivariate normal cumulative distribution function (CDF) with location zero, unit variance, and correlation $\rho$. That is, it computes

\[
\text{binormal}_\text{cdf}(z_1, z_2, \rho) = \Pr[Z_1 > z_1 \text{ and } Z_2 > z_2]
\]

where the random 2-vector $Z$ has the distribution

\[
Z \sim \text{ multivariate normal } \left( \begin{bmatrix} 0 \\ 0 \end{bmatrix}, \begin{bmatrix} 1 & \rho \\ \rho & 1 \end{bmatrix} \right).
\]

The following Stan program implements this function,

```stan
real binormal_cdf(real z1, real z2, real rho) {
  if (z1 != 0 || z2 != 0) {
    real denom = fabs(rho) < 1.0 ? sqrt((1 + rho) * (1 - rho)) : not_a_number();
    real a1 = (z2 / z1 - rho) / denom;
    real a2 = (z1 / z2 - rho) / denom;
    real product = z1 * z2;
    real delta = product < 0 || (product == 0 && (z1 + z2) < 0);
    return 0.5 * (Phi(z1) + Phi(z2) - delta) - owens_t(z1, a1) - owens_t(z2, a2);
  }
  return 0.25 + asin(rho) / (2 * pi());
}
```

20. Proportionality Constants

When evaluating a likelihood or prior as part of the log density computation in MCMC, variational inference, or optimization, it is usually only necessary to compute the functions up to a proportionality constant (or similarly compute log densities up to an additive constant). In MCMC this comes from the fact that the distribution being sampled does not need to be normalized (and so it is the normalization constant that is ignored). Similarly the distribution does not need normalized to perform variational inference or do optimizations. The advantage of working with unnormalized distributions is they can make computation quite a bit cheaper.

There are three different syntaxes to work with distributions in Stan. The way to select between them is by determining if the proportionality constants are necessary. If performance is not a problem, it is always safe to use the normalized densities.

The first two syntaxes use unnormalized densities (dropping proportionality constants):

\[
x \sim \text{normal}(0, 1);
\]
\[
\text{target} += \text{normal\_lupdf}(x | 0, 1); \quad // \text{the 'u' is for unnormalized}
\]

The final syntax uses the full normalized density (dropping no constants):

\[
\text{target} += \text{normal\_lpdf}(x | 0, 1);
\]

For discrete distributions, the target += syntax is \_lupmf and \_lpmf instead:

\[
y \sim \text{bernoulli}(0.5);
\]
\[
\text{target} += \text{bernoulli\_lupmf}(y | 0.5);
\]
\[
\text{target} += \text{bernoulli\_lpmf}(y | 0.5);
\]

20.1. Dropping Proportionality Constants

If a density \( p(\theta) \) can be factored into \( Kg(\theta) \) where \( K \) are all the factors that are not a function of \( \theta \) and \( g(\theta) \) are all the terms that are a function of \( \theta \), then it is said that \( g(\theta) \) is proportional to \( p(\theta) \) up to a constant.

The advantage of all this is that sometimes \( K \) is expensive to compute and if it is not a function of the distribution that is to be sampled (or optimized or approximated with variational inference), there is no need to compute it because it will not affect the results.
Stan takes advantage of the proportionality constant fact with the \( \sim \) syntax. Take for instance the normal likelihood:

```stan
data {
  real mu;
  real<lower = 0.0> sigma;
}
parameters {
  real x;
}
model {
  x ~ normal(mu, sigma);
}
```

Syntactically, this is just shorthand for the equivalent model that replaces the \( \sim \) syntax with a `target +=` statement and a `normal_lupdf` function call:

```stan
data {
  real mu;
  real<lower = 0.0> sigma;
}
parameters {
  real x;
}
model {
  target += normal_lupdf(x | mu, sigma)
}
```

The function `normal_lupdf` is only guaranteed to return the log density of the normal distribution up to a proportionality constant density to be sampled. The proportionality constant itself is not defined. The full log density of the statement here is:

\[
normal_lpdf(x | \mu, \sigma) = -\log \left( \sigma \sqrt{2\pi} \right) - \frac{1}{2} \left( \frac{x - \mu}{\sigma} \right)^2
\]

Now because the density here is only a function of \( x \), the additive terms in the log density that are not a function of \( x \) can be dropped. In this case it is enough to know only the quadratic term:
normal\_lupdf(x|\mu, \sigma) = -\frac{1}{2} \left( \frac{x - \mu}{\sigma} \right)^2

20.2. Keeping Proportionality Constants
In the case that the proportionality constants were needed for a normal log density the function \texttt{normal\_lpdf} can be used. For clarity, if there is ever a situation where it is unclear if the normalization is necessary, it should always be safe to include it. Only use the ~ or \texttt{target += normal\_lupdf} syntaxes if it is absolutely clear that the proportionality constants are not necessary.

20.3. User-defined Distributions
When a custom \texttt{\_lpdf} or \texttt{\_lpmf} function is defined, the compiler will automatically make available a \texttt{\_lupdf} or \texttt{\_lupmf} version of the function. It is only possible to define custom distributions in the normalized form in Stan. Any attempt to define an unnormalized distribution directly will result in an error.

The difference in the normalized and unnormalized versions of custom probability functions is how probability functions are treated inside these functions. Any internal unnormalized probability function call will be replaced with its normalized equivalent if the normalized version of the parent custom distribution is called.

The following code demonstrates the different behaviors:

```stan
functions {
  real custom1_lpdf(x) {
    return normal_lupdf(x | 0.0, 1.0)
  }
  real custom2_lpdf(x) {
    return normal_lpdf(x | 0.0, 1.0)
  }
}
parameters {
  real mu;
}
model {
  mu ~ custom1(); // Normalization constants dropped
  target += custom1_lupdf(mu); // Normalization constants dropped
  target += custom1_lpdf(mu); // Normalization constants kept
  mu ~ custom2(); // Normalization constants kept
```
target += custom2_lupdf(mu);  // Normalization constants kept
target += custom2_lpdf(mu);  // Normalization constants kept
}

20.4. Limitations on Using _lupdf and _lupmf Functions

To avoid ambiguities in how the normalization constants work, functions ending in _lupdf and _lupmf can only be used in the model block or user-defined probability functions (functions ending in _lpdf or _lpmf).
21. Problematic Posteriors

Mathematically speaking, with a proper posterior, one can do Bayesian inference and that’s that. There is not even a need to require a finite variance or even a finite mean—all that’s needed is a finite integral. Nevertheless, modeling is a tricky business and even experienced modelers sometimes code models that lead to improper priors. Furthermore, some posteriors are mathematically sound, but ill-behaved in practice. This chapter discusses issues in models that create problematic posterior inferences, either in general for Bayesian inference or in practice for Stan.

21.1. Collinearity of predictors in regressions

This section discusses problems related to the classical notion of identifiability, which lead to ridges in the posterior density and wreak havoc with both sampling and inference.

Examples of collinearity

Redundant intercepts

The first example of collinearity is an artificial example involving redundant intercept parameters.\(^1\)

Suppose there are observations \(y_n\) for \(n \in \{1, \ldots, N\}\), two intercept parameters \(\lambda_1\) and \(\lambda_2\), a scale parameter \(\sigma > 0\), and the sampling distribution

\[
y_n \sim \text{normal}(\lambda_1 + \lambda_2, \sigma).
\]

For any constant \(q\), the sampling density for \(y\) does not change if we add \(q\) to \(\lambda_1\) and subtract it from \(\lambda_2\), i.e.,

\[
p(y \mid \lambda_1, \lambda_2, \sigma) = p(y \mid \lambda_1 + q, \lambda_2 - q, \sigma).
\]

The consequence is that an improper uniform prior \(p(\mu, \sigma) \propto 1\) leads to an improper posterior. This impropriety arises because the neighborhoods around \(\lambda_1 + q, \lambda_2 - q\) have the same mass no matter what \(q\) is. Therefore, a sampler would need to spend as much time in the neighborhood of \(\lambda_1 = 100000000\) and \(\lambda_2 = -100000000\)

\(^1\)This example was raised by Richard McElreath on the Stan users group in a query about the difference in behavior between Gibbs sampling as used in BUGS and JAGS and the Hamiltonian Monte Carlo (HMC) and no-U-turn samplers (NUTS) used by Stan.
as it does in the neighborhood of $\lambda_1 = 0$ and $\lambda_2 = 0$, and so on for ever more far-ranging values.

The marginal posterior $p(\lambda_1, \lambda_2 \mid y)$ for this model is thus improper.\(^2\)

The impropriety shows up visually as a ridge in the posterior density, as illustrated in the left-hand plot. The ridge for this model is along the line where $\lambda_2 = \lambda_1 + c$ for some constant $c$.

Contrast this model with a simple regression with a single intercept parameter $\mu$ and sampling distribution

$$y_n \sim \text{normal}(\mu, \sigma).$$

Even with an improper prior, the posterior is proper as long as there are at least two data points $y_n$ with distinct values.

**Ability and difficulty in IRT models**

Consider an item-response theory model for students $j \in 1:J$ with abilities $\alpha_j$ and test items $i \in 1:I$ with difficulties $\beta_i$. The observed data are an $I \times J$ array with entries $y_{i,j} \in \{0, 1\}$ coded such that $y_{i,j} = 1$ indicates that student $j$ answered question $i$ correctly. The sampling distribution for the data is

$$y_{i,j} \sim \text{Bernoulli}(\logit^{-1}(\alpha_j - \beta_i)).$$

For any constant $c$, the probability of $y$ is unchanged by adding a constant $c$ to all the abilities and subtracting it from all the difficulties, i.e.,

$$p(y \mid \alpha, \beta) = p(y \mid \alpha + c, \beta - c).$$

This leads to a multivariate version of the ridge displayed by the regression with two intercepts discussed above.

**General collinear regression predictors**

The general form of the collinearity problem arises when predictors for a regression are collinear. For example, consider a linear regression sampling distribution

$$y_n \sim \text{normal}(x_n \beta, \sigma)$$

\(^2\)The marginal posterior $p(\sigma \mid y)$ for $\sigma$ is proper here as long as there are at least two distinct data points.
for an $N$-dimensional observation vector $y$, an $N \times K$ predictor matrix $x$, and a $K$-dimensional coefficient vector $\beta$.

Now suppose that column $k$ of the predictor matrix is a multiple of column $k'$, i.e., there is some constant $c$ such that $x_{n,k} = c x_{n,k'}$ for all $n$. In this case, the coefficients $\beta_k$ and $\beta_{k'}$ can covary without changing the predictions, so that for any $d \neq 0$,

$$p(y \mid \ldots, \beta_k, \ldots, \beta_{k'}, \ldots, \sigma) = p(y \mid \ldots, d\beta_k, \ldots, -\frac{d}{c} \beta_{k'}, \ldots, \sigma).$$

Even if columns of the predictor matrix are not exactly collinear as discussed above, they cause similar problems for inference if they are nearly collinear.

**Multiplicative issues with discrimination in IRT**

Consider adding a discrimination parameter $\delta_i$ for each question in an IRT model, with data sampling model

$$y_{i,j} \sim \text{Bernoulli}(\logit^{-1}(\delta_i (\alpha_j - \beta_i))).$$

For any constant $c \neq 0$, multiplying $\delta$ by $c$ and dividing $\alpha$ and $\beta$ by $c$ produces the same likelihood,

$$p(y \mid \delta, \alpha, \beta) = p(y \mid c\delta, \frac{1}{c}\alpha, \frac{1}{c}\beta).$$

If $c < 0$, this switches the signs of every component in $\alpha$, $\beta$, and $\delta$ without changing the density.

**Softmax with $K$ vs. $K - 1$ parameters**

In order to parameterize a $K$-simplex (i.e., a $K$-vector with non-negative values that sum to one), only $K - 1$ parameters are necessary because the $K$th is just one minus the sum of the first $K - 1$ parameters, so that if $\theta$ is a $K$-simplex,

$$\theta_K = 1 - \sum_{k=1}^{K-1} \theta_k.$$

The softmax function maps a $K$-vector $\alpha$ of linear predictors to a $K$-simplex $\theta = \text{softmax}(\alpha)$ by defining

$$\theta_k = \frac{\exp(\alpha_k)}{\sum_{k'=1}^K \exp(\alpha_{k'})}.$$
The softmax function is many-to-one, which leads to a lack of identifiability of the unconstrained parameters \( \alpha \). In particular, adding or subtracting a constant from each \( \alpha_k \) produces the same simplex \( \theta \).

**Mitigating the invariances**

All of the examples discussed in the previous section allow translation or scaling of parameters while leaving the data probability density invariant. These problems can be mitigated in several ways.

**Removing redundant parameters or predictors**

In the case of the multiple intercepts, \( \lambda_1 \) and \( \lambda_2 \), the simplest solution is to remove the redundant intercept, resulting in a model with a single intercept parameter \( \mu \) and sampling distribution \( y_n \sim \text{normal}(\mu, \sigma) \). The same solution works for solving the problem with collinearity—just remove one of the columns of the predictor matrix \( x \).

**Pinning parameters**

The IRT model without a discrimination parameter can be fixed by pinning one of its parameters to a fixed value, typically 0. For example, the first student ability \( \alpha_1 \) can be fixed to 0. Now all other student ability parameters can be interpreted as being relative to student 1. Similarly, the difficulty parameters are interpretable relative to student 1’s ability to answer them.

This solution is not sufficient to deal with the multiplicative invariance introduced by the question discrimination parameters \( \delta_i \). To solve this problem, one of the difficulty parameters, say \( \delta_1 \), must also be constrained. Because it’s a multiplicative and not an additive invariance, it must be constrained to a non-zero value, with 1 being a convenient choice. Now all of the discrimination parameters may be interpreted relative to item 1’s discrimination.

The many-to-one nature of softmax(\( \alpha \)) is typically mitigated by pinning a component of \( \alpha \), for instance fixing \( \alpha_K = 0 \). The resulting mapping is one-to-one from \( K - 1 \) unconstrained parameters to a \( K \)-simplex. This is roughly how simplex-constrained parameters are defined in Stan; see the reference manual chapter on constrained parameter transforms for a precise definition. The Stan code for creating a simplex from a \( K - 1 \)-vector can be written as

```stan
vector softmax_id(vector alpha) {
  vector[num_elements(alpha) + 1] alphac1;
  for (k in 1:num_elements(alpha))
    alphac1[k] = alpha[k];
```

Adding priors

So far, the models have been discussed as if the priors on the parameters were improper uniform priors.

A more general Bayesian solution to these invariance problems is to impose proper priors on the parameters. This approach can be used to solve problems arising from either additive or multiplicative invariance.

For example, normal priors on the multiple intercepts,
\[
\lambda_1, \lambda_2 \sim \text{normal}(0, \tau),
\]
with a constant scale \(\tau\), ensure that the posterior mode is located at a point where \(\lambda_1 = \lambda_2\), because this minimizes \(\log \text{normal}(\lambda_1 | 0, \tau) + \log \text{normal}(\lambda_2 | 0, \tau)\).\(^3\)

The following plots show the posteriors for two intercept parameterization without prior, two intercept parameterization with standard normal prior, and one intercept reparameterization without prior. For all three cases, the posterior is plotted for 100 data points drawn from a standard normal.

The two intercept parameterization leads to an improper prior with a ridge extending infinitely to the northwest and southeast.

Adding a standard normal prior for the intercepts results in a proper posterior.

The single intercept parameterization with no prior also has a proper posterior.

The addition of a prior to the two intercepts model is shown in the second plot; the final plot shows the result of reparameterizing to a single intercept.

An alternative strategy for identifying a \(K\)-simplex parameterization \(\theta = \text{softmax}(\alpha)\) in terms of an unconstrained \(K\)-vector \(\alpha\) is to place a prior on the components of \(\alpha\) with a fixed location (that is, specifically avoid hierarchical priors with varying location). Unlike the approaching of pinning \(\alpha_K = 0\), the prior-based approach models the \(K\) outcomes symmetrically rather than modeling \(K - 1\) outcomes relative to the \(K\)-th. The pinned parameterization, on the other hand, is usually more

\(^3\)A Laplace prior (or an L1 regularizer for penalized maximum likelihood estimation) is not sufficient to remove this additive invariance. It provides shrinkage, but does not in and of itself identify the parameters because adding a constant to \(\lambda_1\) and subtracting it from \(\lambda_2\) results in the same value for the prior density.
Figure 21.1: Two intercepts with improper prior

efficient statistically because it does not have the extra degree of (prior constrained) wiggle room.

Vague, strongly informative, and weakly informative priors

Care must be used when adding a prior to resolve invariances. If the prior is taken to be too broad (i.e., too vague), the resolution is in theory only, and samplers will still struggle.

Ideally, a realistic prior will be formulated based on substantive knowledge of the
Problem being modeled. Such a prior can be chosen to have the appropriate strength based on prior knowledge. A strongly informative prior makes sense if there is strong prior information.

When there is not strong prior information, a weakly informative prior strikes the proper balance between controlling computational inference without dominating the data in the posterior. In most problems, the modeler will have at least some notion of the expected scale of the estimates and be able to choose a prior for identification purposes that does not dominate the data, but provides sufficient computational

Figure 21.2: Two intercepts with proper prior
Proper Posterior (without Prior)

Figure 21.3: Single intercepts with improper prior

control on the posterior.

Priors can also be used in the same way to control the additive invariance of the IRT model. A typical approach is to place a strong prior on student ability parameters $\alpha$ to control scale simply to control the additive invariance of the basic IRT model and the multiplicative invariance of the model extended with a item discrimination parameters; such a prior does not add any prior knowledge to the problem. Then a prior on item difficulty can be chosen that is either informative or weakly informative based on prior knowledge of the problem.
21.2. Label switching in mixture models
Where collinearity in regression models can lead to infinitely many posterior maxima, swapping components in a mixture model leads to finitely many posterior maxima.

Mixture models
Consider a normal mixture model with two location parameters $\mu_1$ and $\mu_2$, a shared scale $\sigma > 0$, a mixture ratio $\theta \in [0, 1]$, and likelihood

$$p(y \mid \theta, \mu_1, \mu_2, \sigma) = \prod_{n=1}^{N} \left( \theta \text{normal}(y_n \mid \mu_1, \sigma) + (1 - \theta) \text{normal}(y_n \mid \mu_2, \sigma) \right).$$

The issue here is exchangeability of the mixture components, because

$$p(\theta, \mu_1, \mu_2, \sigma \mid y) = p((1 - \theta), \mu_2, \mu_1, \sigma \mid y).$$

The problem is exacerbated as the number of mixture components $K$ grows, as in clustering models, leading to $K!$ identical posterior maxima.

Convergence monitoring and effective sample size
The analysis of posterior convergence and effective sample size is also difficult for mixture models. For example, the $\hat{R}$ convergence statistic reported by Stan and the computation of effective sample size are both compromised by label switching. The problem is that the posterior mean, a key ingredient in these computations, is affected by label switching, resulting in a posterior mean for $\mu_1$ that is equal to that of $\mu_2$, and a posterior mean for $\theta$ that is always 1/2, no matter what the data are.

Some inferences are invariant
In some sense, the index (or label) of a mixture component is irrelevant. Posterior predictive inferences can still be carried out without identifying mixture components. For example, the log probability of a new observation does not depend on the identities of the mixture components. The only sound Bayesian inferences in such models are those that are invariant to label switching. Posterior means for the parameters are meaningless because they are not invariant to label switching; for example, the posterior mean for $\theta$ in the two component mixture model will always be 1/2.

Highly multimodal posteriors
Theoretically, this should not present a problem for inference because all of the integrals involved in posterior predictive inference will be well behaved. The problem in practice is computation.

Being able to carry out such invariant inferences in practice is an altogether different matter. It is almost always intractable to find even a single posterior mode, much less
balance the exploration of the neighborhoods of multiple local maxima according to the probability masses. In Gibbs sampling, it is unlikely for $\mu_1$ to move to a new mode when sampled conditioned on the current values of $\mu_2$ and $\theta$. For HMC and NUTS, the problem is that the sampler gets stuck in one of the two “bowls” around the modes and cannot gather enough energy from random momentum assignment to move from one mode to another.

Even with a proper posterior, all known sampling and inference techniques are notoriously ineffective when the number of modes grows super-exponentially as it does for mixture models with increasing numbers of components.

**Hacks as fixes**
Several hacks (i.e., “tricks”) have been suggested and employed to deal with the problems posed by label switching in practice.

**Parameter ordering constraints**
One common strategy is to impose a constraint on the parameters that identifies the components. For instance, we might consider constraining $\mu_1 < \mu_2$ in the two-component normal mixture model discussed above. A problem that can arise from such an approach is when there is substantial probability mass for the opposite ordering $\mu_1 > \mu_2$. In these cases, the posteriors are affected by the constraint and true posterior uncertainty in $\mu_1$ and $\mu_2$ is not captured by the model with the constraint. In addition, standard approaches to posterior inference for event probabilities is compromised. For instance, attempting to use $M$ posterior samples to estimate $\Pr[\mu_1 > \mu_2]$, will fail, because the estimator

$$\Pr[\mu_1 > \mu_2] \approx \sum_{m=1}^{M} I\left(\mu_1^{(m)} > \mu_2^{(m)}\right)$$

will result in an estimate of 0 because the posterior respects the constraint in the model.

**Initialization around a single mode**
Another common approach is to run a single chain or to initialize the parameters near realistic values.\(^4\)

\(^4\)Tempering methods may be viewed as automated ways to carry out such a search for modes, though most MCMC tempering methods continue to search for modes on an ongoing basis; see (Swendsen and Wang 1986; Neal 1996b).
This can work better than the hard constraint approach if reasonable initial values can be found and the labels do not switch within a Markov chain. The result is that all chains are glued to a neighborhood of a particular mode in the posterior.

### 21.3. Component collapsing in mixture models

It is possible for two mixture components in a mixture model to collapse to the same values during sampling or optimization. For example, a mixture of $K$ normals might devolve to have $\mu_i = \mu_j$ and $\sigma_i = \sigma_j$ for $i \neq j$.

This will typically happen early in sampling due to initialization in MCMC or optimization or arise from random movement during MCMC. Once the parameters match for a given draw $(m)$, it can become hard to escape because there can be a trough of low-density mass between the current parameter values and the ones without collapsed components.

It may help to use a smaller step size during warmup, a stronger prior on each mixture component’s membership responsibility. A more extreme measure is to include additional mixture components to deal with the possibility that some of them may collapse.

In general, it is difficult to recover exactly the right $K$ mixture components in a mixture model as $K$ increases beyond one (yes, even a two-component mixture can have this problem).

### 21.4. Posteriors with unbounded densities

In some cases, the posterior density grows without bounds as parameters approach certain poles or boundaries. In such, there are no posterior modes and numerical stability issues can arise as sampled parameters approach constraint boundaries.

#### Mixture models with varying scales

One such example is a binary mixture model with scales varying by component, $\sigma_1$ and $\sigma_2$ for locations $\mu_1$ and $\mu_2$. In this situation, the density grows without bound as $\sigma_1 \to 0$ and $\mu_1 \to y_n$ for some $n$; that is, one of the mixture components concentrates all of its mass around a single data item $y_n$.

#### Beta-binomial models with skewed data and weak priors

Another example of unbounded densities arises with a posterior such as $\text{beta}(\phi | 0.5, 0.5)$, which can arise if seemingly weak beta priors are used for groups that have no data. This density is unbounded as $\phi \to 0$ and $\phi \to 1$. Similarly, a Bernoulli
likelihood model coupled with a “weak” beta prior, leads to a posterior

\[
p(\phi \mid y) \propto \text{beta}(\phi \mid 0.5, 0.5) \times \prod_{n=1}^{N} \text{Bernoulli}(y_n \mid \phi)
\]

\[
= \text{beta} \left( \phi \bigg| 0.5 + \sum_{n=1}^{N} y_n, 0.5 + N - \sum_{n=1}^{N} y_n \right).
\]

If \( N = 9 \) and each \( y_n = 1 \), the posterior is \( \text{beta}(\phi \mid 9.5, 0.5) \). This posterior is unbounded as \( \phi \to 1 \). Nevertheless, the posterior is proper, and although there is no posterior mode, the posterior mean is well-defined with a value of exactly 0.95.

**Constrained vs. unconstrained scales**

Stan does not sample directly on the constrained \((0, 1)\) space for this problem, so it doesn’t directly deal with unconstrained density values. Rather, the probability values \( \phi \) are logit-transformed to \((-\infty, \infty)\). The boundaries at 0 and 1 are pushed out to \(-\infty\) and \(\infty\) respectively. The Jacobian adjustment that Stan automatically applies ensures the unconstrained density is proper. The adjustment for the particular case of \((0, 1)\) is \(\log \text{logit}^{-1}(\phi) + \log \text{logit}(1 - \phi)\).

There are two problems that still arise, though. The first is that if the posterior mass for \( \phi \) is near one of the boundaries, the logit-transformed parameter will have to sweep out long paths and thus can dominate the U-turn condition imposed by the no-U-turn sampler (NUTS). The second issue is that the inverse transform from the unconstrained space to the constrained space can underflow to 0 or overflow to 1, even when the unconstrained parameter is not infinite. Similar problems arise for the expectation terms in logistic regression, which is why the logit-scale parameterizations of the Bernoulli and binomial distributions are more stable.

**21.5. **Posterior with unbounded parameters**

In some cases, the posterior density will not grow without bound, but parameters will grow without bound with gradually increasing density values. Like the models discussed in the previous section that have densities that grow without bound, such models also have no posterior modes.

**Separability in logistic regression**

Consider a logistic regression model with \( N \) observed outcomes \( y_n \in \{0, 1\} \), an \( N \times K \) matrix \( x \) of predictors, a \( K \)-dimensional coefficient vector \( \beta \), and sampling distribution

\[
y_n \sim \text{Bernoulli}(\text{logit}^{-1}(x_n \beta)).
\]
Now suppose that column $k$ of the predictor matrix is such that $x_{n,k} > 0$ if and only if $y_n = 1$, a condition known as “separability.” In this case, predictive accuracy on the observed data continue to improve as $\beta_k \to \infty$, because for cases with $y_n = 1$, $x_n \beta \to \infty$ and hence $\logit^{-1}(x_n \beta) \to 1$.

With separability, there is no maximum to the likelihood and hence no maximum likelihood estimate. From the Bayesian perspective, the posterior is improper and therefore the marginal posterior mean for $\beta_k$ is also not defined. The usual solution to this problem in Bayesian models is to include a proper prior for $\beta$, which ensures a proper posterior.

### 21.6. Uniform posteriors

Suppose your model includes a parameter $\psi$ that is defined on $[0, 1]$ and is given a flat prior $\text{uniform}(\psi | 0, 1)$. Now if the data don’t tell us anything about $\psi$, the posterior is also $\text{uniform}(\psi | 0, 1)$.

Although there is no maximum likelihood estimate for $\psi$, the posterior is uniform over a closed interval and hence proper. In the case of a uniform posterior on $[0, 1]$, the posterior mean for $\psi$ is well-defined with value $1/2$. Although there is no posterior mode, posterior predictive inference may nevertheless do the right thing by simply integrating (i.e., averaging) over the predictions for $\psi$ at all points in $[0, 1]$.

### 21.7. Sampling difficulties with problematic priors

With an improper posterior, it is theoretically impossible to properly explore the posterior. However, Gibbs sampling as performed by BUGS and JAGS, although still unable to properly sample from such an improper posterior, behaves differently in practice than the Hamiltonian Monte Carlo sampling performed by Stan when faced with an example such as the two intercept model discussed in the collinearity section and illustrated in the non-identifiable density plot.

**Gibbs sampling**

Gibbs sampling, as performed by BUGS and JAGS, may appear to be efficient and well behaved for this unidentified model, but as discussed in the previous subsection, will not actually explore the posterior properly.

Consider what happens with initial values $\lambda_{1}^{(0)}, \lambda_{2}^{(0)}$. Gibbs sampling proceeds in iteration $m$ by drawing

\[
\lambda_{1}^{(m)} \sim p(\lambda_{1} \mid \lambda_{2}^{(m-1)}, \sigma^{(m-1)}, y) \\
\lambda_{2}^{(m)} \sim p(\lambda_{2} \mid \lambda_{1}^{(m)}, \sigma^{(m-1)}, y) \\
\sigma^{(m)} \sim p(\sigma \mid \lambda_{1}^{(m)}, \lambda_{2}^{(m)}, y).
\]
Now consider the draw for $\lambda_1$ (the draw for $\lambda_2$ is symmetric), which is conjugate in this model and thus can be done efficiently. In this model, the range from which the next $\lambda_1$ can be drawn is highly constrained by the current values of $\lambda_2$ and $\sigma$. Gibbs will run quickly and provide seemingly reasonable inferences for $\lambda_1 + \lambda_2$. But it will not explore the full range of the posterior; it will merely take a slow random walk from the initial values. This random walk behavior is typical of Gibbs sampling when posteriors are highly correlated and the primary reason to prefer Hamiltonian Monte Carlo to Gibbs sampling for models with parameters correlated in the posterior.

**Hamiltonian Monte Carlo sampling**

Hamiltonian Monte Carlo (HMC), as performed by Stan, is much more efficient at exploring posteriors in models where parameters are correlated in the posterior. In this particular example, the Hamiltonian dynamics (i.e., the motion of a fictitious particle given random momentum in the field defined by the negative log posterior) is going to run up and down along the valley defined by the potential energy (ridges in log posteriors correspond to valleys in potential energy). In practice, even with a random momentum for $\lambda_1$ and $\lambda_2$, the gradient of the log posterior is going to adjust for the correlation and the simulation will run $\lambda_1$ and $\lambda_2$ in opposite directions along the valley corresponding to the ridge in the posterior log density.

**No-U-turn sampling**

Stan’s default no-U-turn sampler (NUTS), is even more efficient at exploring the posterior (see Hoffman and Gelman 2014). NUTS simulates the motion of the fictitious particle representing the parameter values until it makes a U-turn, it will be defeated in most cases, as it will just move down the potential energy valley indefinitely without making a U-turn. What happens in practice is that the maximum number of leapfrog steps in the simulation will be hit in many of the iterations, causing a large number of log probability and gradient evaluations (1000 if the max tree depth is set to 10, as in the default). Thus sampling will appear to be slow. This is indicative of an improper posterior, not a bug in the NUTS algorithm or its implementation. It is simply not possible to sample from an improper posterior! Thus the behavior of HMC in general and NUTS in particular should be reassuring in that it will clearly fail in cases of improper posteriors, resulting in a clean diagnostic of sweeping out large paths in the posterior.

Here are results of Stan runs with default parameters fit to $N = 100$ data points generated from $y_n \sim \text{normal}(0, 1)$:

*Two Scale Parameters, Improper Prior*

Inference for Stan model: improper_stan

Warmup took (2.7, 2.6, 2.9, 2.9) seconds, 11 seconds total
CHAPTER 21. PROBLEMATIC POSTERIORS

Sampling took (3.4, 3.7, 3.6, 3.4) seconds, 14 seconds total

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Two Scale Parameters, Weak Prior

Warmup took (0.40, 0.44, 0.40, 0.36) seconds, 1.6 seconds total
Sampling took (0.47, 0.40, 0.47, 0.39) seconds, 1.7 seconds total

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One Scale Parameter, Improper Prior

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Sampling took (0.017, 0.020, 0.020, 0.019) seconds, 0.077 seconds total

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On the top is the non-identified model with improper uniform priors and likelihood
\[ y_n \sim \text{normal}(\lambda_1 + \lambda_2, \sigma). \]

In the middle is the same likelihood as the middle plus priors \( \lambda_k \sim \text{normal}(0, 10). \)

On the bottom is an identified model with an improper prior, with likelihood \( y_n \sim \text{normal}(\mu, \sigma). \) All models estimate \( \mu \) at roughly 0.16 with low Monte Carlo standard error, but a high posterior standard deviation of 0.1; the true value \( \mu = 0 \) is within the 90\% posterior intervals in all three models.
Examples: fits in Stan
To illustrate the issues with sampling from non-identified and only weakly identified models, we fit three models with increasing degrees of identification of their parameters. The posteriors for these models is illustrated in the non-identifiable density plot. The first model is the unidentified model with two location parameters and no priors discussed in the collinearity section.

```stan
data {
  int N;
  real y[N];
}
parameters {
  real lambda1;
  real lambda2;
  real<lower=0> sigma;
}
transformed parameters {
  real mu;
  mu = lambda1 + lambda2;
}
model {
  y ~ normal(mu, sigma);
}
```

The second adds priors to the model block for \( \lambda_1 \) and \( \lambda_2 \) to the previous model.

\[
\lambda_1 \sim \text{normal}(0, 10);
\lambda_2 \sim \text{normal}(0, 10);
\]

The third involves a single location parameter, but no priors.

```stan
data {
  int N;
  real y[N];
}
parameters {
  real mu;
  real<lower=0> sigma;
}
model {
  y ~ normal(mu, sigma);
}
```
All three of the example models were fit in Stan 2.1.0 with default parameters (1000 warmup iterations, 1000 sampling iterations, NUTS sampler with max tree depth of 10). The results are shown in the non-identified fits figure. The key statistics from these outputs are the following.

- As indicated by R_hat column, all parameters have converged other than $\lambda_1$ and $\lambda_2$ in the non-identified model.
- The average number of leapfrog steps is roughly 3 in the identified model, 150 in the model identified by a weak prior, and 1400 in the non-identified model.
- The number of effective samples per second for $\mu$ is roughly 31,000 in the identified model, 1,900 in the model identified with weakly informative priors, and 200 in the non-identified model; the results are similar for $\sigma$.
- In the non-identified model, the 95% interval for $\lambda_1$ is (-2300,6000), whereas it is only (-12,12) in the model identified with weakly informative priors.
- In all three models, the simulated value of $\mu = 0$ and $\sigma = 1$ are well within the posterior 90% intervals.

The first two points, lack of convergence and hitting the maximum number of leapfrog steps (equivalently maximum tree depth) are indicative of improper posteriors. Thus rather than covering up the problem with poor sampling as may be done with Gibbs samplers, Hamiltonian Monte Carlo tries to explore the posterior and its failure is a clear indication that something is amiss in the model.
22. Reparameterization and Change of Variables

Stan supports a direct encoding of reparameterizations. Stan also supports changes of variables by directly incrementing the log probability accumulator with the log Jacobian of the transform.

22.1. Theoretical and practical background
A Bayesian posterior is technically a probability measure, which is a parameterization-invariant, abstract mathematical object.\(^1\)

Stan’s modeling language, on the other hand, defines a probability density, which is a non-unique, parameterization-dependent function in \(\mathbb{R}^N \rightarrow \mathbb{R}^+\). In practice, this means a given model can be represented different ways in Stan, and different representations have different computational performances.

As pointed out by Andrew Gelman (2004) in a paper discussing the relation between parameterizations and Bayesian modeling, a change of parameterization often carries with it suggestions of how the model might change, because we tend to use certain natural classes of prior distributions. Thus, it’s not just that we have a fixed distribution that we want to sample from, with reparameterizations being computational aids. In addition, once we reparameterize and add prior information, the model itself typically changes, often in useful ways.

22.2. Reparameterizations
Reparameterizations may be implemented directly using the transformed parameters block or just in the model block.

**Beta and Dirichlet priors**
The beta and Dirichlet distributions may both be reparameterized from a vector of counts to use a mean and total count.

**Beta distribution**
For example, the Beta distribution is parameterized by two positive count parameters \(\alpha, \beta > 0\). The following example illustrates a hierarchical Stan model with a vector

\(^1\)This is in contrast to (penalized) maximum likelihood estimates, which are not parameterization invariant.
of parameters $\theta$ are drawn i.i.d. for a Beta distribution whose parameters are themselves drawn from a hyperprior distribution.

```plaintext
parameters {
    real<lower = 0> alpha;
    real<lower = 0> beta;
    ...
}
model {
    alpha ~ ...
    beta ~ ...
    for (n in 1:N)
    theta[n] ~ beta(alpha, beta);
    ...
}
```

It is often more natural to specify hyperpriors in terms of transformed parameters. In the case of the Beta, the obvious choice for reparameterization is in terms of a mean parameter

$$\phi = \frac{\alpha}{\alpha + \beta}$$

and total count parameter

$$\lambda = \alpha + \beta.$$

Following @GelmanEtAl:2013, Chapter 5] the mean gets a uniform prior and the count parameter a Pareto prior with $p(\lambda) \propto \lambda^{-2.5}$.

```plaintext
parameters {
    real<lower=0,upper=1> phi;
    real<lower=0.1> lambda;
    ...
}
transformed parameters {
    real<lower=0> alpha = lambda * phi;
    real<lower=0> beta = lambda * (1 - phi);
    ...
}
model {
    phi ~ beta(1, 1); // uniform on phi, could drop
    lambda ~ pareto(0.1, 1.5);
    for (n in 1:N)
    theta[n] ~ beta(alpha, beta);
    ...
}
```

The new parameters, $\phi$ and $\lambda$, are declared in the parameters block and the parameters for the Beta distribution, $\alpha$ and $\beta$, are declared and defined in
the transformed parameters block. And if their values are not of interest, they could instead be defined as local variables in the model as follows.

    model {
        real alpha = lambda * phi
        real beta = lambda * (1 - phi);
        ...
        for (n in 1:N)
            theta[n] ~ beta(alpha, beta);
        ...
    }

With vectorization, this could be expressed more compactly and efficiently as follows.

    model {
        theta ~ beta(lambda * phi, lambda * (1 - phi));
        ...
    }

If the variables alpha and beta are of interest, they can be defined in the transformed parameter block and then used in the model.

*Jacobians not necessary*  

Because the transformed parameters are being used, rather than given a distribution, there is no need to apply a Jacobian adjustment for the transform. For example, in the beta distribution example, alpha and beta have the correct posterior distribution.

*Dirichlet priors*  

The same thing can be done with a Dirichlet, replacing the mean for the Beta, which is a probability value, with a simplex. Assume there are $K > 0$ dimensions being considered ($K = 1$ is trivial and $K = 2$ reduces to the beta distribution case). The traditional prior is

    parameters {
        vector[K] alpha;
        simplex[K] theta[N];
        ...
    }

    model {
        alpha ~ ...;
        for (n in 1:N)
            theta[n] ~ dirichlet(alpha);
This provides essentially $K$ degrees of freedom, one for each dimension of $alpha$, and it is not obvious how to specify a reasonable prior for $alpha$.

An alternative coding is to use the mean, which is a simplex, and a total count.

```stan
parameters {
  simplex[K] phi;
  real<lower=0> kappa;
  simplex[K] theta[N];
  ...
}
transformed parameters {
  vector[K] alpha = kappa * phi;
  ...
}
model {
  phi ~ ...;
  kappa ~ ...;
  for (n in 1:N)
    theta[n] ~ dirichlet(alpha);
}
```

Now it is much easier to formulate priors, because $phi$ is the expected value of $theta$ and $kappa$ (minus $K$) is the strength of the prior mean measured in number of prior observations.

**Transforming unconstrained priors: probit and logit**

If the variable $u$ has a $\text{uniform}(0,1)$ distribution, then $\text{logit}(u)$ is distributed as $\text{logistic}(0,1)$. This is because inverse logit is the cumulative distribution function (cdf) for the logistic distribution, so that the logit function itself is the inverse CDF and thus maps a uniform draw in $(0,1)$ to a logistically-distributed quantity.

Things work the same way for the probit case: if $u$ has a $\text{uniform}(0,1)$ distribution, then $\Phi^{-1}(u)$ has a $\text{normal}(0,1)$ distribution. The other way around, if $v$ has a $\text{normal}(0,1)$ distribution, then $\Phi(v)$ has a $\text{uniform}(0,1)$ distribution.

In order to use the probit and logistic as priors on variables constrained to $(0,1)$, create an unconstrained variable and transform it appropriately. For comparison, the following Stan program fragment declares a $(0,1)$-constrained parameter $theta$ and gives it a beta prior, then uses it as a parameter in a distribution (here using `foo` as a placeholder).

```stan
parameters {
  ...
}
```
real<lower = 0, upper = 1> theta;
...
model {
    theta ~ beta(a, b);
    ...
    y ~ foo(theta);
    ...
}

If the variables $a$ and $b$ are one, then this imposes a uniform distribution $\theta$. If $a$ and $b$ are both less than one, then the density on $\theta$ has a U shape, whereas if they are both greater than one, the density of $\theta$ has an inverted-U or more bell-like shape.

Roughly the same result can be achieved with unbounded parameters that are probit or inverse-logit-transformed. For example,

parameters {
    real theta_raw;
    ...
}
transformed parameters {
    real<lower = 0, upper = 1> theta = inv_logit(theta_raw);
    ...
}
model {
    theta_raw ~ logistic(mu, sigma);
    ...
    y ~ foo(theta);
    ...
}

In this model, an unconstrained parameter $\theta_{raw}$ gets a logistic prior, and then the transformed parameter $\theta$ is defined to be the inverse logit of $\theta_{raw}$. In this parameterization, $\text{inv\_logit}(\mu)$ is the mean of the implied prior on $\theta$. The prior distribution on $\theta$ will be flat if $\sigma$ is one and $\mu$ is zero, and will be U-shaped if $\sigma$ is larger than one and bell shaped if $\sigma$ is less than one.

When moving from a variable in $(0, 1)$ to a simplex, the same trick may be performed using the softmax function, which is a multinomial generalization of the inverse logit function. First, consider a simplex parameter with a Dirichlet prior.

parameters {
    simplex[K] theta;
    ...
}
model {

theta ~ dirichlet(a);
...
y ~ foo(theta);

Now a is a vector with K rows, but it has the same shape properties as the pair a and b for a beta; the beta distribution is just the distribution of the first component of a Dirichlet with parameter vector \([ab]^\top\). To formulate an unconstrained prior, the exact same strategy works as for the beta.

parameters {
  vector[K] theta_raw;
  ...
}

transformed parameters {
  simplex[K] theta = softmax(theta_raw);
  ...
}

model {
  theta_raw ~ multi_normal_cholesky(mu, L_Sigma);
}

The multivariate normal is used for convenience and efficiency with its Cholesky-factor parameterization. Now the mean is controlled by \(\text{softmax}(\mu)\), but we have additional control of covariance through \(L_{\Sigma}\) at the expense of having on the order of \(K^2\) parameters in the prior rather than order \(K\). If no covariance is desired, the number of parameters can be reduced back to \(K\) using a vectorized normal distribution as follows.

theta_raw ~ normal(mu, sigma);

where either or both of \(\mu\) and \(\sigma\) can be vectors.

22.3. Changes of variables

Changes of variables are applied when the transformation of a parameter is characterized by a distribution. The standard textbook example is the lognormal distribution, which is the distribution of a variable \(y > 0\) whose logarithm \(\log y\) has a normal distribution. The distribution is being assigned to \(\log y\).

The change of variables requires an adjustment to the probability to account for the distortion caused by the transform. For this to work, univariate changes of variables must be monotonic and differentiable everywhere in their support. Multivariate changes of variables must be injective and differentiable everywhere in their support, and they must map \(\mathbb{R}^N \to \mathbb{R}^N\).

The probability must be scaled by a Jacobian adjustment equal to the absolute determinant of the Jacobian of the transform. In the univariate case, the Jacobian
adjustment is simply the absolute derivative of the transform.

In the case of log normals, if \( y \)'s logarithm is normal with mean \( \mu \) and deviation \( \sigma \), then the distribution of \( y \) is given by

\[
p(y) = \text{normal}(\log y \mid \mu, \sigma) \left| \frac{d}{dy} \log y \right| = \text{normal}(\log y \mid \mu, \sigma) \frac{1}{y}.
\]

Stan works on the log scale to prevent underflow, where

\[
\log p(y) = \log \text{normal}(\log y \mid \mu, \sigma) - \log y.
\]

In Stan, the change of variables can be applied in the sampling statement. To adjust for the curvature, the log probability accumulator is incremented with the log absolute derivative of the transform. The lognormal distribution can thus be implemented directly in Stan as follows.\(^2\)

```stan
parameters {
  real<lower=0> y;
  ...
}
model {
  log(y) ~ normal(mu, sigma);
  target += -log(y);
  ...
}
```

It is important, as always, to declare appropriate constraints on parameters; here \( y \) is constrained to be positive.

It would be slightly more efficient to define a local variable for the logarithm, as follows.

```stan
model {
  real log_y;
  log_y = log(y);
  log_y ~ normal(mu, sigma);
  target += -log_y;
  ...
}
```

If \( y \) were declared as data instead of as a parameter, then the adjustment can be ignored because the data will be constant and Stan only requires the log probability up to a constant.

---

\(^2\)This example is for illustrative purposes only; the recommended way to implement the lognormal distribution in Stan is with the built-in `lognormal` probability function; see the functions reference manual for details.
Change of variables vs. transformations
This section illustrates the difference between a change of variables and a simple variable transformation. A transformation samples a parameter, then transforms it, whereas a change of variables transforms a parameter, then samples it. Only the latter requires a Jacobian adjustment.

It does not matter whether the probability function is expressed using a sampling statement, such as
\[
\log(y) \sim \text{normal}(\mu, \sigma);
\]
or as an increment to the log probability function, as in
\[
\text{target} += \text{normal}_\text{lpdf}(\log(y) | \mu, \sigma);
\]

Gamma and inverse gamma distribution
Like the log normal, the inverse gamma distribution is a distribution of variables whose inverse has a gamma distribution. This section contrasts two approaches, first with a transform, then with a change of variables.

The transform based approach to sampling $y_{\text{inv}}$ with an inverse gamma distribution can be coded as follows.

```plaintext
parameters {
  real<lower=0> y;
}
transformed parameters {
  real<lower=0> y_inv;
  y_inv = 1 / y;
}
model {
  y ~ gamma(2,4);
}
```

The change-of-variables approach to sampling $y_{\text{inv}}$ with an inverse gamma distribution can be coded as follows.

```plaintext
parameters {
  real<lower=0> y_inv;
}
transformed parameters {
  real<lower=0> y;
  y = 1 / y_inv;  // change variables
```

The Jacobian adjustment is the log of the absolute derivative of the transform, which in this case is

$$\log \left| \frac{d}{du} \left( \frac{1}{u} \right) \right| = \log |-u^{-2}| = \log u^{-2} = -2 \log u.$$ 

Multivariate changes of variables

In the case of a multivariate transform, the log of the absolute determinant of the Jacobian of the transform must be added to the log probability accumulator. In Stan, this can be coded as follows in the general case where the Jacobian is not a full matrix.

```stan
parameters {
  vector[K] u; // multivariate parameter
  ...
}

transformed parameters {
  vector[K] v; // transformed parameter
  matrix[K, K] J; // Jacobian matrix of transform
  ... compute v as a function of u ...
  ... compute J[m, n] = d.v[m] / d.u[n] ...
  target += log(fabs(determinant(J)));
  ...
}

model {
  v ~ ...;
  ...
}
```

If the determinant of the Jacobian is known analytically, it will be more efficient to apply it directly than to call the determinant function, which is neither efficient nor particularly stable numerically.

In many cases, the Jacobian matrix will be triangular, so that only the diagonal elements will be required for the determinant calculation. Triangular Jacobians arise when each element $v[k]$ of the transformed parameter vector only depends on elements $u[1], \ldots, u[k]$ of the parameter vector. For triangular matrices, the determinant is the product of the diagonal elements, so the transformed parameters
block of the above model can be simplified and made more efficient by recoding as follows.

```
transformed parameters {
  ...
  vector[K] J_diag; // diagonals of Jacobian matrix
  ...
  ... compute J[k, k] = d.v[k] / d.u[k] ...
  target += sum(log(J_diag));
  ...}
```

### 22.4. Vectors with varying bounds
Stan only allows a single lower and upper bound to be declared in the constraints for a container data type. But suppose we have a vector of parameters and a vector of lower bounds? Then the transforms are calculated and their log Jacobians added to the log density accumulator; the Jacobian calculations are described in detail in the reference manual chapter on constrained parameter transforms.

**Varying lower bounds**
For example, suppose there is a vector parameter $\alpha$ with a vector $L$ of lower bounds. The simplest way to deal with this if $L$ is a constant is to shift a lower-bounded parameter.

```
data {
  int N;
  vector[N] L; // lower bounds
  ...
parameters {
  vector<lower=0>[N] alpha_raw;
  ...
transformed parameters {
  vector[N] alpha = L + alpha_raw;
  ...
```

The Jacobian for adding a constant is one, so its log drops out of the log density. Even if the lower bound is a parameter rather than data, there is no Jacobian required, because the transform from $(L, \alpha_{\text{raw}})$ to $(L + \alpha_{\text{raw}}, \alpha_{\text{raw}})$ produces a Jacobian derivative matrix with a unit determinant.

It's also possible implement the transform by directly transforming an unconstrained parameter and accounting for the Jacobian.
data {
  int N;
  vector[N] L; // lower bounds
  ...}

parameters {
  vector[N] alpha_raw;
  ...}

transformed parameters {
  vector[N] alpha = L + exp(alpha_raw);
  ...}

model {
  target += sum(alpha_raw); // log Jacobian
  ...}

The adjustment in the log Jacobian determinant of the transform mapping \( \alpha_{\text{raw}} \) to \( \alpha = L + \exp(\alpha_{\text{raw}}) \). The details are simple in this case because the Jacobian is diagonal; see the reference manual chapter on constrained parameter transforms for full details. Here \( L \) can even be a vector containing parameters that don’t depend on \( \alpha_{\text{raw}} \); if the bounds do depend on \( \alpha_{\text{raw}} \) then a revised Jacobian needs to be calculated taking into account the dependencies.

Varying upper and lower bounds
Suppose there are lower and upper bounds that vary by parameter. These can be applied to shift and rescale a parameter constrained to \((0, 1)\).

data {
  int N;
  vector[N] L; // lower bounds
  vector[N] U; // upper bounds
  ...}

parameters {
  vector<lower=0, upper=1>[N] alpha_raw;
  ...}

transformed parameters {
  vector[N] alpha = L + (U - L) .* alpha_raw;
  ...}

The expression \( U - L \) is multiplied by \( \alpha_{\text{raw}} \) elementwise to produce a vector of variables in \((0, U - L)\), then adding \( L \) results in a variable ranging between \((L, U)\).

In this case, it is important that \( L \) and \( U \) are constants, otherwise a Jacobian would be required when multiplying by \( U - L \).
23. Efficiency Tuning

This chapter provides a grab bag of techniques for optimizing Stan code, including vectorization, sufficient statistics, and conjugacy. At a coarse level, efficiency involves both the amount of time required for a computation and the amount of memory required. For practical applied statistical modeling, we are mainly concerned with reducing wall time (how long a program takes as measured by a clock on the wall) and keeping memory requirements within available bounds.

23.1. What is efficiency?
The standard algorithm analyses in computer science measure efficiency asymptotically as a function of problem size (such as data, number of parameters, etc.) and typically do not consider constant additive factors like startup times or multiplicative factors like speed of operations. In practice, the constant factors are important; if run time can be cut in half or more, that’s a huge gain. This chapter focuses on both the constant factors involved in efficiency (such as using built-in matrix operations as opposed to naive loops) and on asymptotic efficiency factors (such as using linear algorithms instead of quadratic algorithms in loops).

23.2. Efficiency for probabilistic models and algorithms
Stan programs express models which are intrinsically statistical in nature. The algorithms applied to these models may or may not themselves be probabilistic. For example, given an initial value for parameters (which may itself be given deterministically or generated randomly), Stan’s optimization algorithm (L-BFGS) for penalized maximum likelihood estimation is purely deterministic. Stan’s sampling algorithms are based on Markov chain Monte Carlo algorithms, which are probabilistic by nature at every step. Stan’s variational inference algorithm (ADVI) is probabilistic despite being an optimization algorithm; the randomization lies in a nested Monte Carlo calculation for an expected gradient.

With probabilistic algorithms, there will be variation in run times (and maybe memory usage) based on the randomization involved. For example, by starting too far out in the tail, iterative algorithms underneath the hood, such as the solvers for ordinary differential equations, may take different numbers of steps. Ideally this variation will be limited; when there is a lot of variation it can be a sign that there is a problem with the model’s parameterization in a Stan program or with initialization.
A well-behaved Stan program will have low variance between runs with different random initializations and differently seeded random number generators. But sometimes an algorithm can get stuck in one part of the posterior, typically due to high curvature. Such sticking almost always indicates the need to reparameterize the model. Just throwing away Markov chains with apparently poor behavior (slow, or stuck) can lead to bias in posterior estimates. This problem with getting stuck can often be overcome by lowering the initial step size to avoid getting stuck during adaptation and increasing the target acceptance rate in order to target a lower step size. This is because smaller step sizes allow Stan’s gradient-based algorithms to better follow the curvature in the density or penalized maximum likelihood being fit.

23.3. Statistical vs. computational efficiency
There is a difference between pure computational efficiency and statistical efficiency for Stan programs fit with sampling-based algorithms. Computational efficiency measures the amount of time or memory required for a given step in a calculation, such as an evaluation of a log posterior or penalized likelihood.

Statistical efficiency typically involves requiring fewer steps in algorithms by making the statistical formulation of a model better behaved. The typical way to do this is by applying a change of variables (i.e., reparameterization) so that sampling algorithms mix better or optimization algorithms require less adaptation.

23.4. Model conditioning and curvature
Because Stan’s algorithms (other than Riemannian Hamiltonian Monte Carlo) rely on step-based gradient-based approximations of the density (or penalized maximum likelihood) being fitted, posterior curvature not captured by this first-order approximation plays a central role in determining the statistical efficiency of Stan’s algorithms.

A second-order approximation to curvature is provided by the Hessian, the matrix of second derivatives of the log density \( \log p(\theta) \) with respect to the parameter vector \( \theta \), defined as

\[
H(\theta) = \nabla \nabla \log p(\theta \mid y),
\]

so that

\[
H_{i,j}(\theta) = \frac{\partial^2 \log p(\theta \mid y)}{\partial \theta_i \partial \theta_j}.
\]

For pure penalized maximum likelihood problems, the posterior log density \( \log p(\theta \mid y) \) is replaced by the penalized likelihood function \( \mathcal{L}(\theta) = \log p(y \mid \theta) - \lambda(\theta) \).
Condition number and adaptation
A good gauge of how difficult a problem the curvature presents is given by the condition number of the Hessian matrix $H$, which is the ratio of the largest to the smallest eigenvalue of $H$ (assuming the Hessian is positive definite). This essentially measures the difference between the flattest direction of movement and the most curved. Typically, the step size of a gradient-based algorithm is bounded by the most sharply curved direction. With better conditioned log densities or penalized likelihood functions, it is easier for Stan’s adaptation, especially the diagonal adaptations that are used as defaults.

Unit scales without correlation
Ideally, all parameters should be programmed so that they have unit scale and so that posterior correlation is reduced; together, these properties mean that there is no rotation or scaling required for optimal performance of Stan’s algorithms. For Hamiltonian Monte Carlo, this implies a unit mass matrix, which requires no adaptation as it is where the algorithm initializes. Riemannian Hamiltonian Monte Carlo performs this conditioning on the fly at every step, but such conditioning is expensive computationally.

Varying curvature
In all but very simple models (such as multivariate normals), the Hessian will vary as $\theta$ varies (an extreme example is Neal’s funnel, as naturally arises in hierarchical models with little or no data). The more the curvature varies, the harder it is for all of the algorithms with fixed adaptation parameters (that is, everything but Riemannian Hamiltonian Monte Carlo) to find adaptations that cover the entire density well. Many of the variable transforms proposed are aimed at improving the conditioning of the Hessian and/or making it more consistent across the relevant portions of the density (or penalized maximum likelihood function) being fit.

For all of Stan’s algorithms, the curvature along the path from the initial values of the parameters to the solution is relevant. For penalized maximum likelihood and variational inference, the solution of the iterative algorithm will be a single point, so this is all that matters. For sampling, the relevant “solution” is the typical set, which is the posterior volume where almost all draws from the posterior lies; thus, the typical set contains almost all of the posterior probability mass.

With sampling, the curvature may vary dramatically between the points on the path from the initialization point to the typical set and within the typical set. This is why adaptation needs to run long enough to visit enough points in the typical set to get a good first-order estimate of the curvature within the typical set. If adaptation is not run long enough, sampling within the typical set after adaptation will not
be efficient. We generally recommend at least one hundred iterations after the
typical set is reached (and the first effective draw is ready to be realized). Whether
adaptation has run long enough can be measured by comparing the adaptation
parameters derived from a set of diffuse initial parameter values.

**Reparameterizing with a change of variables**
Improving statistical efficiency is achieved by reparameterizing the model so that
the same result may be calculated using a density or penalized maximum likelihood
that is better conditioned. Again, see the example of reparameterizing Neal’s funnel
for an example, and also the examples in the change of variables chapter.

One has to be careful in using change-of-variables reparameterizations when using
maximum likelihood estimation, because they can change the result if the Jacobian
term is inadvertently included in the revised likelihood model.

### 23.5. Well-specified models
Model misspecification, which roughly speaking means using a model that doesn’t
match the data, can be a major source of slow code. This can be seen in cases
where simulated data according to the model runs robustly and efficiently, whereas
the real data for which it was intended runs slowly or may even have convergence
and mixing issues. While some of the techniques recommended in the remaining
sections of this chapter may mitigate the problem, the best remedy is a better model
specification.

Counterintuitively, more complicated models often run faster than simpler models.
One common pattern is with a group of parameters with a wide fixed prior such as $\text{normal}(0, 1000)$). This can fit slowly due to the mismatch between prior and
posterior (the prior has support for values in the hundreds or even thousands,
whereas the posterior may be concentrated near zero). In such cases, replacing the
fixed prior with a hierarchical prior such as $\text{normal}(\mu, \sigma)$, where $\mu$ and $\sigma$
are new parameters with their own hyperpriors, can be beneficial.

### 23.6. Avoiding validation
Stan validates all of its data structure constraints. For example, consider a trans-
formed parameter defined to be a covariance matrix and then used as a covariance
parameter in the model block.

```stan
transformed parameters {
  cov_matrix[K] Sigma;
  ...
} // first validation
model {
```
y ~ multi_normal(mu, Sigma); // second validation

Because Sigma is declared to be a covariance matrix, it will be factored at the end of the transformed parameter block to ensure that it is positive definite. The multivariate normal log density function also validates that Sigma is positive definite. This test is expensive, having cubic run time (i.e., $O(N^3)$ for $N \times N$ matrices), so it should not be done twice.

The test may be avoided by simply declaring Sigma to be a simple unconstrained matrix.

transformed parameters {
  matrix[K, K] Sigma;
  ...
model {
  y ~ multi_normal(mu, Sigma); // only validation
}

Now the only validation is carried out by the multivariate normal.

### 23.7. Reparameterization

Stan’s sampler can be slow in sampling from distributions with difficult posterior geometries. One way to speed up such models is through reparameterization. In some cases, reparameterization can dramatically increase effective sample size for the same number of iterations or even make programs that would not converge well behaved.

**Example: Neal’s funnel**

In this section, we discuss a general transform from a centered to a non-centered parameterization (Papaspiliopoulos, Roberts, and Sköld 2007).

This reparameterization is helpful when there is not much data, because it separates the hierarchical parameters and lower-level parameters in the prior.

Neal (2003) defines a distribution that exemplifies the difficulties of sampling from some hierarchical models. Neal’s example is fairly extreme, but can be trivially reparameterized in such a way as to make sampling straightforward. Neal’s example has support for $y \in \mathbb{R}$ and $x \in \mathbb{R}^9$ with density

---

1. This parameterization came to be known on our mailing lists as the “Matt trick” after Matt Hoffman, who independently came up with it while fitting hierarchical models in Stan.
The probability contours are shaped like ten-dimensional funnels. The funnel’s neck is particularly sharp because of the exponential function applied to \( y \). A plot of the log marginal density of \( y \) and the first dimension \( x_1 \) is shown in the following plot.

The marginal density of Neal’s funnel for the upper-level variable \( y \) and one lower-level variable \( x_1 \) (see the text for the formula). The blue region has log density greater than -8, the yellow region density greater than -16, and the gray background a density less than -16.

![Funnel Density (log scale)](image)

Figure 23.1: Neal’s funnel density
The funnel can be implemented directly in Stan as follows.

```stan
parameters {
  real y;
  vector[9] x;
}
model {
  y ~ normal(0, 3);
  x ~ normal(0, exp(y/2));
}
```

When the model is expressed this way, Stan has trouble sampling from the neck of the funnel, where \( y \) is small and thus \( x \) is constrained to be near 0. This is due to the fact that the density’s scale changes with \( y \), so that a step size that works well in the body will be too large for the neck, and a step size that works in the neck will be inefficient in the body. This can be seen in the following plot.

4000 draws are taken from a run of Stan’s sampler with default settings. Both plots are restricted to the shown window of \( x_1 \) and \( y \) values; some draws fell outside of the displayed area as would be expected given the density. The samples are consistent with the marginal density \( p(y) = \text{normal}(y \mid 0, 3) \), which has mean 0 and standard deviation 3.
In this particular instance, because the analytic form of the density from which samples are drawn is known, the model can be converted to the following more efficient form.

```plaintext
parameters {
  real y_raw;
  vector[9] x_raw;
}
transformed parameters {
  real y;
```
vector[9] x;

y = 3.0 * y_raw;
x = exp(y/2) * x_raw;
}

model {
  y_raw ~ std_normal(); // implies y ~ normal(0, 3)
x_raw ~ std_normal(); // implies x ~ normal(0, exp(y/2))
}

In this second model, the parameters x_raw and y_raw are sampled as independent standard normals, which is easy for Stan. These are then transformed into samples from the funnel. In this case, the same transform may be used to define Monte Carlo samples directly based on independent standard normal samples; Markov chain Monte Carlo methods are not necessary. If such a reparameterization were used in Stan code, it is useful to provide a comment indicating what the distribution for the parameter implies for the distribution of the transformed parameter.

**Reparameterizing the Cauchy**

Sampling from heavy tailed distributions such as the Cauchy is difficult for Hamiltonian Monte Carlo, which operates within a Euclidean geometry.\(^2\)

The practical problem is that tail of the Cauchy requires a relatively large step size compared to the trunk. With a small step size, the No-U-Turn sampler requires many steps when starting in the tail of the distribution; with a large step size, there will be too much rejection in the central portion of the distribution. This problem may be mitigated by defining the Cauchy-distributed variable as the transform of a uniformly distributed variable using the Cauchy inverse cumulative distribution function.

Suppose a random variable of interest \(X\) has a Cauchy distribution with location \(\mu\) and scale \(\tau\), so that \(X \sim \text{Cauchy}(\mu, \tau)\). The variable \(X\) has a cumulative distribution function \(F_X : \mathbb{R} \to (0, 1)\) defined by

\[
F_X(x) = \frac{1}{\pi} \arctan \left( \frac{x - \mu}{\tau} \right) + \frac{1}{2}.
\]

The inverse of the cumulative distribution function, \(F_X^{-1} : (0, 1) \to \mathbb{R}\), is thus

\(^2\)Riemannian Manifold Hamiltonian Monte Carlo (RMHMC) overcomes this difficulty by simulating the Hamiltonian dynamics in a space with a position-dependent metric; see Girolami and Calderhead (2011) and Betancourt (2012).
Thus if the random variable $Y$ has a unit uniform distribution, $Y \sim \text{uniform}(0, 1)$, then $F_X^{-1}(Y)$ has a Cauchy distribution with location $\mu$ and scale $\tau$, i.e., $F_X^{-1}(Y) \sim \text{Cauchy}(\mu, \tau)$.

Consider a Stan program involving a Cauchy-distributed parameter $\beta$.

```stan
parameters {
  real beta;
  ...
}
model {
  beta ~ cauchy(mu, tau);
  ...
}
```

This declaration of $\beta$ as a parameter may be replaced with a transformed parameter $\beta$ defined in terms of a uniform-distributed parameter $\beta_{\text{unif}}$.

```stan
parameters {
  real<lower = -pi()/2, upper = pi()/2> beta_unif;
  ...
} transformed parameters {
  real beta;
  beta = mu + tau * tan(beta_unif);  // beta ~ cauchy(mu, tau)
} model {
  beta_unif ~ uniform(-pi()/2, pi()/2);  // not necessary
  ...
}
```

It is more convenient in Stan to transform a uniform variable on $(-\pi/2, \pi/2)$ than one on $(0, 1)$. The Cauchy location and scale parameters, $\mu$ and $\tau$, may be defined as data or may themselves be parameters. The variable $\beta$ could also be defined as a local variable if it does not need to be included in the sampler’s output.

The uniform distribution on $\beta_{\text{unif}}$ is defined explicitly in the model block, but it could be safely removed from the program without changing sampling behavior. This is because $\log \text{uniform}(\beta_{\text{unif}} | -\pi/2, \pi/2) = -\log \pi$ is a constant and Stan only
needs the total log probability up to an additive constant. Stan will spend some time checking that that beta_unif is between \(-\pi()/2\) and \(\pi()/2\), but this condition is guaranteed by the constraints in the declaration of beta_unif.

**Reparameterizing a Student-t distribution**

One thing that sometimes works when you’re having trouble with the heavy-tailedness of Student-t distributions is to use the gamma-mixture representation, which says that you can generate a Student-t distributed variable \(\beta\),

\[
\beta \sim \text{Student-t}(\nu, 0, 1),
\]

by first generating a gamma-distributed precision (inverse variance) \(\tau\) according to

\[
\tau \sim \text{Gamma}(\nu/2, \nu/2),
\]

and then generating \(\beta\) from the normal distribution,

\[
\beta \sim \text{normal}(0, \tau^{-\frac{1}{2}}).
\]

Because \(\tau\) is precision, \(\tau^{-\frac{1}{2}}\) is the scale (standard deviation), which is the parameterization used by Stan.

The marginal distribution of \(\beta\) when you integrate out \(\tau\) is Student-t(\(\nu, 0, 1\), i.e.,

\[
\text{Student-t}(\beta \mid \nu, 0, 1) = \int_0^\infty \text{normal}(\beta \mid 0, \tau^{-0.5}) \times \text{Gamma}(\tau \mid \nu/2, \nu/2) \ d\tau.
\]

To go one step further, instead of defining a \(\beta\) drawn from a normal with precision \(\tau\), define \(\alpha\) to be drawn from a unit normal,

\[
\alpha \sim \text{normal}(0, 1)
\]

and rescale by defining

\[
\beta = \alpha \tau^{-\frac{1}{2}}.
\]

Now suppose \(\mu = \beta x\) is the product of \(\beta\) with a regression predictor \(x\). Then the reparameterization \(\mu = \alpha \tau^{-\frac{1}{2}} x\) has the same distribution, but in the original, direct parameterization, \(\beta\) has (potentially) heavy tails, whereas in the second, neither \(\tau\) nor \(\alpha\) have heavy tails.

To translate into Stan notation, this reparameterization replaces
parameters {
  real<lower=0> nu;
  real beta;
  ...
}

model {
  beta ~ student_t(nu, 0, 1);
  ...
}

with

parameters {
  real<lower=0> nu;
  real<lower=0> tau;
  real alpha;
  ...
}

transformed parameters {
  real beta;
  beta = alpha / sqrt(tau);
  ...
}

model {
  real half_nu;
  half_nu = 0.5 * nu;
  tau ~ gamma(half_nu, half_nu);
  alpha ~ std_normal();
  ...
}

Although set to 0 here, in most cases, the lower bound for the degrees of freedom parameter \( \nu \) can be set to 1 or higher; when \( \nu \) is 1, the result is a Cauchy distribution with fat tails and as \( \nu \) approaches infinity, the Student-t distribution approaches a normal distribution. Thus the parameter \( \nu \) characterizes the heaviness of the tails of the model.

Hierarchical models and the non-centered parameterization

Unfortunately, the usual situation in applied Bayesian modeling involves complex geometries and interactions that are not known analytically. Nevertheless, reparameterization can still be effective for separating parameters.

Centered parameterization

For example, a vectorized hierarchical model might draw a vector of coefficients \( \beta \) with definitions as follows. The so-called centered parameterization is as follows.
parameters {
    real mu_beta;
    real<lower = 0> sigma_beta;
    vector[K] beta;
    ...
}

model {
    beta ~ normal(mu_beta, sigma_beta);
    ...
}

Although not shown, a full model will have priors on both mu_beta and sigma_beta along with data modeled based on these coefficients. For instance, a standard binary logistic regression with data matrix x and binary outcome vector y would include a likelihood statement such as form y ~ bernoulli_logit(x * beta), leading to an analytically intractable posterior.

A hierarchical model such as the above will suffer from the same kind of inefficiencies as Neal’s funnel, because the values of beta, mu_beta and sigma_beta are highly correlated in the posterior. The extremity of the correlation depends on the amount of data, with Neal’s funnel being the extreme with no data. In these cases, the non-centered parameterization, discussed in the next section, is preferable; when there is a lot of data, the centered parameterization is more efficient. See Betancourt and Girolami (2013) for more information on the effects of centering in hierarchical models fit with Hamiltonian Monte Carlo.

**Non-centered parameterization**

Sometimes the group-level effects do not constrain the hierarchical distribution tightly. Examples arise when there are not many groups, or when the inter-group variation is high. In such cases, hierarchical models can be made much more efficient by shifting the data’s correlation with the parameters to the hyperparameters. Similar to the funnel example, this will be much more efficient in terms of effective sample size when there is not much data (see Betancourt and Girolami (2013)), and in more extreme cases will be necessary to achieve convergence.

parameters {
    vector[K] beta_raw;
    ...
}

transformed parameters {
    vector[K] beta;
    // implies: beta ~ normal(mu_beta, sigma_beta)
    beta = mu_beta + sigma_beta * beta_raw;
    model {

Any priors defined for mu_beta and sigma_beta remain as defined in the original model.

Reparameterization of hierarchical models is not limited to the normal distribution, although the normal distribution is the best candidate for doing so. In general, any distribution of parameters in the location-scale family is a good candidate for reparameterization. Let $\beta = l + s\alpha$ where $l$ is a location parameter and $s$ is a scale parameter. The parameter $l$ need not be the mean, $s$ need not be the standard deviation, and neither the mean nor the standard deviation need to exist. If $\alpha$ and $\beta$ are from the same distributional family but $\alpha$ has location zero and unit scale, while $\beta$ has location $l$ and scale $s$, then that distribution is a location-scale distribution. Thus, if $\alpha$ were a parameter and $\beta$ were a transformed parameter, then a prior distribution from the location-scale family on $\alpha$ with location zero and unit scale implies a prior distribution on $\beta$ with location $l$ and scale $s$. Doing so would reduce the dependence between $\alpha$, $l$, and $s$.

There are several univariate distributions in the location-scale family, such as the Student t distribution, including its special cases of the Cauchy distribution (with one degree of freedom) and the normal distribution (with infinite degrees of freedom). As shown above, if $\alpha$ is distributed standard normal, then $\beta$ is distributed normal with mean $\mu = l$ and standard deviation $\sigma = s$. The logistic, the double exponential, the generalized extreme value distributions, and the stable distribution are also in the location-scale family.

Also, if $z$ is distributed standard normal, then $z^2$ is distributed chi-squared with one degree of freedom. By summing the squares of $K$ independent standard normal variates, one can obtain a single variate that is distributed chi-squared with $K$ degrees of freedom. However, for large $K$, the computational gains of this reparameterization may be overwhelmed by the computational cost of specifying $K$ primitive parameters just to obtain one transformed parameter to use in a model.

### Multivariate reparameterizations

The benefits of reparameterization are not limited to univariate distributions. A parameter with a multivariate normal prior distribution is also an excellent candidate for reparameterization. Suppose you intend the prior for $\beta$ to be multivariate normal with mean vector $\mu$ and covariance matrix $\Sigma$. Such a belief is reflected by the following code.

```r
data {
  beta_raw ~ std_normal();
...
In this case \( \mu \) and \( \Sigma \) are fixed data, but they could be unknown parameters, in which case their priors would be unaffected by a reparameterization of \( \beta \).

If \( \alpha \) has the same dimensions as \( \beta \) but the elements of \( \alpha \) are independently and identically distributed standard normal such that \( \beta = \mu + L\alpha \), where \( LL^\top = \Sigma \), then \( \beta \) is distributed multivariate normal with mean vector \( \mu \) and covariance matrix \( \Sigma \). One choice for \( L \) is the Cholesky factor of \( \Sigma \). Thus, the model above could be reparameterized as follows.

data {
  int<lower=2> K;
  vector[K] mu;
  cov_matrix[K] Sigma;
...
}
transformed data {
  matrix[K, K] L;
  L = cholesky_decompose(Sigma);
}
parameters {
  vector[K] alpha;
...}
transformed parameters {
  vector[K] beta;
  beta = mu + L * alpha;
}
model {
  alpha ~ std_normal();
  // implies: beta ~ multi_normal(mu, Sigma)
  ...
}
This reparameterization is more efficient for two reasons. First, it reduces dependence among the elements of alpha and second, it avoids the need to invert Sigma every time multi_normal is evaluated.

The Cholesky factor is also useful when a covariance matrix is decomposed into a correlation matrix that is multiplied from both sides by a diagonal matrix of standard deviations, where either the standard deviations or the correlations are unknown parameters. The Cholesky factor of the covariance matrix is equal to the product of a diagonal matrix of standard deviations and the Cholesky factor of the correlation matrix. Furthermore, the product of a diagonal matrix of standard deviations and a vector is equal to the elementwise product between the standard deviations and that vector. Thus, if for example the correlation matrix Tau were fixed data but the vector of standard deviations sigma were unknown parameters, then a reparameterization of beta in terms of alpha could be implemented as follows.

data {
    int<lower=2> K;
    vector[K] mu;
    corr_matrix[K] Tau;
    ...}
transformed data {
    matrix[K, K] L;
    L = cholesky_decompose(Tau);
}
parameters {
    vector[K] alpha;
    vector<lower=0>[K] sigma;
    ...}
transformed parameters {
    vector[K] beta;
    // This equals mu + diag_matrix(sigma) * L * alpha;
    beta = mu + sigma .* (L * alpha);
}
model {
    sigma ~ cauchy(0, 5);
    alpha ~ std_normal();
    // implies: beta ~ multi_normal(mu,
    //    // diag_matrix(sigma) * L * L' * diag_matrix(sigma)))
    ...

This reparameterization of a multivariate normal distribution in terms of standard
normal variates can be extended to other multivariate distributions that can be conceptualized as contaminations of the multivariate normal, such as the multivariate Student t and the skew multivariate normal distribution.

A Wishart distribution can also be reparameterized in terms of standard normal variates and chi-squared variates. Let $L$ be the Cholesky factor of a $K \times K$ positive definite scale matrix $S$ and let $\nu$ be the degrees of freedom. If

$$A = \begin{pmatrix} \sqrt{c_1} & 0 & \cdots & 0 \\ z_{21} & \sqrt{c_2} & \cdots & \vdots \\ \vdots & \ddots & \ddots & 0 \\ z_{K1} & \cdots & z_{K(K-1)} & \sqrt{c_K} \end{pmatrix},$$

where each $c_i$ is distributed chi-squared with $\nu - i + 1$ degrees of freedom and each $z_{ij}$ is distributed standard normal, then $W = L A A^T L^T$ is distributed Wishart with scale matrix $S = L L^T$ and degrees of freedom $\nu$. Such a reparameterization can be implemented by the following Stan code:

data {
  int<lower=1> N;
  int<lower=1> K;
  int<lower=K+2> nu
  matrix[K, K] L; // Cholesky factor of scale matrix
  vector[K] mu;
  matrix[N, K] y;
  ...
}

parameters {
  vector<lower=0>[K] c;
  vector[0.5 * K * (K - 1)] z;
  ...
}

model {
  matrix[K, K] A;
  int count = 1;
  for (j in 1:(K-1)) {
    for (i in (j+1):K) {
      A[i, j] = z[count];
      count += 1;
    }
  }
  for (i in 1:(j - 1)) {
    A[i, j] = 0.0;
  }
}
A[j, j] = sqrt(c[j]);
for (i in 1:(K-1))
A[i, K] = 0;
A[K, K] = sqrt(c[K]);
for (i in 1:K)
c[i] ~ chi_square(nu - i + 1);
z ~ std_normal();
// implies: L * A * A' * L' ~ wishart(nu, L * L')
y ~ multi_normal_cholesky(mu, L * A);
...

This reparameterization is more efficient for three reasons. First, it reduces dependence among the elements of z and second, it avoids the need to invert the covariance matrix, W every time wishart is evaluated. Third, if W is to be used with a multivariate normal distribution, you can pass LA to the more efficient multi_normal_cholesky function, rather than passing W to multi_normal.

If W is distributed Wishart with scale matrix S and degrees of freedom ν, then $W^{-1}$ is distributed inverse Wishart with inverse scale matrix $S^{-1}$ and degrees of freedom ν. Thus, the previous result can be used to reparameterize the inverse Wishart distribution. Since $W = LAA^T L^T$, $W^{-1} = L^{-T} A^{-T} A^{-1} L^{-1}$, where all four inverses exist, but $L^{-1T} = L^{-T}$ and $A^{-1T} = A^{-T}$. We can slightly modify the above Stan code for this case:

data {
  int<lower=1> K;
  int<lower=K+2> nu
  matrix[K, K] L; // Cholesky factor of scale matrix
  ...
}
transformed data {
  matrix[K, K] eye;
  matrix[K, K] L_inv;
  for (j in 1:K) {
    for (i in 1:K) {
      eye[i, j] = 0.0;
    }
    eye[j, j] = 1.0;
L_inv = mdivide_left_tri_low(L, eye);
}
parameters {
  vector<lower=0>[K] c;
  vector[0.5 * K * (K - 1)] z;
...
model {
  matrix[K, K] A;
  matrix[K, K] A_inv_L_inv;
  int count;
  count = 1;
  for (j in 1:(K-1)) {
    for (i in (j+1):K) {
      A[i, j] = z[count];
      count += 1;
    }
    for (i in 1:(j - 1)) {
      A[i, j] = 0.0;
    }
    A[j, j] = sqrt(c[j]);
  }
  for (i in 1:(K-1))
    A[i, K] = 0;
  A[K, K] = sqrt(c[K]);

  A_inv_L_inv = mdivide_left_tri_low(A, L_inv);
  for (i in 1:K)
    c[i] ~ chi_square(nu - i + 1);

  z ~ std_normal(); // implies: crossprod(A_inv_L_inv) ~
  // inv_wishart(nu, L_inv' * L_inv)
...

Another candidate for reparameterization is the Dirichlet distribution with all $K$ shape parameters equal. Zyczkowski and Sommers (2001) shows that if $\theta_i$ is equal to the sum of $\beta$ independent squared standard normal variates and $\rho_i = \sum \frac{\theta_i}{\theta_1}$, then the $K$-vector $\rho$ is distributed Dirichlet with all shape parameters equal to $\frac{\beta}{2}$. In particular, if $\beta = 2$, then $\rho$ is uniformly distributed on the unit simplex. Thus, we
can make $\rho$ be a transformed parameter to reduce dependence, as in:

```stan
data {
  int<lower=1> beta;
  ...
}
parameters {
  vector[beta] z[K];
  ...
}
transformed parameters {
  simplex[K] rho;
  for (k in 1:K)
    rho[k] = dot_self(z[k]); // sum-of-squares
  rho = rho / sum(rho);
}
model {
  for (k in 1:K)
    z[k] ~ std_normal();
  // implies: rho ~ dirichlet(0.5 * beta * ones)
  ...
}
```

### 23.8. Vectorization

**Gradient bottleneck**

Stan spends the vast majority of its time computing the gradient of the log probability function, making gradients the obvious target for optimization. Stan's gradient calculations with algorithmic differentiation require a template expression to be allocated and constructed for each subexpression of a Stan program involving parameters or transformed parameters. This section defines optimization strategies based on vectorizing these subexpressions to reduce the work done during algorithmic differentiation.

**Vectorizing summations**

Because of the gradient bottleneck described in the previous section, it is more efficient to collect a sequence of summands into a vector or array and then apply the `sum()` operation than it is to continually increment a variable by assignment and addition. For example, consider the following code snippet, where `foo()` is some operation that depends on `n`.

```stan
for (n in 1:N)
  total += foo(n,...);
```

---

3Stan uses its own arena-based allocation, so allocation and deallocation are faster than with a raw call to `new`. 
This code has to create intermediate representations for each of the $N$ summands. A faster alternative is to copy the values into a vector, then apply the $\text{sum()}$ operator, as in the following refactoring.

```{vector}[N] summands;
  for (n in 1:N)
    summands[n] = foo(n,...);
  total = sum(summands);
```

Syntactically, the replacement is a statement block delineated by curly brackets ($\{,\}$), starting with the definition of the local variable $\text{summands}$.

Even though it involves extra work to allocate the $\text{summands}$ vector and copy $N$ values into it, the savings in differentiation more than make up for it. Perhaps surprisingly, it will also use substantially less memory overall than incrementing $\text{total}$ within the loop.

**Vectorization through matrix operations**

The following program directly encodes a linear regression with fixed unit noise using a two-dimensional array $x$ of predictors, an array $y$ of outcomes, and an array $beta$ of regression coefficients.

```data {
  int<lower=1> K;
  int<lower=1> N;
  real x[K, N];
  real y[N];
}

parameters {
  real beta[K];
}

model {
  for (n in 1:N) {
    real gamma = 0;
    for (k in 1:K)
      gamma += x[n, k] * beta[k];
    y[n] ~ normal(gamma, 1);
  }
}
The following model computes the same log probability function as the previous model, even supporting the same input files for data and initialization.

data {
    int<lower=1> K;
    int<lower=1> N;
    vector[K] x[N];
    real y[N];
}
parameters {
    vector[K] beta;
}
model {
    for (n in 1:N)
        y[n] ~ normal(dot_product(x[n], beta), 1);
}

Although it produces equivalent results, the dot product should not be replaced with a transpose and multiply, as in

```
y[n] ~ normal(x[n]' * beta, 1);
```

The relative inefficiency of the transpose and multiply approach is that the transposition operator allocates a new vector into which the result of the transposition is copied. This consumes both time and memory.\(^4\)

The inefficiency of transposition could itself be mitigated by reordering the product and pulling the transposition out of the loop, as follows.

```
...  
transformed parameters {
    row_vector[K] beta_t;
    beta_t = beta';
}
model {
    for (n in 1:N)
        y[n] ~ normal(beta_t * x[n], 1);
}
```

\(^4\)Future versions of Stan may remove this inefficiency by more fully exploiting expression templates inside the Eigen C++ matrix library. This will require enhancing Eigen to deal with mixed-type arguments, such as the type `double` used for constants and the algorithmic differentiation type `stan::math::var` used for variables.
The problem with transposition could be completely solved by directly encoding the $x$ as a row vector, as in the following example.

```stan
data {
  ...
  row_vector[K] x[N];
  ...
}
parameters {
  vector[K] beta;
}
model {
  for (n in 1:N)
    y[n] ~ normal(x[n] * beta, 1);
}
```

Declaring the data as a matrix and then computing all the predictors at once using matrix multiplication is more efficient still, as in the example discussed in the next section.

Having said all this, the most efficient way to code this model is with direct matrix multiplication, as in

```stan
data {
  matrix[N, K] x;
  vector[N] y;
}
parameters {
  vector[K] beta;
}
model {
  y ~ normal(x * beta, 1);
}
```

In general, encapsulated single operations that do the work of loops will be more efficient in their encapsulated forms. Rather than performing a sequence of row-vector/vector multiplications, it is better to encapsulate it as a single matrix/vector multiplication.

**Vectorized probability functions**

The final and most efficient version replaces the loops and transformed parameters by using the vectorized form of the normal probability function, as in the following
example.

data {
  int<lower=1> K;
  int<lower=1> N;
  matrix[N, K] x;
  vector[N] y;
}
parameters {
  vector[K] beta;
}
model {
  y ~ normal(x * beta, 1);
}

The variables are all declared as either matrix or vector types. The result of the
matrix-vector multiplication $x \times \beta$ in the model block is a vector of the same
length as $y$.

The probability function documentation in the function reference manual indicates
which of Stan’s probability functions support vectorization; see the function reference
manual for full details. Vectorized probability functions accept either vector or scalar
inputs for all arguments, with the only restriction being that all vector arguments
are the same dimensionality. In the example above, $y$ is a vector of size $N$, $x \times \beta$
is a vector of size $N$, and 1 is a scalar.

**Reshaping data for vectorization**

Sometimes data does not arrive in a shape that is ideal for vectorization, but can be
put into such shape with some munging (either inside Stan’s transformed data block
or outside).

John Hall provided a simple example on the Stan users group. Simplifying notation
a bit, the original model had a sampling statement in a loop, as follows.

```plaintext
for (n in 1:N)
  y[n] ~ normal(mu[ii[n]], sigma);
```

The brute force vectorization would build up a mean vector and then vectorize all at
once.

```plaintext
{
  vector[N] mu_ii;
  for (n in 1:N)
```
mu_ii[n] = mu[ii[n]];  
y ~ normal(mu_ii, sigma);
}

If there aren’t many levels (values ii[n] can take), then it behooves us to reorganize the data by group in a case like this. Rather than having a single observation vector y, there are K of them. And because Stan doesn’t support ragged arrays, it means K declarations. For instance, with 5 levels, we have

y_1 ~ normal(mu[1], sigma);
...
y_5 ~ normal(mu[5], sigma);

This way, both the mu and sigma parameters are shared. Which way works out to be more efficient will depend on the shape of the data; if the sizes are small, the simple vectorization may be faster, but for moderate to large sized groups, the full expansion should be faster.

### 23.9. Exploiting sufficient statistics

In some cases, models can be recoded to exploit sufficient statistics in estimation. This can lead to large efficiency gains compared to an expanded model. For example, consider the following Bernoulli sampling model.

```stan
data {
  int<lower=0> N;
  int<lower=0, upper=1> y[N];
  real<lower=0> alpha;
  real<lower=0> beta;
}
parameters {
  real<lower=0, upper=1> theta;
}
model {
  theta ~ beta(alpha, beta);
  for (n in 1:N)
    y[n] ~ bernoulli(theta);
}
```

In this model, the sum of positive outcomes in y is a sufficient statistic for the chance of success theta. The model may be recoded using the binomial distribution as follows.

```stan
theta ~ beta(alpha, beta);
```
sum(y) ~ binomial(N, theta);

Because truth is represented as one and falsehood as zero, the sum \( \sum y \) of a binary vector \( y \) is equal to the number of positive outcomes out of a total of \( N \) trials. This can be generalized to other discrete cases (one wouldn’t expect continuous observations to be duplicated if they are random). Suppose there are only \( K \) possible discrete outcomes, \( z_1, \ldots, z_K \), but there are \( N \) observations, where \( N \) is much larger than \( K \). If \( f_k \) is the frequency of outcome \( z_k \), then the entire likelihood with distribution \( f_{oo} \) can be coded as follows.

```plaintext
for (k in 1:K)
    target += f[k] * foo_lpmf(z[k] | ...);
```

where the ellipses are the parameters of the log probability mass function for distribution \( f_{oo} \) (there’s no distribution called “foo”; this is just a placeholder for any discrete distribution name).

The resulting program looks like a “weighted” regression, but here the weights \( f[k] \) are counts and thus sufficient statistics for the PMF and simply amount to an alternative, more efficient coding of the same likelihood. For efficiency, the frequencies \( f[k] \) should be counted once in the transformed data block and stored.

### 23.10. Aggregating common subexpressions
If an expression is calculated once, the value should be saved and reused wherever possible. That is, rather than using \( \exp(\theta) \) in multiple places, declare a local variable to store its value and reuse the local variable.

Another case that may not be so obvious is with two multilevel parameters, say \( a[ii[n]] + b[jj[n]] \). If \( a \) and \( b \) are small (i.e., do not have many levels), then a table \( a_b \) of their sums can be created, with

```plaintext
matrix[size(a), size(b)] a_b;
for (i in 1:size(a))
    for (j in 1:size(b))
        a_b[i, j] = a[i] + b[j];
```

Then the sum can be replaced with \( a_b[ii[n], jj[n]] \).

### 23.11. Exploiting conjugacy
Continuing the model from the previous section, the conjugacy of the beta prior and binomial sampling distribution allow the model to be further optimized to the following equivalent form.
theta ~ beta(alpha + sum(y), beta + N - sum(y));

To make the model even more efficient, a transformed data variable defined to be
\sum(y) could be used in the place of \sum(y).

### 23.12. Standardizing predictors and outputs

Stan programs will run faster if the input is standardized to have a zero sample
mean and unit sample variance. This section illustrates the principle with a simple
linear regression.

Suppose that \( y = (y_1, \ldots, y_N) \) is a sequence of \( N \) outcomes and \( x = (x_1, \ldots, x_N) \) a
parallel sequence of \( N \) predictors. A simple linear regression involving an intercept
coefficient \( \alpha \) and slope coefficient \( \beta \) can be expressed as

\[
y_n = \alpha + \beta x_n + \epsilon_n,
\]

where

\[
\epsilon_n \sim \text{normal}(0, \sigma).
\]

If either vector \( x \) or \( y \) has very large or very small values or if the sample mean
of the values is far away from 0 (on the scale of the values), then it can be more
efficient to standardize the outputs \( y_n \) and predictors \( x_n \). The data are first centered
by subtracting the sample mean, and then scaled by dividing by the sample deviation.
Thus a data point \( u \) is standardized with respect to a vector \( y \) by the function \( z_y \),
declared by

\[
  z_y(u) = \frac{u - \bar{y}}{\text{sd}(y)},
\]

where the sample mean of \( y \) is

\[
  \bar{y} = \frac{1}{N} \sum_{n=1}^{N} y_n,
\]

and the sample standard deviation of \( y \) is

\[
  \text{sd}(y) = \left( \frac{1}{N} \sum_{n=1}^{N} (y_n - \bar{y})^2 \right)^{1/2}.
\]

The inverse transform is defined by reversing the two normalization steps, first
rescaling by the same deviation and relocating by the sample mean,

\[
  z_y^{-1}(v) = \text{sd}(y)v + \bar{y}.
\]
To standardize a regression problem, the predictors and outcomes are standardized. This changes the scale of the variables, and hence changes the scale of the priors. Consider the following initial model.

```plaintext
data {
    int<lower=0> N;
    vector[N] y;
    vector[N] x;
}
parameters {
    real alpha;
    real beta;
    real<lower=0> sigma;
}
model {
    // priors
    alpha ~ normal(0, 10);
    beta ~ normal(0, 10);
    sigma ~ cauchy(0, 5);
    // likelihood
    for (n in 1:N)
        y[n] ~ normal(alpha + beta * x[n], sigma);
}
```

The data block for the standardized model is identical. The standardized predictors and outputs are defined in the transformed data block.

```plaintext
data {
    int<lower=0> N;
    vector[N] y;
    vector[N] x;
}
transformed data {
    vector[N] x_std;
    vector[N] y_std;
    x_std = (x - mean(x)) / sd(x);
    y_std = (y - mean(y)) / sd(y);
}
parameters {
    real alpha_std;
    real beta_std;
```
real<lower=0> sigma_std;
}
model {
  alpha_std ~ normal(0, 10);
  beta_std ~ normal(0, 10);
  sigma_std ~ cauchy(0, 5);
  for (n in 1:N)
    y_std[n] ~ normal(alpha_std + beta_std * x_std[n],
                        sigma_std);
}

The parameters are renamed to indicate that they aren’t the “natural” parameters, but the model is otherwise identical. In particular, the fairly diffuse priors on the coefficients and error scale are the same. These could have been transformed as well, but here they are left as is, because the scales make sense as diffuse priors for standardized data; the priors could be made more informative. For instance, because the outputs $y$ have been standardized, the error $\sigma$ should not be greater than 1, because that’s the scale of the noise for predictors $\alpha = \beta = 0$.

The original regression

$$y_n = \alpha + \beta x_n + \epsilon_n$$

has been transformed to a regression on the standardized variables,

$$z_y(y_n) = \alpha' + \beta' z_x(x_n) + \epsilon'_{n}.$$  

The original parameters can be recovered with a little algebra,

$$y_n = z_y^{-1}(z_y(y_n))$$

$$= z_y^{-1}(\alpha' + \beta' z_x(x_n) + \epsilon'_{n})$$

$$= z_y^{-1}\left(\alpha' + \beta' \left(\frac{x_n - \bar{x}}{\text{sd}(x)}\right) + \epsilon'_{n}\right)$$

$$= \text{sd}(y) \left(\alpha' + \beta' \left(\frac{x_n - \bar{x}}{\text{sd}(x)}\right) + \epsilon'_{n}\right) + \bar{y}$$

$$= \left(\text{sd}(y) \left(\alpha' - \beta' \frac{\bar{x}}{\text{sd}(x)}\right) + \bar{y}\right) + \left(\beta' \frac{\text{sd}(y)}{\text{sd}(x)}\right)x_n + \text{sd}(y)\epsilon'_n,$$

from which the original scale parameter values can be read off,

$$\alpha = \text{sd}(y) \left(\alpha' - \beta' \frac{\bar{x}}{\text{sd}(x)}\right) + \bar{y}; \quad \beta = \beta' \frac{\text{sd}(y)}{\text{sd}(x)}; \quad \sigma = \text{sd}(y)\sigma'.$$
These recovered parameter values on the original scales can be calculated within Stan using a generated quantities block following the model block,

```stan
generated quantities {
  real alpha;
  real beta;
  real<lower=0> sigma;
  alpha = sd(y) * (alpha_std - beta_std * mean(x) / sd(x))
          + mean(y);
  beta = beta_std * sd(y) / sd(x);
  sigma = sd(y) * sigma_std;
}
```

It is inefficient to compute all of the means and standard deviations every iteration; for more efficiency, these can be calculated once and stored as transformed data. Furthermore, the model sampling statement can be easily vectorized, for instance, in the transformed model, to

```stan
y_std ~ normal(alpha_std + beta_std * x_std, sigma_std);
```

**Standard normal distribution**

For many applications on the standard scale, normal distributions with location zero and scale one will be used. In these cases, it is more efficient to use

```stan
y ~ std_normal();
```

than to use

```stan
y ~ normal(0, 1);
```

because the subtraction of the location and division by the scale cancel, as does subtracting the log of the scale.

### 23.13. Using map-reduce

The map-reduce operation, even without multi-core MPI support, can be used to make programs more scalable and also more efficient. See the map-reduce chapter for more information on implementing map-reduce operations.

Map-reduce allows greater scalability because only the Jacobian of the mapped function for each shard is stored. The Jacobian consists of all of the derivatives of the outputs with respect to the parameters. During execution, the derivatives of the shard are evaluated using nested automatic differentiation. As often happens with modern CPUs, reduced memory overhead leads to increased memory locality and faster execution. The Jacobians are all computed with local memory and their
outputs stored contiguously in memory.
24. Parallelization

Stan has support for different types of parallelization: multi-threading with Intel Threading Building Blocks (TBB), multi-processing with Message Passing Interface (MPI) and manycore processing with OpenCL.

Multi-threading in Stan can be used with two mechanisms: reduce with summation and rectangular map. The latter can also be used with multi-processing.

The advantages of reduce with summation are:

1. More flexible argument interface, avoiding the packing and unpacking that is necessary with rectangular map.
2. Partitions data for parallelization automatically (this is done manually in rectangular map).
3. Is easier to use.

The advantages of rectangular map are:

1. Returns a list of vectors, while the reduce summation returns only a scalar.
2. Can be parallelized across multiple cores and multiple computers, while reduce summation can only parallelized across multiple cores on a single machine.

The actual speedup gained from using these functions will depend on many details. It is strongly recommended to only parallelize the computationally most expensive operations in a Stan program. Oftentimes this is the evaluation of the log likelihood for the observed data. When it is not clear which parts of the model is the most computationally expensive, we recommend using profiling, which is available in Stan 2.26 and newer.

Since only portions of a Stan program will run in parallel, the maximal speedup one can achieve is capped, a phenomenon described by Amdahl’s law.

24.1. Reduce-sum

It is often necessary in probabilistic modeling to compute the sum of a number of independent function evaluations. This occurs, for instance, when evaluating a number of conditionally independent terms in a log-likelihood. If \( g: U \rightarrow \text{real} \) is the function and \( \{ x_1, x_2, \ldots \} \) is an array of inputs, then that sum looks like:

\[
g(x_1) + g(x_2) + \ldots
\]
reduce_sum and reduce_sum_static are tools for parallelizing these calculations. For efficiency reasons the reduce function doesn’t work with the element-wise evaluated function \( g \), but instead the partial sum function \( f: \mathbb{U}[] \to \text{real} \), where \( f \) computes the partial sum corresponding to a slice of the sequence \( x \) passed in. Due to the associativity of the sum reduction it holds that:

\[
g(x_1) + g(x_2) + g(x_3) = f\left(\{x_1, x_2, x_3\}\right) \\
= f\left(\{x_1, x_2\}\right) + f\left(\{x_3\}\right) \\
= f\left(\{x_1\}\right) + f\left(\{x_2, x_3\}\right) \\
= f\left(\{x_1\}\right) + f\left(\{x_2\}\right) + f\left(\{x_3\}\right)
\]

With the partial sum function \( f: \mathbb{U}[] \to \text{real} \) reduction of a large number of terms can be evaluated in parallel automatically, since the overall sum can be partitioned into arbitrary smaller partial sums. The exact partitioning into the partial sums is not under the control of the user. However, since the exact numerical result will depend on the order of summation, Stan provides two versions of the reduce summation facility:

- reduce_sum: Automatically choose partial sums partitioning based on a dynamic scheduling algorithm.
- reduce_sum_static: Compute the same sum as reduce_sum, but partition the input in the same way for given data set (in reduce_sum this partitioning might change depending on computer load).

grainsize is the one tuning parameter. For reduce_sum, grainsize is a suggested partial sum size. A grainsize of 1 leaves the partitioning entirely up to the scheduler. This should be the default way of using reduce_sum unless time is spent carefully picking grainsize. For picking a grainsize, see details below.

For reduce_sum_static, grainsize specifies the maximal partial sum size. With reduce_sum_static it is more important to choose grainsize carefully since it entirely determines the partitioning of work. See details below.

For efficiency and convenience additional shared arguments can be passed to every term in the sum. So for the array \( \{x_1, x_2, \ldots\} \) and the shared arguments \( s_1, s_2, \ldots \) the effective sum (with individual terms) looks like:

\[
g(x_1, s_1, s_2, \ldots) + g(x_2, s_1, s_2, \ldots) + g(x_3, s_1, s_2, \ldots) + \ldots
\]

which can be written equivalently with partial sums to look like:

\[
f\left(\{x_1, x_2\}, s_1, s_2, \ldots\right) + f\left(\{x_3\}, s_1, s_2, \ldots\right)
\]

where the particular slicing of the \( x \) array can change.
Given this, the signatures are:

```plaintext
real reduce_sum(F f, T[] x, int grainsize, T1 s1, T2 s2, ...)
real reduce_sum_static(F f, T[] x, int grainsize, T1 s1, T2 s2, ...)
```

1. \( f \) - User defined function that computes partial sums
2. \( x \) - Array to slice, each element corresponds to a term in the summation
3. \( \text{grainsize} \) - Target for size of slices
4. \( s1, s2, ... \) - Arguments shared in every term

The user-defined partial sum functions have the signature:

```plaintext
real f(T[] x_slice, int start, int end, T1 s1, T2 s2, ...)
```

and take the arguments:

1. \( x\_slice \) - The subset of \( x \) (from `reduce_sum` / `reduce_sum_static`) for which this partial sum is responsible (\( x\_slice = x[start:end] \))
2. \( \text{start} \) - An integer specifying the first term in the partial sum
3. \( \text{end} \) - An integer specifying the last term in the partial sum (inclusive)
4. \( s1, s2, ... \) - Arguments shared in every term (passed on without modification from the `reduce_sum` / `reduce_sum_static` call)

The user-provided function \( f \) is expected to compute the partial sum with the terms \( \text{start} \) through \( \text{end} \) of the overall sum. The user function is passed the subset \( x[start:end] \) as \( x\_slice \). \( \text{start} \) and \( \text{end} \) are passed so that \( f \) can index any of the trailing \( sM \) arguments as necessary. The trailing \( sM \) arguments are passed without modification to every call of \( f \).

A `reduce_sum` (or `reduce_sum_static`) call:

```plaintext
real sum = reduce_sum(f, x, grainsize, s1, s2, ...);
```

can be replaced by either:

```plaintext
real sum = f(x, 1, size(x), s1, s2, ...);
```

or the code:

```plaintext
real sum = 0.0;
for(i in 1:size(x)) {
    sum += f({ x[i] }, i, i, s1, s2, ...);
}
```
**Example: logistic regression**

Logistic regression is a useful example to clarify both the syntax and semantics of reduce summation and how it can be used to speed up a typical model. A basic logistic regression can be coded in Stan as:

```stan
data {
  int N;
  int y[N];
  vector[N] x;
}
parameters {
  vector[2] beta;
}
model {
  beta ~ std_normal();
  y ~ bernoulli_logit(beta[1] + beta[2] * x);
}
```

In this model predictions are made about the $N$ outputs $y$ using the covariate $x$. The intercept and slope of the linear equation are to be estimated. The key point to getting this calculation to use reduce summation, is recognizing that the statement:

```stan
y ~ bernoulli_logit(beta[1] + beta[2] * x);
```

can be rewritten (up to a proportionality constant) as:

```stan
for(n in 1:N) {
}
```

Now it is clear that the calculation is the sum of a number of conditionally independent Bernoulli log probability statements, which is the condition where reduce summation is useful. To use the reduce summation, a function must be written that can be used to compute arbitrary partial sums of the total sum. Using the interface defined in Reduce-Sum, such a function can be written like:

```stan
functions {
  real partial_sum(int[] y_slice, 
                    int start, int end, 
                    vector x, 
                    vector beta) {
  }
}
```
The likelihood statement in the model can now be written:

```cpp
target += partial_sum(y, 1, N, x, beta); // Sum terms 1 to N of the likelihood
```

In this example, `y` was chosen to be sliced over because there is one term in the summation per value of `y`. Technically `x` would have worked as well. Use whatever conceptually makes the most sense for a given model, e.g. slice over independent terms like conditionally independent observations or groups of observations as in hierarchical models. Because `x` is a shared argument, it is subset accordingly with `start:end`. With this function, reduce summation can be used to automatically parallelize the likelihood:

```cpp
int grainsize = 1;
target += reduce_sum(partial_sum, y,
                      grainsize,
                      x, beta);
```

The reduce summation facility automatically breaks the sum into pieces and computes them in parallel. `grainsize = 1` specifies that the `grainsize` should be estimated automatically. The final model is:

```cpp
functions {
    real partial_sum(int[] y_slice,
                      int start, int end,
                      vector x,
                      vector beta) {
    }
}
data {
    int N;
    int y[N];
    vector[N] x;
}
parameters {
    vector[2] beta;
}
model {
    int grainsize = 1;
    beta ~ std_normal();
}
```
target += reduce_sum(partial_sum, y, 
    grainsize, 
    x, beta);
}

**Picking the grainsize**
The rational for choosing a sensible grainsize is based on balancing the overhead implied by creating many small tasks versus creating fewer large tasks which limits the potential parallelism.

In `reduce_sum`, grainsize is a recommendation on how to partition the work in the partial sum into smaller pieces. A grainsize of 1 leaves this entirely up to the internal scheduler and should be chosen if no benchmarking of other grainsizes is done. Ideally this will be efficient, but there are no guarantees.

In `reduce_sum_static`, grainsize is an upper limit on the worksize. Work will be split until all partial sums are just smaller than grainsize (and the split will happen the same way every time for the same inputs). For the static version it is more important to select a sensible grainsize.

In order to figure out an optimal grainsize, if there are N terms and M cores, run a quick test model with grainsize set roughly to N / M. Record the time, cut the grainsize in half, and run the test again. Repeat this iteratively until the model runtime begins to increase. This is a suitable grainsize for the model, because this ensures the calculations can be carried out with the most parallelism without losing too much efficiency.

For instance, in a model with N=10000 and M = 4, start with grainsize = 25000, and sequentially try grainsize = 12500, grainsize = 6250, etc.

It is important to repeat this process until performance gets worse. It is possible after many halvings nothing happens, but there might still be a smaller grainsize that performs better. Even if a sum has many tens of thousands of terms, depending on the internal calculations, a grainsize of thirty or forty or smaller might be the best, and it is difficult to predict this behavior. Without doing these halvings until performance actually gets worse, it is easy to miss this.

**24.2. Map-rect**
Map-reduce allows large calculations (e.g., log likelihoods) to be broken into components which may be calculated modularly (e.g., data blocks) and combined (e.g., by summation and incrementing the target log density).

A *map function* is a higher-order function that applies an argument function to
every member of some collection, returning a collection of the results. For example, mapping the square function, \( f(x) = x^2 \), over the vector \([3, 5, 10]\) produces the vector \([9, 25, 100]\). In other words, map applies the square function elementwise.

The output of mapping a sequence is often fed into a reduction. A reduction function takes an arbitrarily long sequence of inputs and returns a single output. Examples of reduction functions are summation (with the return being a single value) or sorting (with the return being a sorted sequence). The combination of mapping and reducing is so common it has its own name, map-reduce.

**Map function**

In order to generalize the form of functions and results that are possible and accommodate both parameters (which need derivatives) and data values (which don’t), Stan’s map function operates on more than just a sequence of inputs.

**Map function signature**

Stan’s map function has the following signature

```plaintext
vector map_rect((vector, vector, real[], int[]):vector f, vector phi, vector[] thetas, data real[ , ] x_rs, data int[ , ] x_is);
```

The arrays `thetas` of parameters, `x_rs` of real data, and `x_is` of integer data have the suffix “s” to indicate they are arrays. These arrays must all be the same size, as they will be mapped in parallel by the function `f`. The value of `phi` is reused in each mapped operation.

The `_rect` suffix in the name arises because the data structures it takes as arguments are rectangular. In order to deal with ragged inputs, ragged inputs must be padded out to rectangular form.

The last two arguments are two dimensional arrays of real and integer data values. These argument types are marked with the `data` qualifier to indicate that they must only contain variables originating in the data or transformed data blocks. This will allow such data to be pinned to a processor on which it is being processed to reduce communication overhead.

The notation `(vector, vector, real[], int[]):vector` indicates that the function argument `f` must have the following signature.

```plaintext
vector f(vector phi, vector theta, data real[] x_r, data int[] x_i);
```

Although `f` will often return a vector of size one, the built-in flexibility allows general
multivariate functions to be mapped, even raggedly.

**Map function semantics**

Stan’s map function applies the function $f$ to the shared parameters along with one element each of the job parameters, real data, and integer data arrays. Each of the arguments $\theta$, $x_r$, and $x_i$ must be arrays of the same size. If the arrays are all size $N$, the result is defined as follows.

\[
\text{map_rect}(f, \phi, \theta, x, n) = f(\phi, \theta[1], x[1], n[1]) \cdot f(\phi, \theta[2], x[2], n[2]) \cdots f(\phi, \theta[N], x[N], n[N])
\]

The dot operators in the notation above are meant to indicate concatenation (implemented as `append_row` in Stan). The output of each application of $f$ is a vector, and the sequence of $N$ vectors is concatenated together to return a single vector.

**Example: logistic regression**

An example should help to clarify both the syntax and semantics of the mapping operation and how it may be combined with reductions built into Stan to provide a map-reduce implementation.

**Unmapped logistic regression**

Consider the following simple logistic regression model, which is coded unconventionally to accommodate direct translation to a mapped implementation.

```stan
data {
  int y[12];
  real x[12];
}
parameters {
  vector[2] beta;
}
model {
  beta ~ std_normal();
}
```

The program is unusual in that it (a) hardcodes the data size, which is not required by the map function but is just used here for simplicity, (b) represents the predictors as a real array even though it needs to be used as a vector, and (c) represents the regression coefficients (intercept and slope) as a vector even though they’re used
individually. The `bernoulli_logit` distribution is used because the argument is on the logit scale—it implicitly applies the inverse logit function to map the argument to a probability.

**Mapped logistic regression**

The unmapped logistic regression model described in the previous subsection may be implemented using Stan’s rectangular mapping functionality as follows.

```stan
functions {
  vector lr(vector beta, vector theta, real[] x, int[] y) {
    real lp = bernoulli_logit_lpmf(y | beta[1]
                                + to_vector(x) * beta[2]);
    return [lp]'
  }
}
data {
  int y[12];
  real x[12];
}
transformed data {
  // K = 3 shards
  int ys[3, 4] = { y[1:4], y[5:8], y[9:12] };
  real xs[3, 4] = { x[1:4], x[5:8], x[9:12] };
  vector[0] theta[3];
}
parameters {
  vector[2] beta;
}
model {
  beta ~ std_normal();
  target += sum(map_rect(lr, beta, theta, xs, ys));
}
```

The first piece of the code is the actual function to compute the logistic regression. The argument `beta` will contain the regression coefficients (intercept and slope), as before. The second argument `theta` of job-specific parameters is not used, but nevertheless must be present. The modeled data `y` is passed as an array of integers and the predictors `x` as an array of real values. The function body then computes the log probability mass of `y` and assigns it to the local variable `lp`. This variable is then used in `[lp]'` to construct a row vector and then transpose it to a vector to return.
The data are taken in as before. There is an additional transformed data block that breaks the data up into three shards.\(^1\)

The value 3 is also hard coded; a more practical program would allow the number of shards to be controlled. There are three parallel arrays defined here, each of size three, corresponding to the number of shards. The array \(ys\) contains the modeled data variables; each element of the array \(ys\) is an array of size four. The second array \(xs\) is for the predictors, and each element of it is also of size four. These contained arrays are the same size because the predictors \(x\) stand in a one-to-one relationship with the modeled data \(y\). The final array \(theta\) is also of size three; its elements are empty vectors, because there are no shard-specific parameters.

The parameters and the prior are as before. The likelihood is now coded using map-reduce. The function \(lr\) to compute the log probability mass is mapped over the data \(xs\) and \(ys\), which contain the original predictors and outcomes broken into shards. The parameters \(beta\) are in the first argument because they are shared across shards. There are no shard-specific parameters, so the array of job-specific parameters \(theta\) contains only empty vectors.

**Example: hierarchical logistic regression**

Consider a hierarchical model of American presidential voting behavior based on state of residence.\(^2\)

Each of the fifty states \(k \in \{1, \ldots, 50\}\) will have its own slope \(\beta_k\) and intercept \(\alpha_k\) to model the log odds of voting for the Republican candidate as a function of income. Suppose there are \(N\) voters and with voter \(n \in 1:N\) being in state \(s[n]\) with income \(x_n\). The likelihood for the vote \(y_n \in \{0, 1\}\) is

\[
y_n \sim \text{Bernoulli}\left(\text{logit}^{-1}\left(\alpha_{s[n]} + \beta_{s[n]} x_n\right)\right).
\]

The slopes and intercepts get hierarchical priors,

\[
\alpha_k \sim \text{normal}(\mu_\alpha, \sigma_\alpha)
\]

\[
\beta_k \sim \text{normal}(\mu_\beta, \sigma_\beta)
\]

**Unmapped implementation**

This model can be coded up in Stan directly as follows.

---

\(^1\)The term “shard” is borrowed from databases, where it refers to a slice of the rows of a database. That is exactly what it is here if we think of rows of a dataframe. Stan’s shards are more general in that they need not correspond to rows of a dataframe.

\(^2\)This example is a simplified form of the model described in (Andrew Gelman and Hill 2007, sec. 14.2)
data {
  int<lower = 0> K;
  int<lower = 0> N;
  int<lower = 1, upper = K> kk[N];
  vector[N] x;
  int<lower = 0, upper = 1> y[N];
}

parameters {
  matrix[K,2] beta;
  vector[2] mu;
  vector<lower=0>[2] sigma;
}

model {
  mu ~ normal(0, 2);
  sigma ~ normal(0, 2);
  for (i in 1:2)
    beta[ , i] ~ normal(mu[i], sigma[i]);
  y ~ bernoulli_logit(beta[kk, 1] + beta[kk, 2] .* x);
}

For this model the vector of predictors $x$ is coded as a vector, corresponding to how it is used in the likelihood. The priors for $\mu$ and $\sigma$ are vectorized. The priors on the two components of $\beta$ (intercept and slope, respectively) are stored in a $K \times 2$ matrix.

The likelihood is also vectorized using multi-indexing with index $kk$ for the states and elementwise multiplication ($\cdot\ast$) for the income $x$. The vectorized likelihood works out to the same thing as the following less efficient looped form.

\begin{verbatim}
for (n in 1:N)
  y[n] ~ bernoulli_logit(beta[kk[n], 1] + beta[kk[n], 2] * x[n]);
\end{verbatim}

**Mapped implementation**

The mapped version of the model will map over the states $K$. This means the group-level parameters, real data, and integer-data must be arrays of the same size.

The mapped implementation requires a function to be mapped. The following function evaluates both the likelihood for the data observed for a group as well as the prior for the group-specific parameters (the name `bl_glm` derives from the fact that it’s a generalized linear model with a Bernoulli likelihood and logistic link function).
functions {
    vector bl_glm(vector mu_sigma, vector beta,  
        real[] x, int[] y) {
        vector[2] mu = mu_sigma[1:2];
        vector[2] sigma = mu_sigma[3:4];
        real lp = normal_lpdf(beta | mu, sigma);
        real ll = bernoulli_logit_lpmf(y | beta[1] + beta[2] * to_vector(x));
        return [lp + ll];
    }
}

The shared parameter mu_sigma contains the locations (mu_sigma[1:2]) and scales (mu_sigma[3:4]) of the priors, which are extracted in the first two lines of the program. The variable lp is assigned the log density of the prior on beta. The vector beta is of size two, as are the vectors mu and sigma, so everything lines up for the vectorization. Next, the variable ll is assigned to the log likelihood contribution for the group. Here beta[1] is the intercept of the regression and beta[2] the slope. The predictor array x needs to be converted to a vector allow the multiplication.

The data block is identical to that of the previous program, but repeated here for convenience. A transformed data block computes the data structures needed for the mapping by organizing the data into arrays indexed by group.

data {
    int<lower = 0> K;
    int<lower = 0> N;
    int<lower = 1, upper = K> kk[N];
    vector[N] x;
    int<lower = 0, upper = 1> y[N];
}

transformed data {
    int<lower = 0> J = N / K;
    real x_r[K, J];
    int<lower = 0, upper = 1> x_i[K, J];
    {
        int pos = 1;
        for (k in 1:K) {
            int end = pos + J - 1;
            x_r[k] = to_array_1d(x[pos:end]);
            x_i[k] = to_array_1d(y[pos:end]);
            pos += J;
        }
    }
}
The integer $J$ is set to the number of observations per group.\(^3\)

The real data array $x_r$ holds the predictors and the integer data array $x_i$ holds the outcomes. The grouped data arrays are constructed by slicing the predictor vector $x$ (and converting it to an array) and slicing the outcome array $y$.

Given the transformed data with groupings, the parameters are the same as the previous program. The model has the same priors for the hyperparameters $\mu$ and $\sigma$, but moves the prior for $\beta$ and the likelihood to the mapped function.

```stan
parameters {
  vector[2] beta[K];
  vector[2] mu;
  vector<lower=0>[2] sigma;
}

model {
  mu ~ normal(0, 2);
  sigma ~ normal(0, 2);
  target += sum(map_rect(bl_glm, append_row(mu, sigma), beta, x_r, x_i));
}
```

The model as written here computes the priors for each group’s parameters along with the likelihood contribution for the group. An alternative mapping would leave the prior in the model block and only map the likelihood. In a serial setting this shouldn’t make much of a difference, but with parallelization, there is reduced communication (the prior’s parameters need not be transmitted) and also reduced parallelization with the version that leaves the prior in the model block.

**Ragged inputs and outputs**

The previous examples included rectangular data structures and single outputs. Despite the name, this is not technically required by `map_rect`.

**Ragged inputs**

If each group has a different number of observations, then the rectangular data structures for predictors and outcomes will need to be padded out to be rectangular.

\(^3\)This makes the strong assumption that each group has the same number of observations!
In addition, the size of the ragged structure will need to be passed as integer data. This holds for shards with varying numbers of parameters as well as varying numbers of data points.

**Ragged outputs**

The output of each mapped function is concatenated in order of inputs to produce the output of map_rect. When every shard returns a singleton (size one) array, the result is the same size as the number of shards and is easy to deal with downstream. If functions return longer arrays, they can still be structured using the \texttt{to\_matrix} function if they are rectangular.

If the outputs are of varying sizes, then there will have to be some way to convert it back to a usable form based on the input, because there is no way to directly return sizes or a ragged structure.

### 24.3. OpenCL

OpenCL (Open Computing Language) is a framework that enables writing programs that execute across heterogeneous platforms. An OpenCL program can be run on CPUs and GPUs. In order to run OpenCL programs, an OpenCL runtime be installed on the target system.

Stan’s OpenCL backend is currently supported in CmdStan and its wrappers. In order to use it, the model must be compiled with the \texttt{STAN\_OPENCL} makefile flag. Setting this flag means that the Stan-to-C++ translator (\texttt{stanc3}) will be supplied the \texttt{--use-opencl} flag and that the OpenCL enabled backend (Stan Math functions) will be enabled.

In Stan, the following distributions can be automatically run in parallel on both CPUs and GPUs with OpenCL:

- bernoulli_lpmf
- bernoulli_logit_lpmf
- bernoulli_logit glm_lpmf
- beta_lpdf
- beta_proportion_lpdf
- binomial_lpmf
- categorical_logit glm_lpmf
- cauchy_lpdf
• chi_square_lpdf
• double_exponential_lpdf
• exp_mod_normal_lpdf
• exponential_lpdf
• frechet_lpdf
• gamma_lpdf
• gumbel_lpdf
• inv_chi_square_lpdf
• inv_gamma_lpdf
• logistic_lpdf
• lognormal_lpdf
• neg_binomial_lpmf
• neg_binomial_2_lpmf
• neg_binomial_2_log_lpmf
• neg_binomial_2_log_glm_lpmf*
• normal_lpdf
• normal_id(glm_lpdf)*
• ordered_logistic(glm_lpmf)*
• pareto_lpdf
• pareto_type_2_lpdf
• poisson_lpmf
• poisson_log_lpmf
• poisson_log(glm_lpmf)*
• rayleigh_lpdf
• scaled_inv_chi_square_lpdf
• skew_normal_lpdf
• std_normal_lpdf
• student_t_lpdf
• uniform_lpdf
• weibull_lpdf
• OpenCL is not used when the covariate argument to the GLM functions is a row_vector.
Part 3. Posterior Inference & Model Checking

This part of the user's guide surveys algorithms that perform posterior predictive inference and model checking. Many of these algorithms require multiple runs of Stan in order to fit varying simulated data samples.
25. **Posterior Predictive Sampling**

The goal of inference is often posterior prediction, that is evaluating or sampling from the posterior predictive distribution \( p(\tilde{y} \mid y) \), where \( y \) is observed data and \( \tilde{y} \) is yet to be observed data. Often there are unmodeled predictors \( x \) and \( \tilde{x} \) for the observed data \( y \) and unobserved data \( \tilde{y} \). With predictors, the posterior predictive density is \( p(\tilde{y} \mid \tilde{x}, x, y) \). All of these variables may represent multivariate quantities.

This chapter explains how to sample from the posterior predictive distribution in Stan, including applications to posterior predictive simulation and calculating event probabilities. These techniques can be coded in Stan using random number generation in the generated quantities block. Further, a technique for fitting and performing inference in two stages is presented in a section on stand-alone generated quantities in Stan.

### 25.1. **Posterior predictive distribution**

Given a full Bayesian model \( p(y, \theta) \), the posterior predictive density for new data \( \tilde{y} \) given observed data \( y \) is

\[
p(\tilde{y} \mid y) = \int p(\tilde{y} \mid \theta) \cdot p(\theta \mid y) \, d\theta.
\]

The product under the integral reduces to the joint posterior density \( p(\tilde{y}, \theta \mid y) \), so that the integral is simply marginalizing out the parameters \( \theta \), leaving the predictive density \( p(\tilde{y} \mid y) \) of future observations given past observations.

### 25.2. **Computing the posterior predictive distribution**

The posterior predictive density (or mass) of a prediction \( \tilde{y} \) given observed data \( y \) can be computed using Monte Carlo draws

\[
\theta^{(m)} \sim p(\theta \mid y)
\]

from the posterior as

\[
p(\tilde{y} \mid y) \approx \frac{1}{M} \sum_{m=1}^{M} p(\tilde{y} \mid \theta^{(m)}).
\]

Computing directly using this formula will lead to underflow in many situations, but the log posterior predictive density, \( \log p(\tilde{y} \mid y) \) may be computed using the stable
CHAPTER 25. POSTERIOR PREDICTIVE SAMPLING

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log sum of exponents function as

$$\log p(\tilde{y} | y) \approx \log \frac{1}{M} \sum_{m=1}^{M} p(\tilde{y} | \theta^{(m)}).$$

$$= -\log M + \log\text{-sum-exp}_{m=1}^{M} \log p(\tilde{y} | \theta^{(m)}),$$

where

$$\log\text{-sum-exp}_{m=1}^{M} v_m = \log \sum_{m=1}^{M} \exp v_m$$

is used to maintain arithmetic precision. See the section on log sum of exponentials for more details.

25.3. Sampling from the posterior predictive distribution

Given draws from the posterior $\theta^{(m)} \sim p(\theta | y)$, draws from the posterior predictive $\tilde{y}^{(m)} \sim p(\tilde{y} | y)$ can be generated by randomly generating from the sampling distribution with the parameter draw plugged in,

$$\tilde{y}^{(m)} \sim p(y | \theta^{(m)}).$$

Randomly drawing $\tilde{y}$ from the sampling distribution is critical because there are two forms of uncertainty in posterior predictive quantities, sampling uncertainty and estimation uncertainty. Estimation uncertainty arises because $\theta$ is being estimated based only on a sample of data $y$. Sampling uncertainty arises because even a known value of $\theta$ leads to a sampling distribution $p(\tilde{y} | \theta)$ with variation in $\tilde{y}$. Both forms of uncertainty show up in the factored form of the posterior predictive distribution,

$$p(\tilde{y} | y) = \int \left[ \underbrace{p(\tilde{y} | \theta)}_{\text{sampling uncertainty}} \cdot \underbrace{p(\theta | y)}_{\text{estimation uncertainty}} \right] \, d\theta.$$

25.4. Posterior predictive simulation in Stan

Posterior predictive quantities can be coded in Stan using the generated quantities block.

Simple Poisson model

For example, consider a simple Poisson model for count data with a rate parameter $\lambda > 0$ following a gamma-distributed prior,

$$\lambda \sim \text{gamma}(1, 1).$$
The likelihood for $N$ observations $y_1, \ldots, y_N$ is modeled as Poisson,

$$y_n \sim \text{poisson}(\lambda).$$

**Stan code**

The following Stan program defines a variable for $\tilde{y}$ by random number generation in the generated quantities block.

```stan
data {
  int<lower = 0> N;
  int<lower = 0> y[N];
}
parameters {
  real<lower = 0> lambda;
}
model {
  lambda ~ gamma(1, 1);
  y ~ poisson(lambda);
}
generated quantities {
  int<lower = 0> y_tilde = poisson_rng(lambda);
}
```

The random draw from the sampling distribution for $\tilde{y}$ is coded using Stan’s Poisson random number generator in the generated quantities block. This accounts for the sampling component of the uncertainty; Stan’s posterior sampler will account for the estimation uncertainty, generating a new $\tilde{y}^{(m)} \sim p(y \mid \lambda^{(m)})$ for each posterior draw $\lambda^{(m)} \sim p(\theta \mid y)$.

The posterior draws $\tilde{y}^{(m)}$ may be used to estimate the expected value of $\tilde{y}$ or any of its quantiles or posterior intervals, as well as event probabilities involving $\tilde{y}$. In general, $E[f(\tilde{y}, \theta) \mid y]$ may be evaluated as

$$E[f(\tilde{y}, \theta) \mid y] \approx \frac{1}{M} \sum_{m=1}^{M} f(\tilde{y}^{(m)}, \theta^{(m)}),$$

which is just the posterior mean of $f(\tilde{y}, \theta)$. This quantity is computed by Stan if the value of $f(\tilde{y}, \theta)$ is assigned to a variable in the generated quantities block. That is, if we have

```stan
generated quantities {
  real f_val = f(y_tilde, theta);
}
```
where the value of \( f(\bar{y}, \theta) \) is assigned to variable \( f_{\text{val}} \), then the posterior mean of \( f_{\text{val}} \) will be the expectation \( \mathbb{E}[f(\bar{y}, \theta) \mid y] \).

**Analytic posterior and posterior predictive**

The gamma distribution is the conjugate prior distribution for the Poisson distribution, so the posterior density \( p(\lambda \mid y) \) will also follow a gamma distribution. Because the posterior follows a gamma distribution and the sampling distribution is Poisson, the posterior predictive \( p(\tilde{y} \mid y) \) will follow a negative binomial distribution, because the negative binomial is defined as a compound gamma-Poisson. That is, \( y \sim \text{negative-binomial}(\alpha, \beta) \) if \( \lambda \sim \text{gamma}(\alpha, \beta) \) and \( y \sim \text{poisson}(\lambda) \). Rather than marginalizing out the rate parameter \( \lambda \) analytically as can be done to define the negative binomial probability mass function, the rate \( \lambda^{(m)} \sim p(\lambda \mid y) \) is sampled from the posterior and then used to generate a draw of \( \tilde{y}^{(m)} \sim p(y \mid \lambda^{(m)}) \).

### 25.5. Posterior prediction for regressions

**Posterior predictive distributions for regressions**

Consider a regression with a single predictor \( x_n \) for the training outcome \( y_n \) and \( \tilde{x}_n \) for the test outcome \( \tilde{y}_n \). Without considering the parametric form of any of the distributions, the posterior predictive distribution for a general regression in

\[
p(\bar{y} \mid \tilde{x}, y, x) = \int p(\bar{y} \mid x, \theta) \cdot p(\theta \mid y, x) \, d\theta
\]

\[
\approx \frac{1}{M} \sum_{m=1}^{M} p(\bar{y} \mid \tilde{x}, \theta^{(m)}),
\]

where \( \theta^{(m)} \sim p(\theta \mid x, y) \).

**Stan program**

The following program defines a Poisson regression with a single predictor. These predictors are all coded as data, as are their sizes. Only the observed \( y \) values are coded as data. The predictive quantities \( \bar{y} \) appear in the generated quantities block, where they are generated by random number generation.

```stan
data {
  int<lower = 0> N;
  vector[N] x;
  int<lower = 0> y[N];
  int<lower = 0> N_tilde;
  vector[N_tilde] x_tilde;
}...
parameters {
    real alpha;
    real beta;
}

model {
    y ~ poisson_log(alpha + beta * x);
    { alpha, beta } ~ normal(0, 1);
}

generated quantities {
    int<lower = 0> y_tilde[N_tilde]
    = poisson_log_rng(alpha + beta * x_tilde);
}

The Poisson distributions in both the model and generated quantities block are coded using the log rate as a parameter (that’s \( \text{poisson\_log} \) vs. \( \text{poisson} \), with the suffixes defining the scale of the parameter). The regression coefficients, an intercept \( \alpha \) and slope \( \beta \), are given standard normal priors.

In the model block, the log rate for the Poisson is a linear function of the training data \( x \), whereas in the generated quantities block it is a function of the test data \( \tilde{x} \). Because the generated quantities block does not affect the posterior draws, the model fits \( \alpha \) and \( \beta \) using only the training data, reserving \( \tilde{x} \) to generate \( \tilde{y} \).

The result from running Stan is a predictive sample \( \tilde{y}^{(1)}, \ldots, \tilde{y}^{(M)} \) where each \( \tilde{y}^{(m)} \sim p(\tilde{y} | \tilde{x}, x, y) \).

The mean of the posterior predictive distribution is the expected value

\[
\mathbb{E}[\tilde{y} | \tilde{x}, x, y] = \int \tilde{y} \cdot p(\tilde{y} | \tilde{x}, \theta) \cdot p(\theta | x, y) \, d\theta \tag{25.3}
\]

\[
\approx \frac{1}{M} \sum_{m=1}^{M} \tilde{y}^{(m)}, \tag{25.4}
\]

where the \( \tilde{y}^{(m)} \sim p(\tilde{y} | \tilde{x}, x, y) \) are drawn from the posterior predictive distribution. Thus the posterior mean of \( y_{\text{tilde}}[n] \) after running Stan is the expected value of \( \tilde{y}_n \) conditioned on the training data \( x, y \) and predictor \( \tilde{x}_n \). This is the Bayesian estimate for \( \tilde{y} \) with minimum expected squared error. The posterior draws can also be used to estimate quantiles for the median and any posterior intervals of interest for \( \tilde{y} \), as well as covariance of the \( \tilde{y}_n \). The posterior draws \( \tilde{y}^{(m)} \) may also be used to
estimate predictive event probabilities, such as $\Pr[\tilde{y}_1 > 0]$ or $\Pr[\prod_{n=1}^{\tilde{N}} \tilde{y}_n > 1]$, as expectations of indicator functions.

All of this can be carried out by running Stan only a single time to draw a single sample of $M$ draws,

$$\tilde{y}^{(1)}, \ldots, \tilde{y}^{(M)} \sim p(\tilde{y} | \tilde{x}, x, y).$$

It’s only when moving to cross-validation where multiple runs are required.

### 25.6. Estimating event probabilities

Event probabilities involving either parameters or predictions or both may be coded in the generated quantities block. For example, to evaluate $\Pr[\lambda > 5 | y]$ in the simple Poisson example with only a rate parameter $\lambda$, it suffices to define a generated quantity

```stan
generated quantities {
  int<lower = 0, upper = 1> lambda_gt_5 = lambda > 5;
  ...}
```

The value of the expression $\lambda > 5$ is 1 if the condition is true and 0 otherwise. The posterior mean of this parameter is the event probability

$$\Pr[\lambda > 5 | y] = \int I(\lambda > 5) \cdot p(\lambda | y) \, d\lambda$$

$$\approx \frac{1}{M} \sum_{m=1}^{M} I(\lambda^{(m)} > 5),$$

where each $\lambda^{(m)} \sim p(\lambda | y)$ is distributed according to the posterior. In Stan, this is recovered as the posterior mean of the parameter `lambda_gt_5`.

In general, event probabilities may be expressed as expectations of indicator functions. For example,

$$\Pr[\lambda > 5 | y] = \mathbb{E}[I[\lambda > 5] | y]$$

$$\approx \frac{1}{M} \sum_{m=1}^{M} I(\lambda^{(m)} > 5).$$

The last line above is the posterior mean of the indicator function as coded in Stan.
Event probabilities involving posterior predictive quantities $\tilde{y}$ work exactly the same way as those for parameters. For example, if $\tilde{y}_n$ is the prediction for the $n$-th unobserved outcome (such as the score of a team in a game or a level of expression of a protein in a cell), then

$$
\Pr[\tilde{y}_3 > \tilde{y}_7 | \tilde{x}, x, y] = \mathbb{E}[I(\tilde{y}_3 > \tilde{y}_7) | \tilde{x}, x, y] = \int I(\tilde{y}_3 > \tilde{y}_7) \cdot p(\tilde{y} | \tilde{x}, x, y) \, d\tilde{y}
$$

$$
\approx \frac{1}{M} \sum_{m=1}^{M} I(\tilde{y}_3^{(m)} > \tilde{y}_7^{(m)}),
$$

where $\tilde{y}^{(m)} \sim p(\tilde{y} | \tilde{x}, x, y)$.

### 25.7. Stand-alone generated quantities and ongoing prediction

Stan’s sampling algorithms take a Stan program representing a posterior $p(\theta | y, x)$ along with actual data $x$ and $y$ to produce a set of draws $\theta^{(1)}, \ldots, \theta^{(M)}$ from the posterior. Posterior predictive draws $\tilde{y}^{(m)} \sim p(\tilde{y} | \tilde{x}, \theta^{(m)})$ can be generated by drawing

$$
\tilde{y}^{(m)} \sim p(\tilde{y} | \tilde{x}, \theta^{(m)})
$$

from the sampling distribution. Note that drawing $\tilde{y}^{(m)}$ only depends on the new predictors $\tilde{x}$ and the posterior draws $\theta^{(m)}$. Most importantly, neither the original data or the model density is required.

By saving the posterior draws, predictions for new data items $\tilde{x}$ may be generated whenever needed. In Stan’s interfaces, this is done by writing a second Stan program that inputs the original program’s parameters and the new predictors. For example, for the linear regression case, the program to take posterior draws declares the data and parameters, and defines the model.

```stan
data {
    int<lower = 0> N;
    vector[N] x;
    vector[N] y;
}
parameters {
    real alpha;
    real beta;
    real<lower = 0> sigma;
}
```
model {
  y ~ normal(alpha + beta * x, sigma);
  alpha ~ normal(0, 5);
  beta ~ normal(0, 1);
  sigma ~ lognormal(0, 0.5);
}

A second program can be used to generate new observations. This follow-on program need only declare the parameters as they were originally defined. This may require defining constants in the data block such as sizes and hyperparameters that are involved in parameter size or constraint declarations. Then additional data is read in corresponding to predictors for new outcomes that have yet to be observed. There is no need to repeat the model or unneeded transformed parameters or generated quantities. The complete follow-on program for prediction just declares the predictors in the data, the original parameters, and then the predictions in the generated quantities block.

data {
  int<lower = 0> N_tilde;
  vector[N_tilde] x_tilde;
}

parameters {
  real alpha;
  real beta;
  real<lower = 0> sigma;
}

generated quantities {
  vector[N_tilde] y_tilde
  = normal_rng(alpha + beta * x_tilde, sigma);
}

When running stand-alone generated quantities, the inputs required are the original draws for the parameters and any predictors corresponding to new predictions, and the output will be draws for \( \tilde{y} \) or derived quantities such as event probabilities.

Any posterior predictive quantities desired may be generated this way. For example, event probabilities are estimated in the usual way by defining indicator variables in the generated quantities block.
26. Simulation-Based Calibration

A Bayesian posterior is calibrated if the posterior intervals have appropriate coverage. For example, 80% intervals are expected to contain the true parameter 80% of the time. If data is generated according to a model, Bayesian posterior inference with respect to that model is calibrated by construction. Simulation-based calibration (SBC) exploits this property of Bayesian inference to assess the soundness of a posterior sampler. Roughly, the way it works is by simulating parameters according to the prior, then simulating data conditioned on the simulated parameters, then testing posterior calibration of the inference algorithm over independently simulated data sets. This chapter follows Talts et al. (2018), which improves on the original approach developed by Cook, Gelman, and Rubin (2006).

26.1. Bayes is calibrated by construction

Suppose a Bayesian model is given in the form of a prior density \( p(\theta) \) and sampling density \( p(y \mid \theta) \). Now consider a process that first simulates parameters from the prior,

\[
\theta^{\text{sim}} \sim p(\theta),
\]

and then simulates data given the parameters,

\[
y^{\text{sim}} \sim p(y \mid \theta^{\text{sim}}).
\]

By the definition of conditional densities, the simulated data and parameters constitute an independent draw from the model’s joint distribution,

\[
(y^{\text{sim}}, \theta^{\text{sim}}) \sim p(y, \theta).
\]

From Bayes’s rule, it follows that for any observed (fixed) data \( y \),

\[
p(\theta \mid y) \propto p(y, \theta).
\]

Therefore, the simulated parameters constitute a draw from the posterior for the simulated data,

\[
\theta^{\text{sim}} \sim p(\theta \mid y^{\text{sim}}).
\]

Now consider an algorithm that produces a sequence of draws from the posterior given this simulated data,

\[
\theta^{(1)}, \ldots, \theta^{(M)} \sim p(\theta \mid y^{\text{sim}}).
\]
Because $\theta^{\text{sim}}$ is also distributed as a draw from the posterior, the rank statistics of $\theta^{\text{sim}}$ with respect to $\theta^{(1)}, \ldots, \theta^{(M)}$ should be uniform.

This is one way to define calibration, because it follows that posterior intervals will have appropriate coverage (A. Philip Dawid 1982; Gneiting, Balabdaoui, and Raftery 2007). If the rank of $\theta^{\text{sim}}$ is uniform among the draws $\theta^{(1)}, \ldots, \theta^{(M)}$, then for any 90% interval selected, the probability the true value $\theta^{\text{sim}}$ falls in it will also be 90%. The same goes for any other posterior interval.

### 26.2. Simulation-based calibration

Suppose the Bayesian model to test has joint density

$$p(y, \theta) = p(y \mid \theta) \cdot p(\theta),$$

with data $y$ and parameters $\theta$ (both are typically multivariate). Simulation-based calibration works by generating $N$ simulated parameter and data pairs according to the joint density,

$$(y^{\text{sim}(1)}, \theta^{\text{sim}(1)}), \ldots, (y^{\text{sim}(N)}, \theta^{\text{sim}(N)}), \sim p(y, \theta).$$

For each simulated data set $y^{\text{sim}(n)}$, use the algorithm to be tested to generate $M$ posterior draws, which if everything is working properly, will be distributed marginally as

$$\theta^{(n,1)}, \ldots, \theta^{(n,M)} \sim p(\theta \mid y^{\text{sim}(n)}).$$

For a simulation $n$ and parameter $k$, the rank of the simulated parameter among the posterior draws is

$$r_{n,k} = \text{rank}(\theta^{\text{sim}(n)}_k, (\theta^{(n,1)}, \ldots, \theta^{(n,M)}))$$

$$= \sum_{m=1}^{M} \mathbb{I}[\theta^{(n,m)}_k < \theta^{\text{sim}(n)}_k].$$

That is, the rank is the number of posterior draws $\theta^{(n,m)}_k$ that are less than the simulated draw $\theta^{\text{sim}(n)}_k$.

If the algorithm generates posterior draws according to the posterior, the ranks should be uniformly distributed from 0 to $M$, so that the ranks plus one are uniformly distributed from 1 to $M + 1$,

$$r_{n,k} + 1 \sim \text{categorical}\left(\frac{1}{M + 1}, \ldots, \frac{1}{M + 1}\right).$$
Simulation-based calibration uses this expected behavior to test the calibration of each parameter of a model on simulated data. Talts et al. (2018) suggest plotting binned counts of $r_{1:N,k}$ for different parameters $k$; Cook, Gelman, and Rubin (2006) automate the process with a hypothesis test for uniformity.

26.3. SBC in Stan

Running simulation-based calibration in Stan will test whether Stan’s sampling algorithm can sample from the posterior associated with data generated according to the model. The data simulation and posterior fitting and rank calculation can all be done within a single Stan program. Then Stan’s posterior sampler has to be run multiple times. Each run produces a rank for each parameter being assessed for uniformity. The total set of ranks can then be tested for uniformity.

Example model

For illustration, a very simple model will suffice. Suppose there are two parameters $(\mu, \sigma)$ with independent priors,

$$\mu \sim \text{normal}(0, 1),$$

and

$$\sigma \sim \text{lognormal}(0, 1).$$

The data $y = y_1, \ldots, y_N$ is drawn conditionally independently given the parameters,

$$y_n \sim \text{normal}(\mu, \sigma).$$

The joint prior density is thus

$$p(\mu, \sigma) = \text{normal}(\mu \mid 0, 1) \cdot \text{lognormal}(\sigma \mid 0, 1),$$

and the sampling density is

$$p(y \mid \mu, \sigma) = \prod_{n=1}^{N} \text{normal}(y_n \mid \mu, \sigma).$$

For example, suppose the following two parameter values are drawn from the prior in the first simulation,

$$(\mu^{\text{sim}(1)}, \sigma^{\text{sim}(1)}) = (1.01, 0.23).$$

Then data $y^{\text{sim}(1)} \sim p(y \mid \mu^{\text{sim}(1)}, \sigma^{\text{sim}(1)})$ is drawn according to the sampling distribution. Next, $M = 4$ draws are taken from the posterior $\mu^{(1,m)}, \sigma^{(1,m)} \sim p(\mu, \sigma \mid$
Then the comparisons on which ranks are based look as follows,

<table>
<thead>
<tr>
<th>$m$</th>
<th>$\mu^{(1,m)}$</th>
<th>$\sigma^{(1,m)}$</th>
<th>$I(\mu^{(1,m)} &lt; \mu^{\text{sim}(1)})$</th>
<th>$I(\sigma^{(1,m)} &lt; \sigma^{\text{sim}(1)})$</th>
</tr>
</thead>
<tbody>
<tr>
<td>1</td>
<td>1.07</td>
<td>0.33</td>
<td>0</td>
<td>0</td>
</tr>
<tr>
<td>2</td>
<td>-0.32</td>
<td>0.14</td>
<td>1</td>
<td>1</td>
</tr>
<tr>
<td>3</td>
<td>-0.99</td>
<td>0.26</td>
<td>1</td>
<td>0</td>
</tr>
<tr>
<td>4</td>
<td>1.51</td>
<td>0.31</td>
<td>0</td>
<td>0</td>
</tr>
</tbody>
</table>

The ranks are the column sums, $r_{1,1} = 2$ and $r_{1,2} = 1$. Because the simulated parameters are distributed according to the posterior, these ranks should be distributed uniformly between 0 and $M$, the number of posterior draws.

**Testing a Stan program with simulation-based calibration**

To code simulation-based calibration in a Stan program, the transformed data block can be used to simulate parameters and data from the model. The parameters, transformed parameters, and model block then define the model over the simulated data. Then, in the generated quantities block, the program records an indicator for whether each parameter is less than the simulated value. As shown above, the rank is then the sum of the simulated indicator variables.

```stan
transformed data {
  real mu_sim = normal_rng(0, 1);
  real<lower = 0> sigma_sim = lognormal_rng(0, 1);
  int<lower = 0> J = 10;
  for (j in 1:J)
    y_sim[j] = normal_rng(mu_sim, sigma_sim);
}
parameters {
  real mu;
  real<lower = 0> sigma;
}
model {
  mu ~ normal(0, 1);
  sigma ~ lognormal(0, 1);
  y_sim ~ normal(mu, sigma);
}
To avoid confusion with the number of simulated data sets used for simulation-based calibration, \( J \) is used for the number of simulated data points.

The model is implemented twice—once as a data generating process using random number generators in the transformed data block, then again in the parameters and model block. This duplication is a blessing and a curse. The curse is that it’s more work and twice the chance for errors. The blessing is that by implementing the model twice and comparing results, the chance of there being a mistake in the model is reduced.

**Pseudocode for simulation-based calibration**

The entire simulation-based calibration process is as follows, where

- \( p(\theta) \) is the prior density
- \( p(y \mid \theta) \) is the sampling density
- \( K \) is the number of parameters
- \( N \) is the total number of simulated data sets and fits
- \( M \) is the number of posterior draws per simulated data set

\[
SBC(p(\theta), p(y \mid \theta), K, N, M)
\]

---

for (n in 1:N) {
    // simulate parameters and data
    theta(sim(n)) ~ p(\theta)
    y(sim(n)) ~ p(y \mid \theta(sim(n)))

    // posterior draws given simulated data
    for (m in 1:M)
        theta(n, m) ~ p(\theta \mid y(sim(n)))

    // calculate rank of sim among posterior draws
    for (k in 1:K)
        rank(n, k) = \sum_m I(\theta[k](n, m) < \theta[k](sim(n)))
}

// test uniformity of each parameter
for (k in 1:K)
test uniformity of \text{rank}(1:N, k)

**The importance of thinning**

The draws from the posterior are assumed to be roughly independent. If they are not, artifacts may arise in the uniformity tests due to correlation in the posterior draws. Thus it is best to think the posterior draws down to the point where the effective sample size is roughly the same as the number of thinned draws. This may require running the code a few times to judge the number of draws required to produce a target effective sample size. This operation that can be put into a loop that doubles the number of iterations until all parameters have an effective sample size of \( M \), then thinning down to \( M \) draws.

**26.4. Testing uniformity**

A simple, though not very highly powered, \( \chi^2 \)-squared test for uniformity can be formulated by binning the ranks \( 0 : M \) into \( J \) bins and testing that the bins all have roughly the expected number of draws in them. Many other tests for uniformity are possible. For example, Cook, Gelman, and Rubin (2006) transform the ranks using the inverse cumulative distribution function for the standard normal and then perform a test for normality. Talts et al. (2018) recommend visual inspection of the binned plots.

The bins don’t need to be exactly the same size. In general, if \( b_j \) is the number of ranks that fall into bin \( j \) and \( e_j \) is the number of ranks expected to fall into bin \( j \) (which will be proportional to its size under uniformity), the test statistic is

\[
X^2 = \sum_{j=1}^{J} \frac{(b_j - e_j)^2}{e_j}.
\]

The terms are approximately square standard normal, so that under the null hypothesis of uniformity,

\[
X^2 \sim \text{chiSquared}(J - 1),
\]

with the corresponding \( p \)-value given by the complementary cumulative distribution function (CCDF) of \( \text{chiSquared}(J - 1) \) applied to \( X^2 \). Because this test relies on the binomial being approximately normal, the traditional advice is to make sure the expected count in each bin is at least five, i.e., \( e_j \geq 5 \).

**Indexing to simplify arithmetic**

Because there are \( M + 1 \) possible ranks, with \( J \) bins, it is easiest to have \( M + 1 \) be divisible by \( J \). For instance, if \( J = 20 \) and \( M = 999 \), then there are 1000 possible ranks and an expected count in each bin of \( \frac{M+1}{J} = 50 \).
Distributing the ranks into bins is another fiddly operation that can be done with integer arithmetic or the floor operation. Using floor, the following function determines the bin for a rank,

\[
\text{bin}(r_{n,m}, M, J) = 1 + \left\lfloor \frac{r_{n,m}}{(M + 1)/J} \right\rfloor.
\]

For example, with \(M = 999\) and \(J = 20\), \((M + 1)/J = 50\). The lowest rank checks out,

\[
\text{bin}(0, 999, 20) = 1 + \left\lfloor \frac{0}{50} \right\rfloor = 1,
\]

as does the 50th rank,

\[
\text{bin}(49, 999, 20) = 1 + \left\lfloor \frac{49}{50} \right\rfloor = 1,
\]

and the 51st is appropriately put in the second bin,

\[
\text{bin}(50, 999, 20) = 1 + \left\lfloor \frac{50}{50} \right\rfloor = 2.
\]

The highest rank also checks out, with \(\text{bin}(1000, 999, 20) = 50\).

To summarize, the following pseudocode computes the \(b_j\) values for the \(\chi^2\) test or for visualization in a histogram.

Inputs: \(M\) draws, \(J\) bins, \(N\) parameters, ranks \(r[n, m]\)

\[
\begin{align*}
& b[1:J] = 0 \\
& \text{for } (m \text{ in } 1:M) \\
& \quad \text{++b[1 + floor}(r[n, m] * J / (M + 1))])
\end{align*}
\]

where the \(\text{++b[n]}\) notation is a common form of syntactic sugar for \(b[n] = b[n] + 1\).

In general, a great deal of care must be taken in visualizing discrete data because it’s easy to introduce off-by-one errors and artifacts at the edges because of the way boundaries are computed by default. That’s why so much attention must be devoted to indexing and binning.

### 26.5. Examples of simulation-based calibration

This section will show what the results look like when the tests pass and then when they fail. The passing test will compare a normal model and normal data generating process, whereas the second will compare a normal model with a Student-t data generating process. The first will produce calibrated posteriors, the second will not.
When things go right
Consider the following simple model for a normal distribution with standard normal and lognormal priors on the location and scale parameters.

\[
\begin{align*}
\mu & \sim \text{normal}(0, 1) \\
\sigma & \sim \text{lognormal}(0, 1) \\
y_{1:10} & \sim \text{normal}(\mu, \sigma).
\end{align*}
\]

The Stan program for evaluating SBC for this model is

```
transformed data {
  real mu_sim = normal_rng(0, 1);
  real<lower = 0> sigma_sim = lognormal_rng(0, 1);

  int<lower = 0> J = 10;
  vector[J] y_sim;
  for (j in 1:J)
    y_sim[j] = student_t_rng(4, mu_sim, sigma_sim);
}
parameters {
  real mu;
  real<lower = 0> sigma;
}
model {
  mu ~ normal(0, 1);
  sigma ~ lognormal(0, 1);

  y_sim ~ normal(mu, sigma);
}
generated quantities {
  int<lower = 0, upper = 1> I_lt_sim[2]
    = { mu < mu_sim, sigma < sigma_sim };
}
```

After running this for enough iterations so that the effective sample size is larger than \( M \), then thinning to \( M \) draws (here \( M = 999 \)), the ranks are computed and binned, and then plotted.

When things go wrong
Now consider using a Student-t data generating process with a normal model. Compare the apparent uniformity of the well specified model with the ill-specified
situation with Student-t generative process and normal model.

**When Stan’s sampler goes wrong**
The example in the previous sections show hard-coded pathological behavior. The usual application of SBC is to diagnose problems with a sampler.

This can happen in Stan with well-specified models if the posterior geometry is too difficult (usually due to extreme stiffness that varies). A simple example is the eight schools problem, the data for which consists of sample means $y_j$ and standard deviations $\sigma_j$ of differences in test score after the same intervention in $J = 8$ different schools. Donald B. Rubin (1981) applies a hierarchical model for a meta-analysis of the results, estimating the mean intervention effect and a varying effect for each school. With a standard parameterization and weak priors, this model has very challenging posterior geometry, as shown by Talts et al. (2018); this section replicates their results.

The meta-analysis model has parameters for a population mean $\mu$ and standard deviation $\tau > 0$ as well as the effect $\theta_j$ of the treatment in each school. The model has weak normal and half-normal priors for the population-level parameters,

\[
\mu \sim \text{normal}(0, 5) \\
\tau \sim \text{normal}_+(0, 5).
\]
CHAPTER 26. SIMULATION-BASED CALIBRATION

Figure 26.2: Simulation based calibration plots for location and scale of a normal model with standard normal prior on the location standard lognormal prior on the scale with mismatched generative model using a Student-t likelihood with 4 degrees of freedom. The mean histogram appears uniform, but the scale parameter shows simulated values much smaller than fit values, clearly signaling the lack of calibration.

School level effects are modeled as normal given the population parameters,

\[ \theta_j \sim \text{normal}(\mu, \tau). \]

The data is modeled as in a meta-analysis, given the school effect and sample standard deviation in the school,

\[ y_j \sim \text{normal}(\theta_j, \sigma_j). \]

This model can be coded in Stan with a data-generating process that simulates the parameters and then simulates data according to the parameters.

```stan
transformed data {
    real mu_sim = normal_rng(0, 5);
    real tau_sim = fabs(normal_rng(0, 5));
    int<lower = 0> J = 8;
    real theta_sim[J] = normal_rng(rep_vector(mu_sim, J), tau_sim);
    real<lower=0> sigma[J] = fabs(normal_rng(rep_vector(0, J), 5));
    real y[J] = normal_rng(theta_sim, sigma);
}
```
As usual for simulation-based calibration, the transformed data encodes the data-generating process using random number generators. Here, the population parameters $\mu$ and $\tau$ are first simulated, then the school-level effects $\theta$, and then finally the observed data $\sigma_j$ and $y_j$. The parameters and model are a direct encoding of the mathematical presentation using vectorized sampling statements. The generated quantities block includes indicators for parameter comparisons, saving only $\theta_1$ because the schools are exchangeable in the simulation.

When fitting the model in Stan, multiple warning messages are provided that the sampler has diverged. The divergence warnings are in Stan’s sampler precisely to diagnose the sampler’s inability to follow the curvature in the posterior and provide independent confirmation that Stan’s sampler cannot fit this model as specified.

SBC also diagnoses the problem. Here’s the rank plots for running $N = 200$ simulations with 1000 warmup iterations and $M = 999$ draws per simulation used to compute the ranks.

Although the population mean and standard deviation $\mu$ and $\tau$ appear well calibrated, $\theta_1$ tells a very different story. The simulated values are much smaller than the values fit from the data. This is because Stan’s no-U-turn sampler is unable to sample with the model formulated in the centered parameterization—the posterior geometry has regions of extremely high curvature as $\tau$ approaches zero and the $\theta_j$ become highly constrained. The chapter on reparameterization explains how to remedy this
Figure 26.3: Simulation based calibration plots for the eight-schools model with centered parameterization in Stan. The geometry is too difficult for the NUTS sampler to handle, as indicated by the plot for theta[1].

problem and fit this kind of hierarchical model with Stan.
27. Posterior and Prior Predictive Checks

Posterior predictive checks are a way of measuring whether a model does a good job of capturing relevant aspects of the data, such as means, standard deviations, and quantiles (Donald B. Rubin 1984; Andrew Gelman, Meng, and Stern 1996). Posterior predictive checking works by simulating new replicated data sets based on the fitted model parameters and then comparing statistics applied to the replicated data set with the same statistic applied to the original data set.

Prior predictive checks evaluate the prior the same way. Specifically, they evaluate what data sets would be consistent with the prior. They will not be calibrated with actual data, but extreme values help diagnose priors that are either too strong, too weak, poorly shaped, or poorly located.

Prior and posterior predictive checks are two cases of the general concept of predictive checks, just conditioning on different things (no data and the observed data, respectively). For hierarchical models, there are intermediate versions, as discussed in the section on hierarchical models and mixed replication.

27.1. Simulating from the posterior predictive distribution

The posterior predictive distribution is the distribution over new observations given previous observations. It’s predictive in the sense that it’s predicting behavior on new data that is not part of the training set. It’s posterior in that everything is conditioned on observed data $y$.

The posterior predictive distribution for replications $y_{rep}$ of the original data set $y$ given model parameters $\theta$ is defined by

$$p(y_{rep} \mid y) = \int p(y_{rep} \mid \theta) \cdot p(\theta \mid y) \, d\theta.$$ 

As with other posterior predictive quantities, generating a replicated data set $y_{rep}$ from the posterior predictive distribution is straightforward using the generated quantities block. Consider a simple regression model with parameters $\theta = (\alpha, \beta, \sigma)$.

data {
  int<lower = 0> N;
  vector[N] x;
  vector[N] y;
}
parameters {
  real alpha;
  real beta;
  real<lower = 0> sigma;
}

model {
  alpha ~ normal(0, 2);
  beta ~ normal(0, 1);
  sigma ~ normal(0, 1);
  y ~ normal(alpha + beta * x, sigma);
}

To generate a replicated data set $y_{\text{rep}}$ for this simple model, the following generated quantities block suffices.

generated quantities {
  real y_rep[N] = normal_rng(alpha + beta * x, sigma);
}

The vectorized form of the normal random number generator is used with the original predictors $x$ and the model parameters $\alpha$, $\beta$, and $\sigma$. The replicated data variable $y_{\text{rep}}$ is declared to be the same size as the original data $y$, but instead of a vector type, it is declared to be an array of reals to match the return type of the function `normal_rng`. Because the vector and real array types have the same dimensions and layout, they can be plotted against one another and otherwise compared during downstream processing.

The posterior predictive sampling for posterior predictive checks is different from usual posterior predictive sampling discussed in the chapter on posterior predictions in that the original predictors $x$ are used. That is, the posterior predictions are for the original data.

### 27.2. Plotting multiples

A standard posterior predictive check would plot a histogram of each replicated data set along with the original data set and compare them by eye. For this purpose, only a few replications are needed. These should be taken by thinning a larger set of replications down to the size needed to ensure rough independence of the replications.

Here’s a complete example where the model is a simple Poisson with a weakly informative exponential prior with a mean of 10 and standard deviation of 10.
CHAPTER 27. POSTERIOR AND PRIOR PREDICTIVE CHECKS

```stan
data {
  int<lower = 0> N;
  int<lower = 0> y[N];
}
transformed data {
  real<lower = 0> mean_y = mean(to_vector(y));
  real<lower = 0> sd_y = sd(to_vector(y));
}
parameters {
  real<lower = 0> lambda;
}
model {
  y ~ poisson(lambda);
  lambda ~ exponential(0.2);
}
generated quantities {
  int<lower = 0> y_rep[N] = poisson_rng(rep_array(lambda, N));
  real<lower = 0> mean_y_rep = mean(to_vector(y_rep));
  real<lower = 0> sd_y_rep = sd(to_vector(y_rep));
  int<lower = 0, upper = 1> mean_gte = (mean_y_rep >= mean_y);
  int<lower = 0, upper = 1> sd_gte = (sd_y_rep >= sd_y);
}
```

The generated quantities block creates a variable `y_rep` for the replicated data, variables `mean_y_rep` and `sd_y_rep` for the statistics of the replicated data, and indicator variables `mean_gte` and `sd_gte` for whether the replicated statistic is greater than or equal to the statistic applied to the original data.

Now consider generating data $y \sim \text{Poisson}(5)$. The resulting small multiples plot shows the original data plotted in the upper left and eight different posterior replications plotted in the remaining boxes.

With a Poisson data-generating process and Poisson model, the posterior replications look similar to the original data. If it were easy to pick the original data out of the lineup, there would be a problem.

Now consider generating over-dispersed data $y \sim \text{negative-binomial2}(5, 1)$. This has the same mean as Poisson$(5)$, namely 5, but a standard deviation of $\sqrt{5 + 5^2}/1 \approx 5.5$. There is no way to fit this data with the Poisson model, because a variable distributed as Poisson$(\lambda)$ has mean $\lambda$ and standard deviation $\sqrt{\lambda}$, which is $\sqrt{5}$ for Poisson$(5)$.

Here’s the resulting small multiples plot, again with original data in the upper left.
Figure 27.1: Posterior predictive checks for Poisson data generating process and Poisson model.

Figure 27.2: Posterior predictive checks for negative binomial data generating process and Poisson model.
CHAPTER 27. POSTERIOR AND PRIOR PREDICTIVE CHECKS

This time, the original data stands out in stark contrast to the replicated data sets, all of which are clearly more symmetric and lower variance than the original data. That is, the model’s not appropriately capturing the variance of the data.

27.3. Posterior “p-values’’

If a model captures the data well, summary statistics such as sample mean and standard deviation, should have similar values in the original and replicated data sets. This can be tested by means of a p-value-like statistic, which here is just the probability the test statistic \( s(\cdot) \) in a replicated data set exceeds that in the original data,

\[
\Pr [ s(\mathbf{y}_{\text{rep}}) \geq s(\mathbf{y}) | \mathbf{y} ] = \int I(s(\mathbf{y}_{\text{rep}}) \geq s(\mathbf{y}) | \mathbf{y}) \cdot p(\mathbf{y}_{\text{rep}} | \mathbf{y}) \, d\mathbf{y}_{\text{rep}}.
\]

It is important to note that “p-values’’ is in quotes because these statistics are not classically calibrated, and thus will not in general have a uniform distribution even when the model is well specified (Bayarri and Berger 2000).

Nevertheless, values of this statistic very close to zero or one are cause for concern that the model is not fitting the data well. Unlike a visual test, this p-value-like test is easily automated for bulk model fitting.

To calculate event probabilities in Stan, it suffices to define indicator variables that take on value 1 if the event occurs and 0 if it does not. The posterior mean is then the event probability. For efficiency, indicator variables are defined in the generated quantities block.

```stan
generated quantities {
  int<lower = 0, upper = 1> mean_gt;
  int<lower = 0, upper = 1> sd_gt;
  {
    real y_rep[N] = normal_rng(alpha + beta * x, sigma);
    mean_gt = mean(y_rep) > mean(y);
    sd_gt = sd(y_rep) > sd(y);
  }
}
```

The indicator variable `mean_gt` will have value 1 if the mean of the simulated data `y_rep` is greater than or equal to the mean of the original data `y`. Because the values of `y_rep` are not needed for the posterior predictive checks, the program saves output space by using a local variable for `y_rep`. The statistics `\( \text{mean}(\mathbf{u}) \)` and `\( \text{sd}(\mathbf{y}) \)` could also be computed in the transformed data block and saved.
For the example in the previous section, where over-dispersed data generated by a negative binomial distribution was fit with a simple Poisson model, the following plot illustrates the posterior p-value calculation for the mean statistic.

![Figure 27.3: Histogram of means of replicated data sets; vertical red line at mean of original data.](image)

The p-value for the mean is just the percentage of replicated data sets whose statistic is greater than or equal that of the original data. Using a Poisson model for negative binomial data still fits the mean well, with a posterior p-value of 0.49. In Stan terms, it is extracted as the posterior mean of the indicator variable `mean_gt`.

The standard deviation statistic tells a different story.

![Figure 27.4: Scatterplot of standard deviations of replicated data sets; the vertical red line is at standard deviation of original data.](image)

Here, the original data has much higher standard deviation than any of the replicated data sets. The resulting p-value estimated by Stan after a large number of iterations
is exactly zero (the absolute error bounds are fine, but a lot of iterations are required to get good relative error bounds on small p-values by sampling). In other words, there were no posterior draws in which the replicated data set had a standard deviation greater than or equal to that of the original data set. Clearly, the model is not capturing the dispersion of the original data. The point of this exercise isn’t just to figure out that there’s a problem with a model, but to isolate where it is. Seeing that the data is over-dispersed compared to the Poisson model would be reason to fit a more general model like the negative binomial or a latent varying effects (aka random effects) model that can account for the over-dispersion.

**Which statistics to test?**
Any statistic may be used for the data, but these can be guided by the quantities of interest in the model itself. Popular choices in addition to mean and standard deviation are quantiles, such as the median, 5% or 95% quantiles, or even the maximum or minimum value to test extremes.

Despite the range of choices, test statistics should ideally be ancillary, in the sense that they should be testing something other than the fit of a parameter. For example, a simple normal model of a data set will typically fit the mean and variance of the data quite well as long as the prior doesn’t dominate the posterior. In contrast, a Poisson model of the same data cannot capture both the mean and the variance of a data set if they are different, so they bear checking in the Poisson case. As we saw with the Poisson case, the posterior mean for the single rate parameter was located near the data mean, not the data variance. Other distributions such as the lognormal and gamma distribution, have means and variances that are functions of two or more parameters.

### 27.4. Prior predictive checks
Prior predictive checks generate data according to the prior in order to assess whether a prior is appropriate (Gabry et al. 2019). A posterior predictive check generates replicated data according to the posterior predictive distribution. In contrast, the prior predictive check generates data according to the prior predictive distribution,

\[ y^{\text{sim}} \sim p(y). \]

The prior predictive distribution is just like the posterior predictive distribution with no observed data, so that a prior predictive check is nothing more than the limiting case of a posterior predictive check with no data.

This is easy to carry out mechanically by simulating parameters

\[ \theta^{\text{sim}} \sim p(\theta) \]
according to the priors, then simulating data
\[ y_{\text{sim}} \sim p(y \mid \theta_{\text{sim}}) \]
according to the sampling distribution given the simulated parameters. The result is a simulation from the joint distribution,
\[ (y_{\text{sim}}, \theta_{\text{sim}}) \sim p(y, \theta) \]
and thus
\[ y_{\text{sim}} \sim p(y) \]
is a simulation from the prior predictive distribution.

**Coding prior predictive checks in Stan**
A prior predictive check is coded just like a posterior predictive check. If a posterior predictive check has already been coded and it’s possible to set the data to be empty, then no additional coding is necessary. The disadvantage to coding prior predictive checks as posterior predictive checks with no data is that Markov chain Monte Carlo will be used to sample the parameters, which is less efficient than taking independent draws using random number generation.

Prior predictive checks can be coded entirely within the generated quantities block using random number generation. The resulting draws will be independent. Predictors must be read in from the actual data set—they do not have a generative model from which to be simulated. For a Poisson regression, prior predictive sampling can be encoded as the following complete Stan program.

```stan
data {
  int<lower = 0> N;
  vector[N] x;
}
generated quantities {
  real alpha = normal_rng(0, 1);
  real beta = normal_rng(0, 1);
  real y_sim[N] = poisson_log_rng(alpha + beta * x);
}
```

Running this program using Stan’s fixed-parameter sampler yields draws from the prior. These may be plotted to consider their appropriateness.

### 27.5. Example of prior predictive checks
Suppose we have a model for a football (aka soccer) league where there are \( J \) teams. Each team has a scoring rate \( \lambda_j \) and in each game will be assumed to score
poisson($\lambda_j$) points. Yes, this model completely ignores defense. Suppose the modeler does not want to “put their thumb on the scale” and would rather “let the data speak for themselves” and so uses a prior with very wide tails, because it seems uninformative, such as the widely deployed

$$\lambda_j \sim \text{gamma}(\epsilon_1, \epsilon_2).$$

This is not just a manufactured example; *The BUGS Book* recommends setting $\epsilon = (0.5, 0.00001)$, which corresponds to a Jeffreys prior for a Poisson rate parameter prior (Lunn et al. 2012, 85).

Suppose the league plays a round-robin tournament wherein every team plays every other team. The following Stan model generates random team abilities and the results of such a round-robin tournament, which may be used to perform prior predictive checks.

```stan
data {
  int<lower = 0> J;
  real<lower = 0> epsilon[2];
}
generated quantities {
  real<lower = 0> lambda[J];
  int y[J, J];
  for (j in 1:J) lambda[j] = gamma_rng(epsilon[1], epsilon[2]);
  for (i in 1:J)
    for (j in 1:J)
      y[i, j] = poisson_rng(lambda[i]) - poisson_rng(lambda[j]);
}
```

In this simulation, teams play each other twice and play themselves once. This could be made more realistic by controlling the combinatorics to only generate a single result for each pair of teams, of which there are $\binom{J}{2} = \frac{J \cdot (J-1)}{2}$.

Using the gamma$(0.5, 0.00001)$ reference prior on team abilities, the following are the first 20 simulated point differences for the match between the first two teams, $y_{1,2}^{(1:20)}$.

<p>| | | | | | | | | | |</p>
<table>
<thead>
<tr>
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</tr>
</thead>
<tbody>
<tr>
<td>2597</td>
<td>-26000</td>
<td>5725</td>
<td>22496</td>
<td>1270</td>
<td>1072</td>
<td>4502</td>
<td>-2809</td>
<td>-302</td>
<td>4987</td>
</tr>
<tr>
<td>7513</td>
<td>7527</td>
<td>-3268</td>
<td>-12374</td>
<td>3828</td>
<td>-158</td>
<td>-29889</td>
<td>2986</td>
<td>-1392</td>
<td>66</td>
</tr>
</tbody>
</table>

That’s some pretty highly scoring football games being simulated; all but one has a score differential greater than 100! In other words, this gamma$(0.5, 0.00001)$ prior is putting around 95% of its weight on score differentials above 100. Given that
two teams combined rarely score 10 points, this prior is way out of line with prior knowledge about football matches; it is not only consistent with outcomes that have never occurred in the history of the sport, it puts most of the prior probability mass there.

The posterior predictive distribution can be strongly affected by the prior when there is not much observed data and substantial prior mass is concentrated around infeasible values (A. Gelman 2006).

Just as with posterior predictive distributions, any statistics of the generated data may be evaluated. Here, the focus was on score difference between a single pair of teams, but it could’ve been on maximums, minimums, averages, variances, etc.

In this textbook example, the prior is univariate and directly related to the expected number of points scored, and could thus be directly inspected for consistency with prior knowledge about scoring rates in football. There will not be the same kind of direct connection when the prior and sampling distributions are multivariate. In these more challenging situations, prior predictive checks are an easy way to get a handle on the implications of a prior in terms of what it says the data is going to look like; for a more complex application involving spatially heterogeneous air pollution concentration, see (Gabry et al. 2019).

Prior predictive checks can also be compared with the data, but one should not expect them to be calibrated in the same way as posterior predictive checks. That would require guessing the posterior and encoding it in the prior. The goal is make sure the prior is not so wide that it will pull probability mass away from feasible values.

### 27.6. Mixed predictive replication for hierarchical models

Andrew Gelman, Meng, and Stern (1996) discuss the case of mixed replication for hierarchical models in which the hyperparameters remain fixed, but varying effects are replicated. This is neither a purely prior nor purely posterior predictive check, but falls somewhere in between.

For example, consider a simple varying intercept logistic regression, with intercepts $\alpha_k$ for $k \in 1:K$. Each data item $y_n \in \{0, 1\}$ is assumed to correspond to group $k k_n \in 1:K$. The sampling distribution is thus

$$y_n \sim \text{bernoulli}(\text{logit}^{-1}(\alpha_{kk[n]}))$$

The varying intercepts have a hierarchical normal prior,

$$\alpha_k \sim \text{normal}(\mu, \sigma).$$
The hyperparameters are themselves given weakly informative priors,

\[
\mu \sim \text{normal}(0, 2) \\
\sigma \sim \text{lognormal}(0, 1).
\]

Like in a posterior predictive check, the hyperparameters \(\mu\) and \(\sigma\) are drawn from the posterior,

\[
\mu^{(m)}, \sigma^{(m)} \sim p(\mu, \sigma \mid y)
\]

Like in a prior predictive check, replicated values of \(\alpha\) are drawn from the hyperparameters,

\[
\alpha_{k}^{\text{rep}(m)} \sim \text{normal}(\alpha_{k} \mid \mu^{(m)}, \sigma^{(m)}).
\]

The data items are then each replicated using the replicated intercepts,

\[
y_{n}^{\text{rep}(m)} \sim \text{bernoulli} \left( \text{logit}^{-1}(\alpha_{kk[n]}^{\text{rep}(m)}) \right).
\]

Thus the \(y^{\text{rep}(m)}\) can be seen as a kind of posterior predictive replication of observations from new groups that were not among the original \(K\) groups.

In Stan, mixed predictive replications \(y^{\text{rep}(m)}\) can be programmed directly.

```stan
data {
  int<lower = 0> K;
  int<lower = 0> N;
  int<lower = 1, upper = K> kk[N];
  int<lower = 0, upper = 1> y[N];
}
parameters {
  real mu;
  real<lower = 0> sigma;
  vector<offset = mu, multiplier = sigma>[K] alpha;
}
model {
  mu ~ normal(0, 2); // hyperprior
  sigma ~ lognormal(0, 1);
  alpha ~ normal(mu, sigma); // hierarchical prior
  y ~ bernoulli_logit(alpha[kk]); // sampling distribution
}
generated quantities {
  // alpha replicated; mu and sigma not replicated
```
real alpha_rep[K]
    = normal_rng(rep_vector(mu, K), sigma);
int<lower = 0, upper = 1> y_rep[N]
    = bernoulli_logit_rng(alpha_rep[kk]);
}

27.7. Joint model representation
Following Andrew Gelman, Meng, and Stern (1996), prior, posterior, and mixed replications may all be defined as posteriors from joint models over parameters and observed and replicated data.

Posterior predictive model
For example, posterior predictive replication may be formulated using sampling notation as follows.

\[
\begin{align*}
\theta & \sim p(\theta) \\
y & \sim p(y | \theta) \\
y^{rep} & \sim p(y | \theta)
\end{align*}
\]

The heavily overloaded sampling notation is meant to indicate that both \(y\) and \(y^{rep}\) are drawn from the same distribution, or more formally using capital letters to distinguish random variables, that the conditional densities \(p_{Y^{rep} | \Theta}\) and \(p_{Y | \Theta}\) are the same.

The joint density is

\[
p(\theta, y, y^{rep}) = p(\theta) \cdot p(y | \theta) \cdot p(y^{rep} | \theta).
\]

This again is assuming that the two distributions for \(y\) and \(y^{rep}\) are identical.

The variable \(y\) is observed, with the predictive simulation \(y^{rep}\) and parameter vector \(\theta\) not observed. The posterior is \(p(y^{rep}, \theta | y)\). Given draws from the posterior, the posterior predictive simulations \(y^{rep}\) are retained.

Prior predictive model
The prior predictive model simply drops the data component of the posterior predictive model.

\[
\begin{align*}
\theta & \sim p(\theta) \\
y^{rep} & \sim p(y | \theta)
\end{align*}
\]
This corresponds to the joint density

\[ p(\theta, y^{\text{rep}}) = p(\theta) \cdot p(y^{\text{rep}} | \theta). \]

It is typically straightforward to draw \( \theta \) from the prior and \( y^{\text{rep}} \) from the sampling distribution given \( \theta \) efficiently. In cases where it is not, the model may be coded and executed just as the posterior predictive model, only with no data.

**Mixed replication for hierarchical models**

The mixed replication corresponds to the model

\[
\begin{align*}
\phi & \sim p(\phi) \\
\alpha & \sim p(\alpha | \phi) \\
y & \sim p(y | \alpha) \\
\alpha^{\text{rep}} & \sim p(\alpha | \phi) \\
y^{\text{rep}} & \sim p(y | \phi)
\end{align*}
\]

The notation here is meant to indicate that \( \alpha \) and \( \alpha^{\text{rep}} \) have identical distributions, as do \( y \) and \( y^{\text{rep}} \).

This corresponds to a joint model

\[
p(\phi, \alpha, \alpha^{\text{rep}}, y, y^{\text{rep}}) = p(\phi) \cdot p(\alpha | \phi) \cdot p(y | \alpha) \cdot p(\alpha^{\text{rep}} | \phi) \cdot p(y^{\text{rep}} | \alpha^{\text{rep}}),
\]

where \( y \) is the only observed variable, \( \alpha \) contains the lower-level parameters and \( \phi \) the hyperparameters. Note that \( \phi \) is not replicated and instead appears in the distribution for both \( \alpha \) and \( \alpha^{\text{rep}} \).

The posterior is \( p(\phi, \alpha, \alpha^{\text{rep}}, y^{\text{rep}} | y) \). From posterior draws, the posterior predictive simulations \( y^{\text{rep}} \) are kept.
28. Held-Out Evaluation and Cross-Validation

Held-out evaluation involves splitting a data set into two parts, a training data set and a test data set. The training data is used to estimate the model and the test data is used for evaluation. Held-out evaluation is commonly used to declare winners in predictive modeling competitions such as those run by Kaggle.

Cross-validation involves repeated held-out evaluations performed by partitioning a single data set in different ways. The training/test split can be done either by randomly selecting the test set, or by partitioning the data set into several equally-sized subsets and then using each subset in turn as the test data with the other folds as training data.

Held-out evaluation and cross-validation may involve any kind of predictive statistics, with common choices being the predictive log density on test data, squared error of parameter estimates, or accuracy in a classification task.

28.1. Evaluating posterior predictive densities
Given training data \((x, y)\) consisting of parallel sequences of predictors and observations and test data \((\tilde{x}, \tilde{y})\) of the same structure, the posterior predictive density is

\[
p(\tilde{y} \mid \tilde{x}, x, y) = \int p(\tilde{y} \mid \tilde{x}, \theta) \cdot p(\theta \mid x, y) \, d\theta,
\]

where \(\theta\) is the vector of model parameters. This predictive density is the density of the test observations, conditioned on both the test predictors \(\tilde{x}\) and the training data \((x, y)\).

This integral may be calculated with Monte Carlo methods as usual,

\[
p(\tilde{y} \mid \tilde{x}, x, y) \approx \frac{1}{M} \sum_{m=1}^{M} p(\tilde{y} \mid \tilde{x}, \theta^{(m)}),
\]

where the \(\theta^{(m)} \sim p(\theta \mid x, y)\) are draws from the posterior given only the training data \((x, y)\).

To avoid underflow in calculations, it will be more stable to compute densities on the log scale. Taking the logarithm and pushing it through results in a stable
computation,

\[
\log p(\tilde{y} \mid \tilde{x}, x, y) \approx \log \frac{1}{M} \sum_{m=1}^{M} p(\tilde{y} \mid \tilde{x}, \theta^{(m)}),
\]

\[
= - \log M + \log \sum_{m=1}^{M} p(\tilde{y} \mid \tilde{x}, \theta^{(m)}),
\]

\[
= - \log M + \log \sum_{m=1}^{M} \exp(\log p(\tilde{y} \mid \tilde{x}, \theta^{(m)}))
\]

\[
= - \log M + \log\text{-sum-exp}_{m=1}^{M} \log p(\tilde{y} \mid \tilde{x}, \theta^{(m)})
\]

where the log sum of exponentials function is defined so as to make the above equation hold,

\[
\log\text{-sum-exp}_{m=1}^{M} \mu_m = \log \sum_{m=1}^{M} \exp(\mu_m).
\]

The log sum of exponentials function can be implemented so as to avoid underflow and maintain high arithmetic precision as

\[
\log\text{-sum-exp}_{m=1}^{M} \mu_m = \max(\mu) + \sum_{m=1}^{M} \exp(\mu_m - \max(\mu)).
\]

Pulling the maximum out preserves all of its precision. By subtracting the maximum, the terms \(\mu_m - \max(\mu) \leq 0\), and thus will not overflow.

**Stan program**

To evaluate the log predictive density of a model, it suffices to implement the log predictive density of the test data in the generated quantities block. The log sum of exponentials calculation must be done on the outside of Stan using the posterior draws of \(\log p(\tilde{y} \mid \tilde{x}, \theta^{(m)})\).

Here is the code for evaluating the log posterior predictive density in a simple linear regression of the test data \(\tilde{y}\) given predictors \(\tilde{x}\) and training data \((x, y)\).

```stan
data {
  int<lower = 0> N;
  vector[N] y;
  vector[N] x;
  int<lower = 0> N_tilde;
```
vector[N_tilde] x_tilde;
vector[N_tilde] y_tilde;
}
parameters {
  real alpha;
  real beta;
  real<lower = 0> sigma;
}
model {
  y ~ normal(alpha + beta * x, sigma);
}
generated quantities {
  real log_p = normal_lpdf(y_tilde | alpha + beta * x_tilde, sigma);
}

Only the training data \( x \) and \( y \) are used in the model block. The test data \( y_{\text{tilde}} \) and test predictors \( x_{\text{tilde}} \) appear in only the generated quantities block. Thus the program is not cheating by using the test data during training. Although this model does not do so, it would be fair to use \( x_{\text{tilde}} \) in the model block—only the test observations \( y_{\text{tilde}} \) are unknown before they are predicted.

Given \( M \) posterior draws from Stan, the sequence \( \log_p[1:M] \) will be available, so that the log posterior predictive density of the test data given training data and predictors is just \( \log_{\text{sum}}(\log_p) - \log(M) \).

### 28.2. Estimation error

**Parameter estimates**

Estimation is usually considered for unknown parameters. If the data from which the parameters were estimated came from simulated data, the true value of the parameters may be known. If \( \theta \) is the true value and \( \hat{\theta} \) the estimate, then error is just the difference between the prediction and the true value,

\[
\text{err} = \hat{\theta} - \theta.
\]

If the estimate is larger than the true value, the error is positive, and if it’s smaller, then error is negative. If an estimator’s unbiased, then expected error is zero. So typically, absolute error or squared error are used, which will always have positive expectations for an imperfect estimator. **Absolute error** is defined as

\[
\text{abs-err} = |\hat{\theta} - \theta|
\]
and squared error as
\[
\text{sq-err} = \left( \hat{\theta} - \theta \right)^2.
\]

Gneiting and Raftery (2007) provide a thorough overview of such scoring rules and their properties.

Bayesian posterior means minimize expected square error, whereas posterior medians minimize expected absolute error. Estimates based on modes rather than probability, such as (penalized) maximum likelihood estimates or maximum a posterior estimates, do not have these properties.

**Predictive estimates**

In addition to parameters, other unknown quantities may be estimated, such as the score of a football match or the effect of a medical treatment given to a subject. In these cases, square error is defined in the same way. If there are multiple exchangeable outcomes being estimated, \(z_1, \ldots, z_N\), then it is common to report mean square error (MSE),

\[
\text{mse} = \frac{1}{N} \sum_{n=1}^{N} \left( \hat{z}_n - z_n \right)^2.
\]

To put the error back on the scale of the original value, the square root may be applied, resulting in what is known prosaically as root mean square error (RMSE),

\[
\text{rmse} = \sqrt{\text{mean-sq-err}}.
\]

**Predictive estimates in Stan**

Consider a simple linear regression model, parameters for the intercept \(\alpha\) and slope \(\beta\), along with predictors \(\tilde{x}_n\). The standard Bayesian estimate is the expected value of \(\tilde{y}\) given the predictors and training data,

\[
\tilde{y}_n = \mathbb{E}[\tilde{y}_n \mid \tilde{x}_n, x, y] \\
\approx \frac{1}{M} \sum_{m=1}^{M} \tilde{y}_n^{(m)}
\]

where \(\tilde{y}_n^{(m)}\) is drawn from the sampling distribution

\[
\tilde{y}_n^{(m)} \sim p(\tilde{y}_n \mid \tilde{x}_n, \alpha^{(m)}, \beta^{(m)}),
\]

for parameters \(\alpha^{(m)}\) and \(\beta^{(m)}\) drawn from the posterior,

\[
(\alpha^{(m)}, \beta^{(m)}) \sim p(\alpha, \beta \mid x, y).
\]
In the linear regression case, two stages of simplification can be carried out, the first of which helpfully reduces the variance of the estimator. First, rather than averaging samples \( \tilde{y}_n^{(m)} \), the same result is obtained by averaging linear predictions,

\[
\hat{y}_n = \mathbb{E}[\alpha + \beta \cdot \tilde{x}_n | \tilde{x}_n, x, y] \\
\approx \frac{1}{M} \sum_{m=1}^{M} \alpha^{(m)} + \beta^{(m)} \cdot \tilde{x}_n
\]

This is possible because

\[
\tilde{y}_n^{(m)} \sim \text{normal}(\tilde{y}_n | \alpha^{(m)} + \beta^{(m)} \cdot \tilde{x}_n, \sigma^{(m)}),
\]

and the normal distribution has symmetric error so that the expectation of \( \tilde{y}_n^{(m)} \) is the same as \( \alpha^{(m)} + \beta^{(m)} \cdot \tilde{x}_n \). Replacing the sampled quantity \( \tilde{y}_n^{(m)} \) with its expectation is a general variance reduction technique for Monte Carlo estimates known as \textit{Rao-Blackwellization} (Rao 1945; Blackwell 1947).

In the linear case, because the predictor is linear in the coefficients, the estimate can be further simplified to use the estimated coefficients,

\[
\hat{y}_n \approx \frac{1}{M} \sum_{m=1}^{M} \left( \alpha^{(m)} + \beta^{(m)} \cdot \tilde{x}_n \right)
\]

\[
= \frac{1}{M} \sum_{m=1}^{M} \alpha^{(m)} + \frac{1}{M} \sum_{m=1}^{M} (\beta^{(m)} \cdot \tilde{x}_n)
\]

\[
= \frac{1}{M} \sum_{m=1}^{M} \alpha^{(m)} + \left( \frac{1}{M} \sum_{m=1}^{M} \beta^{(m)} \right) \cdot \tilde{x}_n
\]

\[
= \hat{\alpha} + \hat{\beta} \cdot \tilde{x}_n.
\]

In Stan, only the first of the two steps (the important variance reduction step) can be coded in the object model. The linear predictor is defined in the generated quantities block.

data {
  int<lower = 0> N_tilde;
  vector[N_tilde] x_tilde;
  ...
}
generated quantities {
  vector[N_tilde] tilde_y = alpha + beta * x_tilde;

The posterior mean of \( \tilde{y} \) calculated by Stan is the Bayesian estimate \( \hat{y} \). The posterior median may also be calculated and used as an estimate, though square error and the posterior mean are more commonly reported.

### 28.3. Cross-validation

Cross-validation involves choosing multiple subsets of a data set as the test set and using the other data as training. This can be done by partitioning the data and using each subset in turn as the test set with the remaining subsets as training data. A partition into ten subsets is common to reduce computational overhead. In the limit, when the test set is just a single item, the result is known as leave-one-out (LOO) cross-validation (Vehtari, Gelman, and Gabry 2017).

Partitioning the data and reusing the partitions is very fiddly in the indexes and may not lead to even divisions of the data. It’s far easier to use random partitions, which support arbitrarily sized test/training splits and can be easily implemented in Stan. The drawback is that the variance of the resulting estimate is higher than with a balanced block partition.

**Stan implementation with random folds**

For the simple linear regression model, randomized cross-validation can be implemented in a single model. To randomly permute a vector in Stan, the simplest approach is the following.

```stan
functions {
  int[] permutation_rng(int N) {
    int N = rows(x);
    int y[N];
    for (n in 1:N)
      y[n] = n;
    vector[N] theta = rep_vector(1.0 / N, N);
    for (n in 1:rows(y))
      int i = categorical_rng(theta);
      int temp = y[n];
      y[n] = y[i];
      y[i] = temp;
    }
    return y;
  }
}
```

The name of the function must end in \_rng because it uses other random functions internally. This will restrict its usage to the transformed data and generated quantities.
block. The code walks through the vector exchanging each item with another randomly chosen item, resulting in a uniformly drawn permutation of the integers 1:N.\(^1\)

The transformed data block uses the permutation RNG to generate training data and test data by taking prefixes and suffixes of the permuted data.

```stan
data {
  int<lower = 0> N;
  vector[N] x;
  vector[N] y;
  int<lower = 0, upper = N> N_test;
}
transformed data {
  int N_train = N - N_test;
  int permutation[N] = permutation_rng(N);
  vector[N_train] x_train = x[permutation[1 : N_train]];
  vector[N_train] y_train = y[permutation[1 : N_train]];
  vector[N_test] x_test = x[permutation[N_train + 1 : N]];
  vector[N_test] y_test = y[permutation[N_train + 1 : N]];
}
```

Recall that in Stan, `permutation[1:N_train]` is an array of integers, so that `x[permutation[1 : N_train]]` is a vector defined for `i` in `1:N_train` by

\[
x[\text{permutation}[1 : \text{N_train}]][i] = x[\text{permutation}[1:\text{N_train}][i]] = x[\text{permutation}[i]]
\]

Given the test/train split, the rest of the model is straightforward.

```stan
parameters {
  real alpha;
  real beta;
  real<lower = 0> sigma;
}
model {
  y_train ~ normal(alpha + beta * x_train, sigma);
  { alpha, beta, sigma } ~ normal(0, 1);
}
```

\(^1\)The traditional approach is to walk through a vector and replace each item with a random element from the remaining elements, which is guaranteed to only move each item once. This was not done here as it'd require new categorical `theta` because Stan does not have a uniform discrete RNG built in.
generated quantities {
    vector[N] y_test_hat = normal_rng(alpha + beta * x_test, sigma);
    vector[N] err = y_test_sim - y_hat;
}

The prediction $y_{\text{test}}$ is defined in the generated quantities block using the general form involving all uncertainty. The posterior of this quantity corresponds to using a posterior mean estimator,

$$
\hat{y}_{test} = \mathbb{E} [y_{\text{test}} \mid x_{\text{test}}, x_{\text{train}}, y_{\text{train}}]
\approx \frac{1}{M} \sum_{m=1}^{M} \hat{y}_{\text{test}(m)}.
$$

Because the test set is constant and the expectation operator is linear, the posterior mean of $\text{err}$ as defined in the Stan program will be the error of the posterior mean estimate,

$$
\hat{y}_{\text{test}} - y_{\text{test}} = \mathbb{E} [\hat{y}_{\text{test}} - y_{\text{test}} \mid x_{\text{test}}, x_{\text{train}}, y_{\text{train}}] - y_{\text{test}}
\approx \frac{1}{M} \sum_{m=1}^{M} \hat{y}_{\text{test}(m)} - y_{\text{test}},
$$

where

$$
\hat{y}_{\text{test}(m)} \sim p(y \mid x_{\text{test}}, x_{\text{train}}, y_{\text{train}}).
$$

This just calculates error; taking absolute value or squaring will compute absolute error and mean square error. Note that the absolute value and square operation should not be done within the Stan program because neither is a linear function and the result of averaging squares is not the same as squaring an average in general.

Because the test set size is chosen for convenience in cross-validation, results should be presented on a per-item scale, such as average absolute error or root mean square error, not on the scale of error in the fold being evaluated.

**User-defined permutations**

It is straightforward to declare the variable `permutation` in the data block instead of the transformed data block and read it in as data. This allows an external program to control the blocking, allowing non-random partitions to be evaluated.
Cross-validation with structured data
Cross-validation must be done with care if the data is inherently structured. For example, in a simple natural language application, data might be structured by document. For cross-validation, one needs to cross-validate at the document level, not at the individual word level. This is related to mixed replication in posterior predictive checking, where there is a choice to simulate new elements of existing groups or generate entirely new groups.

Education testing applications are typically grouped by school district, by school, by classroom, and by demographic features of the individual students or the school as a whole. Depending on the variables of interest, different structured subsets should be evaluated. For example, the focus of interest may be on the performance of entire classrooms, so it would make sense to cross-validate at the class or school level on classroom performance.

Cross-validation with spatio-temporal data
Often data measurements have spatial or temporal properties. For example, home energy consumption varies by time of day, day of week, on holidays, by season, and by ambient temperature (e.g., a hot spell or a cold snap). Cross-validation must be tailored to the predictive goal. For example, in predicting energy consumption, the quantity of interest may be the prediction for next week’s energy consumption given historical data and current weather covariates. This suggests an alternative to cross-validation, wherein individual weeks are each tested given previous data. This often allows comparing how well prediction performs with more or less historical data.

Approximate cross-validation
Vehtari, Gelman, and Gabry (2017) introduce a method that approximates the evaluation of leave-one-out cross validation inexpensively using only the data point log likelihoods from a single model fit. This method is documented and implemented in the R package loo (Gabry et al. 2019).
29. Poststratification

Stratification is a technique developed for survey sampling in which a population is partitioned into subgroups (i.e., stratified) and each group (i.e., stratum) is sampled independently. If the subgroups are more homogeneous (i.e., lower variance) than the population as a whole, this can reduce variance in the estimate of a quantity of interest at the population level.

Poststratification is a technique for adjusting a non-representative sample (i.e., a convenience sample or other observational data) for which there are demographic predictors characterizing the strata. It is carried out after a model is fit to the observed data, hence the name poststratification (Little 1993). Poststratification can be fruitfully combined with regression modeling (or more general parametric modeling), which provides estimates based on combinations of predictors (or general parameters) rather than raw counts in each stratum. Multilevel modeling is useful in determining how much partial pooling to apply in the regressions, leading to the popularity of the combination of multilevel regression and poststratification (MRP) (Park, Gelman, and Bafumi 2004).

29.1. Some examples

Earth science
Stratification and poststratification can be applied to many applications beyond survey sampling (Kennedy and Gelman 2019). For example, large-scale whole-earth soil-carbon models are fit with parametric models of how soil-carbon depends on features of an area such as soil composition, flora, fauna, temperature, humidity, etc. Given a model that predicts soil-carbon concentration given these features, a whole-earth model can be created by stratifying the earth into a grid of say 10km by 10km “squares” (they can’t literally be square because the earth’s surface is topologically a sphere). Each grid area has an estimated makeup of soil type, forestation, climate, etc. The global level of soil carbon is then estimated using poststratification by simply summing the expected soil carbon estimated for each square in the grid (Paustian et al. 1997). Dynamic models can then be constructed by layering a time-series component, varying the poststratification predictors over time, or both (Field et al. 1998).
Polling
Suppose a university’s administration would like to estimate the support for a given proposal among its students. A poll is carried out in which 490 respondents are undergraduates, 112 are graduate students, and 47 are continuing education students. Now suppose that support for the issue among the poll respondents is is 25% among undergraduate students (subgroup 1), 40% among graduate students (subgroup 2), and 80% among continuing education students (subgroup 3). Now suppose that the student body is made up of 20,000 undergraduates, 5,000 graduate students, and 2,000 continuing education students. It is important that our subgroups are exclusive and exhaustive, i.e., they form a partition of the population.

The proportion of support in the poll among students in each group provides a simple maximum likelihood estimate \( \theta^* = (0.25, 0.5, 0.8) \) of support in each group for a simple Bernoulli model where student \( n \)'s vote is modeled as

\[
y_n \sim \text{bernoulli}(\theta_{jj[n]}),
\]

where \( jj[n] \in 1:3 \) is the subgroup to which the \( n \)-th student belongs.

An estimate of the population prevalence of support for the issue among students can be constructed by simply multiplying estimated support in each group by the size of each group. Letting \( N = (20000, 5000, 2000) \) be the subgroup sizes, the poststratified estimate of support in the population \( \phi^* \) is estimated by

\[
\phi^* = \frac{\sum_{j=1}^{3} \theta^*_j \cdot N_j}{\sum_{j=1}^{3} N_j}.
\]

Plugging in our estimates and population counts yields

\[
\phi^* = \frac{0.25 \cdot 20000 + 0.4 \cdot 5000 + 0.8 \cdot 2000}{20000 + 5000 + 2000} = \frac{8600}{27000} \approx 0.32.
\]

29.2. Bayesian poststratification
Considering the same polling data from the previous section in a Bayesian setting, the uncertainty in the estimation of subgroup support is pushed through predictive inference in order to get some idea of the uncertainty of estimated support.
Continuing the example of the previous section, the likelihood remains the same,

\[ y_n \sim \text{bernoulli}(\theta_{jj[n]}), \]

where \( jj[n] \in 1:J \) is the group to which item \( n \) belongs and \( \theta_j \) is the proportion of support in group \( j \).

This can be reformulated from a Bernoulli model to a binomial model in the usual way. Letting \( A_j \) be the number of respondents in group \( j \) and \( a_j \) be the number of positive responses in group \( j \), the likelihood may be reduced to the form

\[ a_j \sim \text{binomial}(A_j, \theta_j). \]

A simple uniform prior on the proportion of support in each group completes the model,

\[ \theta_j \sim \text{beta}(1, 1). \]

A more informative prior could be used if there is prior information available about support among the student body.

Using sampling, draws \( \theta^{(m)} \sim p(\theta | y) \) from the posterior may be combined with the population sizes \( N \) to estimate \( \phi \), the proportion of support in the population,

\[ \phi^{(m)} = \frac{\sum_{j=1}^{J} \theta^{(m)}_j \cdot N_j}{\sum_{j=1}^{J} N_j}. \]

The posterior draws for \( \phi^{(m)} \) characterize expected support for the issue in the entire population. These draws may be used to estimate expected support (the average of the \( \phi^{(m)} \)), posterior intervals (quantiles of the \( \phi^{(m)} \)), or to plot a histogram.

### 29.3. Poststratification in Stan

The maximum likelihood and Bayesian estimates can be handled with the same Stan program. The model of individual votes is collapsed to a binomial, where \( A_j \) is the number of voters from group \( j \), \( a_j \) is the number of positive responses from group \( j \), and \( N_j \) is the size of group \( j \) in the population.

```stan
data {
  int<lower = 1> J;
  int<lower = 0> A[J];
  int<lower = 0> a[J];
```
vector<lower = 0>[J] N;
}
parameters {
  vector<lower = 0, upper = 1>[J] theta;
}
model {
  a ~ binomial(A, theta);
}
generated quantities {
  real<lower = 0, upper = 1> phi = dot(N, theta) / sum(N);
}

The likelihood is vectorized, and implicitly sums over the $j$. The prior is implicitly uniform on $(0, 1)$, the support of $\theta$. The summation is computed using a dot product and the sum function, which is why $N$ was declared as a vector rather than as an array of integers.

29.4. Regression and poststratification

In applications to polling, there are often numerous demographic features like age, gender, income, education, state of residence, etc. If each of these demographic features induces a partition on the population, then their product also induces a partition on the population. Often sources such as the census have matching (or at least matchable) demographic data; otherwise it must be estimated.

The problem facing poststratification by demographic feature is that the number of strata increases exponentially as a function of the number of features. For instance, 4 age brackets, 2 sexes, 5 income brackets, and 50 states of residence leads to $5 \cdot 2 \cdot 5 \cdot 50 = 2000$ strata. Adding another 5-way distinction, say for education level, leads to 10,000 strata. A simple model like the one in the previous section that takes an independent parameter $\theta_j$ for support in each stratum is unworkable in that many groups will have zero respondents and almost all groups will have very few respondents.

A practical approach to overcoming the problem of low data size per stratum is to use a regression model. Each demographic feature will require a regression coefficient for each of its subgroups, but now the parameters add to rather than multiply the total number of parameters. For example, with 4 age brackets, 2 sexes, 5 income brackets, and 50 states of residence, there are only $4 + 2 + 5 + 50 = 61$ regression coefficients to estimate. Now suppose that item $n$ has demographic features $\text{age}_n \in 1 : 5$, $\text{sex}_n \in 1 : 2$, $\text{income}_n \in 1 : 5$, and $\text{state}_n \in 1 : 50$. A logistic
regression may be formulated as

\[ y_n \sim \text{bernoulli}(\logit^{-1}(\alpha + \beta_{\text{age}[n]} + \gamma_{\text{sex}[n]} + \delta_{\text{income}[n]} + \epsilon_{\text{state}[n]})), \]

where \( \text{age}[n] \) is the age of the \( n \)-th respondent, \( \text{sex}[n] \) is their sex, \( \text{income}[n] \) their income and \( \text{state}[n] \) their state of residence. These coefficients can be assigned priors, resulting in a Bayesian regression model.

To poststratify the results, the population size for each combination of predictors must still be known. Then the population estimate is constructed as

\[ \sum_{i=1}^{5} \sum_{j=1}^{2} \sum_{k=1}^{5} \sum_{m=1}^{50} \logit^{-1}(\alpha + \beta_i + \gamma_j + \delta_k + \eta_m) \cdot \text{pop}_{i,j,k,m}, \]

where \( \text{pop}_{i,j,k,m} \) is the size of the subpopulation with age \( i \), sex \( j \), income level \( k \), and state of residence \( m \).

As formulated, it should be clear that any kind of prediction could be used as a basis for poststratification. For example, a Gaussian process or neural network could be used to produce a non-parametric model of outcomes \( y \) given predictors \( x \).

### 29.5. Multilevel regression and poststratification

With large numbers of demographic features, each cell may have very few items in it with which to estimate regression coefficients. For example, even in a national-level poll of 10,000 respondents, if they are divided by the 50 states, that’s only 200 respondents per state on average. When data sizes are small, parameter estimation can be stabilized and sharpened by providing hierarchical priors. With hierarchical priors, the data determines the amount of partial pooling among the groups. The only drawback is that if the number of groups is small, it can be hard to fit these models without strong hyperpriors.

The model introduced in the previous section had likelihood

\[ y_n \sim \text{bernoulli}(\logit^{-1}(\alpha + \beta_{\text{age}[n]} + \gamma_{\text{sex}[n]} + \delta_{\text{income}[n]} + \epsilon_{\text{state}[n]}))). \]

The overall intercept can be given a broad fixed prior,

\[ \alpha \sim \text{normal}(0, 5). \]

The other regression parameters can be given hierarchical priors,

\[ \beta_{1:4} \sim \text{normal}(0, \sigma^{\beta}) \]
\[ \gamma_{1:2} \sim \text{normal}(0, \sigma^{\gamma}) \]
\[ \delta_{1:5} \sim \text{normal}(0, \sigma^{\delta}) \]
\[ \epsilon_{1:50} \sim \text{normal}(0, \sigma^{\epsilon}) \]
The hyperparameters for scale of variation within a group can be given simple standard hyperpriors,
\[ \sigma^\beta, \sigma^\gamma, \sigma^\delta, \sigma^\epsilon \sim \text{normal}(0, 1). \]

The scales of these fixed hyperpriors need to be determined on a problem-by-problem basis, though ideally they will be close to standard (mean zero, unit variance).

Dealing with small partitions and non-identifiability

The multilevel structure of the models used for multilevel regression and poststratification consist of a sum of intercepts that vary by demographic feature. This immediately introduces non-identifiability. A constant added to each state coefficient and subtracted from each age coefficient leads to exactly the same likelihood.

This is non-identifiability that is only mitigated by the (hierarchical) priors. When demographic partitions are small, as they are with several categories in the example, it can be more computationally tractable to enforce a sum-to-zero constraint on the coefficients. Other values than zero will by necessity be absorbed into the intercept, which is why it typically gets a broader prior even with standardized data. With a sum to zero constraint, coefficients for binary features will be negations of each other. For example, because there are only two sex categories, \( \gamma_2 = -\gamma_1 \).

To implement sum-to-zero constraints,

```plaintext
parameters {
  vector[K - 1] alpha_raw;
  ...
transformed parameters {
  vector<multiplier = sigma_alpha>[K] alpha
  = append_row(alpha_raw, -sum(alpha_raw));
  ...
model {
  alpha ~ normal(0, sigma_alpha);
}
```

This prior is hard to interpret in that there are \( K \) normal distributions, but only \( K - 1 \) free parameters. An alternative is to put the prior only on \( \alpha_{raw} \), but that is also difficult to interpret.

Soft constraints can be more computationally tractable. They are also simpler to implement.

```plaintext
parameters {
  vector<multiplier = alpha>[K] alpha;
  ...
```
model {
  alpha ~ normal(0, sigma_alpha);
  sum(alpha) ~ normal(0, 0.001);
}

This leaves the regular prior, but adds a second prior that concentrates the sum near zero. The scale of the second prior will need to be established on a problem and data-set specific basis so that it doesn't shrink the estimates beyond the shrinkage of the hierarchical scale parameters.

Note that in the hierarchical model, the values of the coefficients when there are only two coefficients should be the same absolute value but opposite signs. Any other difference could be combined into the overall intercept \( \alpha \). Even with a wide prior on the intercept, the hyperprior on \( \sigma^2 \) may not be strong enough to enforce that, leading to a weak form non-identifiability in the posterior. Enforcing a (hard or soft) sum-to-zero constraint can help mitigate non-identifiability. Whatever prior is chosen, prior predictive checks can help diagnose problems with it.

None of this work to manage identifiability in multilevel regressions has anything to do with the poststratification; it's just required to fit a large multilevel regression with multiple discrete categories. Having multiple intercepts always leads to weak non-identifiability, even with the priors on the intercepts all centered at zero.

### 29.6. Coding MRP in Stan

Multilevel regression and poststratification can be coded directly in Stan. To code the non-centered parameterization for each coefficient, which will be required for sampling efficiency, the `multiplier` transform is used on each of the parameters. The combination of

```stan
vector<multiplier = s>[K] a;
...
a ~ normal(0, s);
```

implements a non-centered parameterization for \( a \); a centered parameterization would drop the `multiplier` specification. The prior scale \( s \) is being centered here. The prior location is fixed to zero in multilevel regressions because there is an overall intercept; introducing a location parameters in the prior would introduce non-identifiability with the overall intercept. The centered parameterization drops the `multiplier`.

Here is the full Stan model, which performs poststratification in the generated quantities using population sizes made available through data variable \( P \).

data {

```
int<lower = 0> N;
int<lower = 1, upper = 4> age[N];
int<lower = 1, upper = 5> income[N];
int<lower = 1, upper = 50> state[N];
int<lower = 0> y[N];
int<lower = 0> P[4, 5, 50];
}

parameters {
real alpha;
real<lower = 0> sigma_beta;
vector<multiplier = sigma_beta>[4] beta;
real<lower = 0> sigma_gamma;
vector<multiplier = sigma_gamma>[5] gamma;
real<lower = 0> sigma_delta;
vector<multiplier = sigma_delta>[50] delta;
}

model {
y ~ bernoulli_logit(alpha + beta[age] + gamma[income] + delta[state]);
alpha ~ normal(0, 2);
beta ~ normal(0, sigma_beta);
gamma ~ normal(0, sigma_gamma);
delta ~ normal(0, sigma_delta);
{ sigma_beta, sigma_gamma, sigma_delta } ~ normal(0, 1);
}

generated quantities {
real expect_pos = 0;
int total = 0;
for (b in 1:4)
  for (c in 1:5)
    for (d in 1:50) {
      total += P[b, c, d];
      expect_pos += P[b, c, d] * inv_logit(alpha + beta[b] + gamma[c] + delta[d]);
    }
real<lower = 0, upper = 1> phi = expect_pos / total;
}

Unlike in posterior predictive inference aimed at uncertainty, there is no need to introduce binomial sampling uncertainty into the estimate of expected positive votes.
Instead, generated quantities are computed as expectations. In general, it is more efficient to work in expectation if possible (the Rao-Blackwell theorem says it’s at least as efficient to work in expectation, but in practice, it can be much much more efficient, especially for discrete quantities).

**Binomial coding**
In some cases, it can be more efficient to break the data down by group. Suppose there are \(4 \times 5 \times 2 \times 50 = 2000\) groups. The data can be broken down into a size-2000 array, with entries corresponding to total vote counts in that group.

```plaintext
int<lower = 0> G;
int<lower = 1, upper = 4> age[G];
int<lower = 1, upper = 5> income[G];
int<lower = 1, upper = 50> state[G];
```

Then the number of positive votes and the number of total votes are collected into two parallel arrays indexed by group.

```plaintext
int<lower = 0> pos_votes[G];
int<lower = 0> total_votes[G];
```

Finally, the likelihood is converted to binomial.

```plaintext
pos_votes ~ binomial_logit(total_votes,
                           alpha + beta[age] + ...);
```

The predictors look the same because of the way the age and other data items are coded.

**Coding binary groups**
In this first model, sex is not included as a predictor. With only two categories, it needs to be modeled separately, because it is not feasible to build a hierarchical model with only two cases. A sex predictor is straightforward to add to the data block; it takes on values 1 or 2 for each of the \(N\) data points.

```plaintext
int<lower = 1, upper = 2> sex[N];
```

Then add a single regression coefficient as a parameter,

```plaintext
real epsilon;
```

In the log odds calculation, introduce a new term

```plaintext
[epsilon, -epsilon][sex]';
```

That is, the likelihood will now look like
y ~ bernoulli_logit(alpha + beta[age] + gamma[income] + delta[state] + [epsilon, -epsilon][sex]');

For data point n, the expression [epsilon, -epsilon][sex] takes on value [epsilon, -epsilon][sex][n], which with Stan's multi-indexing reduces to [epsilon, -epsilon][sex[n]]. This term evaluates to epsilon if sex[n] is 1 and to -epsilon if sex[n] is 2. The result is effectively a sum-to-zero constraint on two sex coefficients. The ' at the end transposes [epsilon, -epsilon][sex] which is a row_vector into a vector that can be added to the other variables.

Finally, a prior is needed for the coefficient in the model block,

epsilon ~ normal(0, 2);

As with other priors in multilevel models, the posterior for epsilon should be investigated to make sure it is not unrealistically wide.

29.7. Adding group-level predictors

If there are group-level predictors, such as average income in a state, or vote share in a previous election, these may be used as predictors in the regression. They will not pose an obstacle to poststratification because they are at the group level. For example, suppose the average income level in the state is available as the data variable

real<lower = 0> income[50];

then a regression coefficient psi can be added for the effect of average state income,

real psi;

with a fixed prior,

psi ~ normal(0, 2);

This prior assumes the income predictor has been standardized. Finally, a term is added to the regression for the fixed predictor,

y ~ bernoulli_logit(alpha + beta[age] + ... + delta[state] + income[state] * psi);

And finally, the formula in the generated quantities block is also updated,

expect_pos += P[b, c, d] * inv_logit(alpha + beta[b] + gamma[c] + delta[d] + income[d] * psi);
Here $d$ is the loop variable looping over states. This ensures that the poststratification formula matches the likelihood formula.
30. Decision Analysis

Statistical decision analysis is about making decisions under uncertainty. In order to make decisions, outcomes must have some notion of “utility” associated with them. The so-called “Bayes optimal” decision is the one that maximizes expected utility (or equivalently, minimizes expected loss). This chapter shows how Stan can be used to simultaneously estimate the distribution of outcomes based on decisions and compute the required expected utilities.

30.1. Outline of decision analysis

Following Andrew Gelman et al. (2013), Bayesian decision analysis can be factored into the following four steps.

1. Define a set $X$ of possible outcomes and a set $D$ of possible decisions.
2. Define a probability distribution of outcomes conditional on decisions through a conditional density function $p(x \mid d)$ for $x \in X$ and $d \in D$.
3. Define a utility function $U : X \to \mathbb{R}$ mapping outcomes to their utility.
4. Choose action $d^* \in D$ with highest expected utility, 
   
   $$d^* = \arg \max_d \mathbb{E}[U(x) \mid d].$$

The outcomes should represent as much information as possible that is relevant to utility. In Bayesian decision analysis, the distribution of outcomes will typically be a posterior predictive distribution conditioned on observed data. There is a large literature in psychology and economics related to defining utility functions. For example, the utility of money is usually assumed to be strictly concave rather than linear (i.e., the marginal utility of getting another unit of money decreases the more money one has).

30.2. Example decision analysis

This section outlines a very simple decision analysis for a commuter deciding among modes of transportation to get to work: walk, bike share, public transportation, or cab. Suppose the commuter has been taking various modes of transportation for the previous year and the transportation conditions and costs have not changed during that time. Over the year, such a commuter might accumulate two hundred observations of the time it takes to get to work given a choice of commute mode.
Step 1. Define decisions and outcomes
A decision consists of the choice of commute mode and the outcome is a time and cost. More formally,

- the set of decisions is \( D = 1 : 4 \), corresponding to the commute types walking, bicycling, public transportation, and cab, respectively, and
- the set of outcomes \( X = \mathbb{R} \times \mathbb{R}_+ \) contains pairs of numbers \( x = (c, t) \) consisting of a cost \( c \) and time \( t \geq 0 \).

Step 2. Define density of outcome conditioned on decision
The density required is \( p(x \mid d) \), where \( d \in D \) is a decision and \( x = (c, t) \in X \) is an outcome. Being a statistical decision problem, this density will the a posterior predictive distribution conditioned on previously observed outcome and decision pairs, based on a parameter model with parameters \( \theta \),

\[
p(x \mid d, x_{\text{obs}}, d_{\text{obs}}) = \int p(x \mid d, \theta) \cdot p(\theta \mid x_{\text{obs}}, d_{\text{obs}}) \, d\theta.
\]

The observed data for a year of commutes consists of choice of the chosen commute mode \( d_{\text{obs}} \) and observed costs and times \( x_{\text{obs}} = (c_{\text{obs}}, t_{\text{obs}}) \) for \( n \in 1 : 200 \).

For simplicity, commute time \( t_n \) for trip \( n \) will be modeled as lognormal for a given choice of transportation \( d_n \in 1 : 4 \),

\[
t_n \sim \lognormal(\mu_{d[n]}, \sigma_{d[n]}).
\]

To understand the notation, \( d_n \), also written \( d[n] \), is the mode of transportation used for trip \( n \). For example if trip \( n \) was by bicycle, then \( t_n \sim \lognormal(\mu_2, \sigma_2) \), where \( \mu_2 \) and \( \sigma_2 \) are the lognormal parameters for bicycling.

Simple fixed priors are used for each mode of transportation \( k \in 1 : 4 \),

\[
\begin{align*}
\mu_k &\sim \text{normal}(0, 5) \\
\sigma_k &\sim \text{lognormal}(0, 1).
\end{align*}
\]

These priors are consistent with a broad range of commute times; in a more realistic model each commute mode would have its own prior based on knowledge of the city and the time of day would be used as a covariate; here the commutes are taken to be exchangeable.

Cost is usually a constant function for public transportation, walking, and bicycling. Nevertheless, for simplicity, all costs will be modeled as lognormal,

\[
c_n \sim \lognormal(\nu_{d[n]}, \tau_{d[n]}).
\]
Again, the priors are fixed for the modes of transportation,

$$\nu_k \sim \text{normal}(0, 5)$$
$$\tau_k \sim \text{lognormal}(0, 1).$$

A more realistic approach would model cost conditional on time, because the cost of a cab depends on route chosen and the time it takes.

The full set of parameters that are marginalized in the posterior predictive distribution is

$$\theta = (\mu_{1:4}, \sigma_{1:4}, \nu_{1:4}, \tau_{1:4}).$$

**Step 3. Define the utility function**

For the sake of concreteness, the utility function will be assumed to be a simple function of cost and time. Further suppose the commuter values their commute time at $25 per hour and has a utility function that is linear in the commute cost and time. Then the utility function may be defined as

$$U(c, t) = -(c + 25 \cdot t).$$

The sign is negative because high cost is undesirable. A better utility function might have a step function or increasing costs for being late, different costs for different modes of transportation because of their comfort and environmental impact, and non-linearity of utility in cost.

**Step 4. Maximize expected utility**

At this point, all that is left is to calculate expected utility for each decision and choose the optimum. If the decisions consist of a small set of discrete choices, expected utility can be easily coded in Stan. The utility function is coded as a function, the observed data is coded as data, the model parameters coded as parameters, and the model block itself coded to follow the sampling distributions of each parameter.

```stan
functions {
  real U(real c, real t) {
    return -(c + 25 * t);
  }
}
data {
  int<lower = 0> N;
  int<lower = 1, upper = 4> d[N];
  real c[N];
  real<lower = 0> t[N];
}
parameters {
    vector[4] mu;
    vector<lower = 0>[4] sigma;
    real nu[4];
    real<lower = 0> tau[4];
}

model {
    mu ~ normal(0, 1);
    sigma ~ lognormal(0, 0.25);
    nu ~ normal(0, 20);
    tau ~ lognormal(0, 0.25);
    t ~ lognormal(mu[d], sigma[d]);
    c ~ lognormal(nu[d], tau[d]);
}

generated quantities {
    real util[4];
    for (k in 1:4)
        util[k] = U(lognormal_rng(mu[k], sigma[k]),
                    lognormal_rng(nu[k], tau[k]));
}

The generated quantities block defines an array variable util where util[k], which will hold the utility derived from a random commute for choice k generated according to the model parameters for that choice. This randomness is required to appropriately characterize the posterior predictive distribution of utility.

For simplicity in this initial formulation, all four commute options have their costs estimated, even though cost is fixed for three of the options. To deal with the fact that some costs are fixed, the costs would have to be hardcoded or read in as data, nu and tau would be declared as univariate, and the RNG for cost would only be employed when k == 4.

Defining the utility function for pairs of vectors would allow the random number generation in the generated quantities block to be vectorized.

All that is left is to run Stan. The posterior mean for util[k] is the expected utility,
which written out with full conditioning, is

$$
\mathbb{E}\left[ U(x) \mid d = k, d^{\text{obs}}, x^{\text{obs}} \right] = \int U(x) \cdot p(x \mid d = k, \theta) \cdot p(\theta \mid d^{\text{obs}}, x^{\text{obs}}) \, d\theta
$$

$$
\approx \frac{1}{M} \sum_{m=1}^{M} U(x^{(m)}),
$$

where

$$
x^{(m)} \sim p(x \mid d = k, \theta^{(m)})
$$

and

$$
\theta^{(m)} \sim p(\theta \mid d^{\text{obs}}, x^{\text{obs}}).
$$

In terms of Stan’s execution, the random generation of $x^{(m)}$ is carried out with the \texttt{lognormal_rng} operations after $\theta^{(m)}$ is drawn from the model posterior. The average is then calculated after multiple chains are run and combined.

It only remains to make the decision $k$ with highest expected utility, which will correspond to the choice with the highest posterior mean for $\text{util}[k]$. This can be read off of the \texttt{mean} column of the Stan’s summary statistics or accessed programmatically through Stan’s interfaces.

### 30.3. Continuous choices

Many choices, such as how much to invest for retirement or how long to spend at the gym are not discrete, but continuous. In these cases, the continuous choice can be coded as data in the Stan program. Then the expected utilities may be calculated. In other words, Stan can be used as a function from a choice to expected utilities. Then an external optimizer can call that function. This optimization can be difficult without gradient information. Gradients could be supplied by automatic differentiation, but Stan is not currently instrumented to calculate those derivatives.
31. **The Bootstrap and Bagging**

The bootstrap is a technique for approximately sampling from the error distribution for an estimator. Thus it can be used as a Monte Carlo method to estimate standard errors and confidence intervals for point estimates (Efron and Tibshirani 1986; 1994). It works by subsampling the original data and computing sample estimates from the subsample. Like other Monte Carlo methods, the bootstrap is plug-and-play, allowing great flexibility in both model choice and estimator.

Bagging is a technique for combining bootstrapped estimators for model criticism and more robust inference (Breiman 1996; Huggins and Miller 2019).

### 31.1. The bootstrap

**Estimators**

An estimator is nothing more than a function mapping a data set to one or more numbers, which are called “estimates.” For example, the mean function maps a data set \( y_1, \ldots, N \) to a number by

\[
\text{mean}(y) = \frac{1}{N} \sum_{n=1}^{N} y_n, 
\]

and hence meets the definition of an estimator. Given the likelihood function

\[
p(y \mid \mu) = \prod_{n=1}^{N} \text{normal}(y_n \mid \mu, 1),
\]

the mean is the maximum likelihood estimator,

\[
\text{mean}(y) = \arg \max_{\mu} p(y \mid \mu, 1)
\]

A Bayesian approach to point estimation would be to add a prior and use the posterior mean or median as an estimator. Alternatively, a penalty function could be added to the likelihood so that optimization produces a penalized maximum likelihood estimate. With any of these approaches, the estimator is just a function from data to a number.

In analyzing estimators, the data set is being modeled as a random variable. It is assumed that the observed data is just one of many possible random samples of data that may have been produced. If the data is modeled a random variable, then the
estimator applied to the data is also a random variable. The simulations being done for the bootstrap are attempts to randomly sample replicated data sets and compute the random properties of the estimators using standard Monte Carlo methods.

**The bootstrap in pseudocode**

The bootstrap works by applying an estimator to replicated data sets. These replicates are created by subsampling the original data with replacement. The sample quantiles may then be used to estimate standard errors and confidence intervals.

The following pseudocode estimates 95% confidence intervals and standard errors for a generic estimate \( \hat{\theta} \) that is a function of data \( y \).

```plaintext
for (m in 1:M) {
    y_rep[m] <- sample_uniform(y)
    theta_hat[m] <- estimate_theta(y_rep[m])
}
std_error = sd(theta_hat)
conf_95pct = [ quantile(theta_hat, 0.025),
               quantile(theta_hat, 0.975) ]
```

The `sample_uniform` function works by independently assigning each element of \( y_{rep} \) an element of \( y \) drawn uniformly at random. This produces a sample *with replacement*. That is, some elements of \( y \) may show up more than once in \( y_{rep} \) and some may not appear at all.

### 31.2. Coding the bootstrap in Stan

The bootstrap procedure can be coded quite generally in Stan models. The following code illustrates a Stan model coding the likelihood for a simple linear regression. There is a parallel vector \( x \) of predictors in addition to outcomes \( y \). To allow a single program to fit both the original data and random subsamples, the variable `resample` is set to 1 to resample and 0 to use the original data.

```stan
data {
    int<lower = 0> N;
    vector[N] x;
    vector[N] y;
    int<lower = 0, upper = 1> resample;
}
transformed data {
    simplex[N] uniform = rep_vector(1.0 / N, N);
    int<lower = 1, upper = N> boot_idxs[N];
    for (n in 1:N)
```
The model accepts data in the usual form for a linear regression as a number of observations $N$ with a size $N$ vector $x$ of predictors and a size $N$ vector of outcomes. The transformed data block generates a set of indexes into the data that is the same size as the data. This is done by independently sampling each entry of `boot_idxs` from $1:N$, using a discrete uniform distribution coded as a categorical random number generator with an equal chance for each outcome. If resampling is not done, the array `boot_idxs` is defined to be the sequence $1:N$, because $x == x[1:N]$ and $y = y[1:N]$.

For example, when `resample == 1`, if $N = 4$, the value of `boot_idxs` might be $\{2, 1, 1, 3\}$, resulting in a bootstrap sample $\{y[2], y[1], y[1], y[3]\}$ with the first element repeated twice and the fourth element not sampled at all.

The parameters are the usual regression coefficients for the intercept $alpha$, slope $beta$, and error scale $sigma$. The model uses the bootstrap index variable `boot_idx` to index the predictors as $x[boot_idx]$ and outcomes as $y[boot_idx]$. This generates a new size-$N$ vector whose entries are defined by $x[boot_idx][n] = x[boot_idx[n]]$ and similarly for $y$. For example, if $N = 4$ and `boot_idxs = {2, 1, 1, 3}`, then $x[boot_idxs] = [x[2], x[1], x[1], x[3]]'$ and $y[boot_idxs] = [y[2], y[1], y[1], y[3]]'$. The predictor and outcome vectors remain aligned, with both elements of the pair $x[1]$ and $y[1]$ repeated twice.

With the model defined this way, if `resample` is 1, the model is fit to a bootstrap subsample of the data. If `resample` is 0, the model is fit to the original data as given. By running the bootstrap fit multiple times, confidence intervals can be generated from quantiles of the results.

### 31.3. Error statistics from the bootstrap
Running the model multiple times produces a Monte Carlo sample of estimates from multiple alternative data sets subsampled from the original data set. The error
distribution is just the distribution of the bootstrap estimates minus the estimate for the original data set.

To estimate standard errors and confidence intervals for maximum likelihood estimates the Stan program is executed multiple times using optimization (which turns off Jacobian adjustments for constraints and finds maximum likelihood estimates). On the order of one hundred replicates is typically enough to get a good sense of standard error; more will be needed to accurate estimate the boundaries of a 95% confidence interval. On the other hand, given that there is inherent variance due to sampling the original data \( y \), it is usually not worth calculating bootstrap estimates to high precision.

**Standard errors**

Here’s the result of calculating standard errors for the linear regression model above with \( N = 50 \) data points, \( \alpha = 1.2, \beta = -0.5, \) and \( \sigma = 1.5 \). With a total of \( M = 100 \) bootstrap samples, there are 100 estimates of \( \alpha \), 100 of \( \beta \), and 100 of \( \sigma \). These are then treated like Monte Carlo draws. For example, the sample standard deviation of the draws for \( \alpha \) provide the bootstrap estimate of the standard error in the estimate for \( \alpha \). Here’s what it looks like for the above model with \( M = 100 \)

<table>
<thead>
<tr>
<th>parameter</th>
<th>estimate</th>
<th>std err</th>
</tr>
</thead>
<tbody>
<tr>
<td>alpha</td>
<td>1.359</td>
<td>0.218</td>
</tr>
<tr>
<td>beta</td>
<td>-0.610</td>
<td>0.204</td>
</tr>
<tr>
<td>sigma</td>
<td>1.537</td>
<td>0.142</td>
</tr>
</tbody>
</table>

With the data set fixed, these estimates of standard error will display some Monte Carlo error. For example, here are the standard error estimates from five more runs holding the data the same, but allowing the subsampling to vary within Stan:

<table>
<thead>
<tr>
<th>parameter</th>
<th>estimate</th>
<th>std err</th>
</tr>
</thead>
<tbody>
<tr>
<td>alpha</td>
<td>1.359</td>
<td>0.206</td>
</tr>
<tr>
<td>alpha</td>
<td>1.359</td>
<td>0.240</td>
</tr>
<tr>
<td>alpha</td>
<td>1.359</td>
<td>0.234</td>
</tr>
<tr>
<td>alpha</td>
<td>1.359</td>
<td>0.249</td>
</tr>
<tr>
<td>alpha</td>
<td>1.359</td>
<td>0.227</td>
</tr>
</tbody>
</table>

Increasing \( M \) will reduce Monte Carlo error, but this is not usually worth the extra computation time as there is so much other uncertainty due to the original data sample \( y \).
Confidence intervals
As usual with Monte Carlo methods, confidence intervals are estimated using quantiles of the draws. That is, if there are $M = 1000$ estimates of $\hat{\alpha}$ in different subsamples, the 2.5% quantile and 97.5% quantile pick out the boundaries of the 95% confidence interval around the estimate for the actual data set $y$. To get accurate 97.5% quantile estimates requires a much larger number of Monte Carlo simulations (roughly twenty times as large as needed for the median).

31.4. Bagging
When bootstrapping is carried through inference it is known as bootstrap aggregation, or bagging, in the machine-learning literature (Breiman 1996). In the simplest case, this involves bootstrapping the original data, fitting a model to each bootstrapped data set, then averaging the predictions. For instance, rather than using an estimate $\hat{\sigma}$ from the original data set, bootstrapped data sets $y^{\text{boot}(1)}, \ldots, y^{\text{boot}(N)}$ are generated. Each is used to generate an estimate $\hat{\sigma}^{\text{boot}(n)}$. The final estimate is

$$\hat{\sigma} = \frac{1}{N} \sum_{n=1}^{N} \hat{\sigma}^{\text{boot}(n)}.$$ 

The same would be done to estimate a predictive quantity $\tilde{y}$ for as yet unseen data.

$$\hat{\tilde{y}} = \frac{1}{N} \sum_{n=1}^{N} \hat{\tilde{y}}^{\text{boot}(n)}.$$ 

For discrete parameters, voting is used to select the outcome.

One way of viewing bagging is as a classical attempt to get something like averaging over parameter estimation uncertainty.

31.5. Bayesian bootstrap and bagging
A Bayesian estimator may be analyzed with the bootstrap in exactly the same way as a (penalized) maximum likelihood estimate. For example, the posterior mean and posterior median are two different Bayesian estimators. The bootstrap may be used estimate standard errors and confidence intervals, just as for any other estimator.

(Huggins and Miller 2019) use the bootstrap to assess model calibration and fitting in a Bayesian framework and further suggest using bagged estimators as a guard against model misspecification. Bagged posteriors will typically have wider posterior intervals than those fit with just the original data, showing that the method is not a pure Bayesian approach to updating, and indicating it would not be calibrated if the model were well specified. The hope is that it can guard against over-certainty in a poorly specified model.
Appendices

These are the appendices for the book, including a program style guide and a guide to translating BUGS or JAGS programs.
32. Stan Program Style Guide

This chapter describes the preferred style for laying out Stan models. These are not rules of the language, but simply recommendations for laying out programs in a text editor. Although these recommendations may seem arbitrary, they are similar to those of many teams for many programming languages. Like rules for typesetting text, the goal is to achieve readability without wasting white space either vertically or horizontally.

32.1. Choose a consistent style
The most important point of style is consistency. Consistent coding style makes it easier to read not only a single program, but multiple programs. So when departing from this style guide, the number one recommendation is to do so consistently.

32.2. Line length
Line lengths should not exceed 80 characters.\(^1\)

This is a typical recommendation for many programming language style guides because it makes it easier to lay out text edit windows side by side and to view the code on the web without wrapping, easier to view diffs from version control, etc. About the only thing that is sacrificed is laying out expressions on a single line.

32.3. File extensions
The recommended file extension for Stan model files is .stan. For Stan data dump files, the recommended extension is .R, or more informatively, .data.R.

32.4. Variable naming
The recommended variable naming is to follow C/C++ naming conventions, in which variables are lowercase, with the underscore character (\_) used as a separator. Thus it is preferred to use sigma_y, rather than the run together sigmay, camel-case sigmaY, or capitalized camel-case SigmaY. Even matrix variables should be lowercased.

The exception to the lowercasing recommendation, which also follows the C/C++ conventions, is for size constants, for which the recommended form is a single uppercase letter. The reason for this is that it allows the loop variables to match. So

\(^1\)Even 80 characters may be too many for rendering in print; for instance, in this manual, the number of code characters that fit on a line is about 65.
loops over the indices of an $M \times N$ matrix $a$ would look as follows.

```stan
for (m in 1:M)
    for (n in 1:N)
        a[m,n] = ...
```

### 32.5. Local variable scope
Declaring local variables in the block in which they are used aids in understanding programs because it cuts down on the amount of text scanning or memory required to reunite the declaration and definition.

The following Stan program corresponds to a direct translation of a BUGS model, which uses a different element of $\mu$ in each iteration.

```stan
model {
    real mu[N];
    for (n in 1:N) {
        mu[n] = alpha * x[n] + beta;
        y[n] ~ normal(mu[n],sigma);
    }
}
```

Because variables can be reused in Stan and because they should be declared locally for clarity, this model should be recoded as follows.

```stan
model {
    for (n in 1:N) {
        real mu;
        mu = alpha * x[n] + beta;
        y[n] ~ normal(mu,sigma);
    }
}
```

The local variable can be eliminated altogether, as follows.

```stan
model {
    for (n in 1:N)
        y[n] ~ normal(alpha * x[n] + beta, sigma);
}
```

There is unlikely to be any measurable efficiency difference between the last two implementations, but both should be a bit more efficient than the BUGS translation.
Scope of compound structures with componentwise assignment

In the case of local variables for compound structures, such as arrays, vectors, or matrices, if they are built up component by component rather than in large chunks, it can be more efficient to declare a local variable for the structure outside of the block in which it is used. This allows it to be allocated once and then reused.

```stan
model {
  vector[K] mu;
  for (n in 1:N) {
    for (k in 1:K)
      mu[k] = ...;
    y[n] ~ multi_normal(mu,Sigma);
  }
}
```

In this case, the vector `mu` will be allocated outside of both loops, and used a total of `N` times.

### 32.6. Parentheses and brackets

#### Optional parentheses for single-statement blocks

Single-statement blocks can be rendered in one of two ways. The fully explicit bracketed way is as follows.

```stan
for (n in 1:N) {
  y[n] ~ normal(mu,1);
}
```

The following statement without brackets has the same effect.

```stan
for (n in 1:N)
  y[n] ~ normal(mu,1);
```

Single-statement blocks can also be written on a single line, as in the following example.

```stan
for (n in 1:N) y[n] ~ normal(mu,1);
```

These can be much harder to read than the first example. Only use this style if the statement is simple, as in this example. Unless there are many similar cases, it's almost always clearer to put each sampling statement on its own line.

Conditional and looping statements may also be written without brackets.

The use of for loops without brackets can be dangerous. For instance, consider this program.
for (n in 1:N)
    z[n] ~ normal(nu,1);
    y[n] ~ normal(mu,1);

Because Stan ignores whitespace and the parser completes a statement as eagerly as possible (just as in C++), the previous program is equivalent to the following program.

for (n in 1:N) {
    z[n] ~ normal(nu,1);
}

y[n] ~ normal(mu,1);

Parentheses in nested operator expressions
The preferred style for operators minimizes parentheses. This reduces clutter in code that can actually make it harder to read expressions. For example, the expression \(a + b * c\) is preferred to the equivalent \(a + (b * c)\) or \((a + (b * c))\). The operator precedences and associativities follow those of pretty much every programming language including Fortran, C++, R, and Python; full details are provided in the reference manual.

Similarly, comparison operators can usually be written with minimal bracketing, with the form \(y[n] > 0 \text{ || } x[n] != 0\) preferred to the bracketed form \((y[n] > 0) \text{ || } (x[n] != 0)\).

No open brackets on own line
Vertical space is valuable as it controls how much of a program you can see. The preferred Stan style is as shown in the previous section, not as follows.

for (n in 1:N)
{
    y[n] ~ normal(mu,1);
}

This also goes for parameters blocks, transformed data blocks, which should look as follows.

transformed parameters {
    real sigma;
    ...
}
32.7. Conditionals
Stan supports the full C++-style conditional syntax, allowing real or integer values to act as conditions, as follows.

```stan
real x;
...
if (x) {
   // executes if x not equal to 0
   ...
}
```

Explicit comparisons of non-Boolean conditions
The preferred form is to write the condition out explicitly for integer or real values that are not produced as the result of a comparison or boolean operation, as follows.

```stan
if (x != 0) ...
```

32.8. Functions
Functions are laid out the same way as in languages such as Java and C++. For example,

```stan
real foo(real x, real y) {
   return sqrt(x * log(y));
}
```

The return type is flush left, the parentheses for the arguments are adjacent to the arguments and function name, and there is a space after the comma for arguments after the first. The open curly brace for the body is on the same line as the function name, following the layout of loops and conditionals. The body itself is indented; here we use two spaces. The close curly brace appears on its own line.

If function names or argument lists are long, they can be written as

```stan
matrix
function_to_do_some_hairy_algebra(matrix thingamabob,
   vector doohickey2) {
   ...body...
}
```

The function starts a new line, under the type. The arguments are aligned under each other.

Function documentation should follow the Javadoc and Doxygen styles. Here’s an example repeated from the documenting functions section.
/**
 * Return a data matrix of specified size with rows
 * corresponding to items and the first column filled
 * with the value 1 to represent the intercept and the
 * remaining columns randomly filled with unit-normal draws.
 * *
 * @param N Number of rows correspond to data items
 * @param K Number of predictors, counting the intercept, per
 * item.
 * @return Simulated predictor matrix.
 */
matrix predictors_rng(int N, int K) {
  ...
}

The open comment is /**, asterisks are aligned below the first asterisk of the open
comment, and the end comment */ is also aligned on the asterisk. The tags @param
and @return are used to label function arguments (i.e., parameters) and return values.

32.9. White space
Stan allows spaces between elements of a program. The white space characters
allowed in Stan programs include the space (ASCII 0x20), line feed (ASCII 0x0A),
carriage return (0x0D), and tab (0x09). Stan treats all whitespace characters inter-
changeably, with any sequence of whitespace characters being syntactically equiva-
 lent to a single space character. Nevertheless, effective use of whitespace is the key
to good program layout.

Line breaks between statements and declarations
It is dispreferred to have multiple statements or declarations on the same line, as in
the following example.

```stan
transformed parameters {
  real mu_centered; real sigma;
  mu = (mu_raw - mean_mu_raw); sigma = pow(tau,-2);
}
```

These should be broken into four separate lines.

No tabs
Stan programs should not contain tab characters. They are legal and may be
used anywhere other whitespace occurs. Using tabs to layout a program is highly
unportable because the number of spaces represented by a single tab character varies
depending on which program is doing the rendering and how it is configured.

**Two-character indents**
Stan has standardized on two space characters of indentation, which is the standard convention for C/C++ code. Another sensible choice is four spaces, which is the convention for Java and Python. Just be consistent.

**Space between if, { and condition**
Use a space after ifs. For instance, use `if (x < y) ...`, not `if(x < y) ...`.

**No space For function calls**
There is no space between a function name and the function it applies to. For instance, use `normal(0,1)`, not `normal (0,1)`.

**Spaces around operators**
There should be spaces around binary operators. For instance, use `y[1] = x`, not `y[1]=x`, use `(x + y) * z` not `(x+y)*z`.

**Breaking expressions across lines**
Sometimes expressions are too long to fit on a single line. In that case, the recommended form is to break before an operator, aligning the operator to indicate scoping. For example, use the following form (though not the content; inverting matrices is almost always a bad idea).

```plaintext
target += (y - mu)' * inv(Sigma) * (y - mu);
```

Here, the multiplication operator (*) is aligned to clearly signal the multiplicands in the product.

For function arguments, break after a comma and line the next argument up underneath as follows.

```plaintext
y[n] ~ normal(alpha + beta * x + gamma * y, pow(tau,-0.5));
```

**Optional spaces after commas**
Optionally use spaces after commas in function arguments for clarity. For example, `normal(alpha * x[n] + beta, sigma)` can also be written as `normal(alpha * x[n] + beta, sigma).

---

2This is the usual convention in both typesetting and other programming languages. Neither R nor BUGS allows breaks before an operator because they allow newlines to signal the end of an expression or statement.
Unix newlines
Wherever possible, Stan programs should use a single line feed character to separate lines. All of the Stan developers (so far, at least) work on Unix-like operating systems and using a standard newline makes the programs easier for us to read and share.

Platform specificity of newlines
Newlines are signaled in Unix-like operating systems such as Linux and Mac OS X with a single line-feed (LF) character (ASCII code point 0x0A). Newlines are signaled in Windows using two characters, a carriage return (CR) character (ASCII code point 0x0D) followed by a line-feed (LF) character.
33. Transitioning from BUGS

From the outside, Stan and BUGS\(^1\) are similar—they use statistically-themed modeling languages (which are similar but with some differences; see below), they can be called from R, running some specified number of chains to some specified length, producing posterior simulations that can be assessed using standard convergence diagnostics. This is not a coincidence: in designing Stan: we wanted to keep many of the useful features of Bugs.

33.1. Some differences in how BUGS and Stan work

**BUGS is interpreted, Stan is compiled**

Stan is compiled in two steps, first a model is translated to templated C++ and then to a platform-specific executable. Stan, unlike BUGS, allows the user to directly program in C++, but we do not describe how to do this in this Stan manual (see the getting started with C++ section of https://mc-stan.org for more information on using Stan directly from C++).

**BUGS performs MCMC updating one scalar parameter at a time, Stan uses HMC which moves in the entire space of all the parameters at each step**

BUGS performs MCMC updating one scalar parameter at a time, (with some exceptions such as JAGS’s implementation of regression and generalized linear models and some conjugate multivariate parameters), using conditional distributions (Gibbs sampling) where possible and otherwise using adaptive rejection sampling, slice sampling, and Metropolis jumping. BUGS figures out the dependence structure of the joint distribution as specified in its modeling language and uses this information to compute only what it needs at each step. Stan moves in the entire space of all the parameters using Hamiltonian Monte Carlo (more precisely, the no-U-turn sampler), thus avoiding some difficulties that occur with one-dimension-at-a-time sampling in high dimensions but at the cost of requiring the computation of the entire log density at each step.

**Differences in tuning during warmup**

BUGS tunes its adaptive jumping (if necessary) during its warmup phase (traditionally referred to as “burn-in”). Stan uses its warmup phase to tune the no-U-turn sampler (NUTS).

\(^1\)Except where otherwise noted, we use “BUGS” to refer to WinBUGS, OpenBUGS, and JAGS, indiscriminately.
The Stan language is directly executable, the BUGS modeling language is not

The BUGS modeling language is not directly executable. Rather, BUGS parses its model to determine the posterior density and then decides on a sampling scheme. In contrast, the statements in a Stan model are directly executable: they translate exactly into C++ code that is used to compute the log posterior density (which in turn is used to compute the gradient).

Differences in statement order

In BUGS, statements are executed according to the directed graphical model so that variables are always defined when needed. A side effect of the direct execution of Stan’s modeling language is that statements execute in the order in which they are written. For instance, the following Stan program, which sets \( \mu \) before using it to sample \( y \):

\[
\begin{align*}
\text{mu} &= a + b \times x; \\
y &\sim \text{normal} (\mu, \text{sigma});
\end{align*}
\]

translates to the following C++ code:

\[
\begin{align*}
\text{mu} &= a + b \times x; \\
\text{target} &+= \text{normal}_\text{lpdf} (y \mid \mu, \text{sigma});
\end{align*}
\]

Contrast this with the following Stan program:

\[
\begin{align*}
y &\sim \text{normal} (\mu, \text{sigma}); \\
\text{mu} &= a + b \times x;
\end{align*}
\]

This program is well formed, but is almost certainly a coding error, because it attempts to use \( \mu \) before it is set. The direct translation to C++ code highlights the potential error of using \( \mu \) in the first statement:

\[
\begin{align*}
\text{target} &+= \text{normal}_\text{lpdf} (y \mid \mu, \text{sigma}); \\
\text{mu} &= a + b \times x;
\end{align*}
\]

To trap these kinds of errors, variables are initialized to the special not-a-number (NaN) value. If NaN is passed to a log probability function, it will raise a domain exception, which will in turn be reported by the sampler. The sampler will reject the sample out of hand as if it had zero probability.

Stan computes the gradient of the log density, BUGS computes the log density but not its gradient

Stan uses its own C++ algorithmic differentiation packages to compute the gradient of the log density (up to a proportion). Gradients are required during the Hamiltonian dynamics simulations within the leapfrog algorithm of the Hamiltonian Monte
Carlo and NUTS samplers.

Both **BUGS** and **Stan** are **semi-automatic**

Both BUGS and Stan are semi-automatic in that they run by themselves with no outside tuning required. Nevertheless, the user needs to pick the number of chains and number of iterations per chain. We usually pick 4 chains and start with 10 iterations per chain (to make sure there are no major bugs and to approximately check the timing), then go to 100, 1000, or more iterations as necessary. Compared to Gibbs or Metropolis, Hamiltonian Monte Carlo can take longer per iteration (as it typically takes many “leapfrog steps” within each iteration), but the iterations typically have lower autocorrelation. So Stan might work fine with 1000 iterations in an example where BUGS would require 100,000 for good mixing. We recommend monitoring potential scale reduction statistics ($\hat{R}$) and the effective sample size to judge when to stop (stopping when $\hat{R}$ values do not counter-indicate convergence and when enough effective samples have been collected).

**Licensing**

WinBUGS is closed source. OpenBUGS and JAGS are both licensed under the Gnu Public License (GPL), otherwise known as copyleft due to the restrictions it places on derivative works. Stan is licensed under the much more liberal new BSD license.

**Interfaces**

Like WinBUGS, OpenBUGS and JAGS, Stan can be run directly from the command line or through common analytics platforms like R, Python, Julia, MATLAB, Mathematica, and the command line.

**Platforms**

Like OpenBUGS and JAGS, Stan can be run on Linux, Mac, and Windows platforms.

### 33.2. Some differences in the modeling languages

The BUGS modeling language follows an R-like syntax in which line breaks are meaningful. Stan follows the rules of C, in which line breaks are equivalent to spaces, and each statement ends in a semicolon. For example:

```r
y ~ normal(mu, sigma);
```

and

```r
for (i in 1:n) y[i] ~ normal(mu, sigma);
```

Or, equivalently (recall that a line break is just another form of whitespace),

```r
for (i in 1:n)
  y[i] ~ normal(mu, sigma);
```
and also equivalently,

```plaintext
for (i in 1:n) {
    y[i] ~ normal(mu, sigma);
}
```

There's a semicolon after the model statement but not after the brackets indicating the body of the for loop.

In Stan, variables can have names constructed using letters, numbers, and the underscore (_) symbol, but nothing else (and a variable name cannot begin with a number). BUGS variables can also include the dot, or period (.) symbol.

In Stan, the second argument to the “normal” function is the standard deviation (i.e., the scale), not the variance (as in *Bayesian Data Analysis*) and not the inverse-variance (i.e., precision) (as in BUGS). Thus a normal with mean 1 and standard deviation 2 is `normal(1,2)`, not `normal(1,4)` or `normal(1,0.25)`.

Similarly, the second argument to the “multivariate normal” function is the covariance matrix and not the inverse covariance matrix (i.e., the precision matrix) (as in BUGS). The same is true for the “multivariate student” distribution.

The distributions have slightly different names:

<table>
<thead>
<tr>
<th>BUGS</th>
<th>Stan</th>
</tr>
</thead>
<tbody>
<tr>
<td>dnorm</td>
<td>normal</td>
</tr>
<tr>
<td>dbinom</td>
<td>binomial</td>
</tr>
<tr>
<td>dpois</td>
<td>poisson</td>
</tr>
<tr>
<td>...</td>
<td>...</td>
</tr>
</tbody>
</table>

Stan, unlike BUGS, allows intermediate quantities, in the form of local variables, to be reassigned. For example, the following is legal and meaningful (if possibly inefficient) Stan code.

```plaintext
{
    total = 0;
    for (i in 1:n){
        theta[i] ~ normal(total, sigma);
        total = total + theta[i];
    }
}
```
In BUGS, the above model would not be legal because the variable total is defined more than once. But in Stan, the loop is executed in order, so total is overwritten in each step.

Stan uses explicit declarations. Variables are declared with base type integer or real, and vectors, matrices, and arrays have specified dimensions. When variables are bounded, we give that information also. For data and transformed parameters, the bounds are used for error checking. For parameters, the constraints are critical to sampling as they determine the geometry over which the Hamiltonian is simulated.

In Stan, variables can be declared as data, transformed data, parameters, transformed parameters, or generated quantities. They can also be declared as local variables within blocks. For more information, see the part of this manual devoted to the Stan programming language and examine at the example models.

Stan allows all sorts of tricks with vector and matrix operations which can make Stan models more compact. For example, arguments to probability functions may be vectorized, allowing

```stan
for (i in 1:n)
  y[i] ~ normal(mu[i], sigma[i]);
```

to be expressed more compactly as

```stan
y ~ normal(mu, sigma);
```

The vectorized form is also more efficient because Stan can unfold the computation of the chain rule during algorithmic differentiation.

Stan also allows for arrays of vectors and matrices. For example, in a hierarchical model might have a vector of K parameters for each of J groups; this can be declared using

```stan
vector[K] theta[J];
```

Then theta[j] is an expression denoting a K-vector and may be used in the code just like any other vector variable.

An alternative encoding would be with a two-dimensional array, as in

```stan
real theta[J,K];
```

The vector version can have some advantages, both in convenience and in computational speed for some operations.

---

2Most distributions have been vectorized, but currently the truncated versions may not exist and may not be vectorized.
A third encoding would use a matrix:

```c++
matrix[J,K] theta;
```

but in this case, `theta[j]` is a row vector, not a vector, and accessing it as a vector is less efficient than with an array of vectors. The transposition operator, as in `theta[j]'`, may be used to convert the row vector `theta[j]` to a (column) vector. Column vector and row vector types are not interchangeable everywhere in Stan; see the function signature declarations in the programming language section of this manual.

Stan supports general conditional statements using a standard if-else syntax. For example, a zero-inflated (or -deflated) Poisson mixture model is defined using the if-else syntax as described in the zero inflation section.

Stan supports general while loops using a standard syntax. While loops give Stan full Turing equivalent computational power. They are useful for defining iterative functions with complex termination conditions. As an illustration of their syntax, the for-loop

```c++
model {
  ....
  for (n in 1:N) {
    ... do something with n ....
  }
}
```

may be recoded using the following while loop.

```c++
model {
  int n;
  ...
  n = 1;
  while (n <= N) {
    ... do something with n ...
    n = n + 1;
  }
}
```

### 33.3. Some differences in the statistical models that are allowed
Stan does not yet support declaration of discrete parameters. Discrete data variables are supported. Inference is supported for discrete parameters as described in the mixture and latent discrete parameters chapters of the manual.
Stan has some distributions on covariance matrices that do not exist in BUGS, including a uniform distribution over correlation matrices which may be rescaled, and the priors based on C-vines defined in Lewandowski, Kurowicka, and Joe (2009). In particular, the Lewandowski et al. prior allows the correlation matrix to be shrunk toward the unit matrix while the scales are given independent priors.

In BUGS you need to define all variables. In Stan, if you declare but don’t define a parameter it implicitly has a flat prior (on the scale in which the parameter is defined). For example, if you have a parameter $p$ declared as

```stan
real<lower = 0, upper = 1> p;
```

and then have no sampling statement for $p$ in the `model` block, then you are implicitly assigning a uniform $[0, 1]$ prior on $p$.

On the other hand, if you have a parameter $\theta$ declared with

```stan
real theta;
```

and have no sampling statement for $\theta$ in the `model` block, then you are implicitly assigning an improper uniform prior on $(-\infty, \infty)$ to $\theta$.

BUGS models are always proper (being constructed as a product of proper marginal and conditional densities). Stan models can be improper. Here is the simplest improper Stan model:

```stan
parameters {
  real theta;
}
model { }
```

Although parameters in Stan models may have improper priors, we do not want improper posterior distributions, as we are trying to use these distributions for Bayesian inference. There is no general way to check if a posterior distribution is improper. But if all the priors are proper, the posterior will be proper also.

Each statement in a Stan model is directly translated into the C++ code for computing the log posterior. Thus, for example, the following pair of statements is legal in a Stan model:

```stan
y ~ normal(0,1);
y ~ normal(2,3);
```

The second line here does not simply overwrite the first; rather, both statements contribute to the density function that is evaluated. The above two lines have the
effect of including the product, \( \text{normal}(y \mid 0, 1) \times \text{normal}(y \mid 2, 3) \), into the density function.

For a perhaps more confusing example, consider the following two lines in a Stan model:

\[
\begin{align*}
x & \sim \text{normal}(0.8 \times y, \sigma); \\
y & \sim \text{normal}(0.8 \times x, \sigma);
\end{align*}
\]

At first, this might look like a joint normal distribution with a correlation of 0.8. But it is not. The above are not interpreted as conditional entities; rather, they are factors in the joint density. Multiplying them gives, \( \text{normal}(x \mid 0.8y, \sigma) \times \text{normal}(y \mid 0.8x, \sigma) \), which is what it is (you can work out the algebra) but it is not the joint distribution where the conditionals have regressions with slope 0.8.

With censoring and truncation, Stan uses the censored-data or truncated-data likelihood—this is not always done in BUGS. All of the approaches to censoring and truncation discussed in Andrew Gelman et al. (2013) and Andrew Gelman and Hill (2007) may be implemented in Stan directly as written.

Stan, like BUGS, can benefit from human intervention in the form of reparameterization.

33.4. Some differences when running from R

Stan can be set up from within R using two lines of code. Follow the instructions for running Stan from R on the Stan web site. You don’t need to separately download Stan and RStan. Installing RStan will automatically set up Stan.

In practice we typically run the same Stan model repeatedly. If you pass RStan the result of a previously fitted model the model will not need be recompiled. An example is given on the running Stan from R pages available from the Stan web site.

When you run Stan, it saves various conditions including starting values, some control variables for the tuning and running of the no-U-turn sampler, and the initial random seed. You can specify these values in the Stan call and thus achieve exact replication if desired. (This can be useful for debugging.)

When running BUGS from R, you need to send exactly the data that the model needs. When running RStan, you can include extra data, which can be helpful when playing around with models. For example, if you remove a variable \( x \) from the model, you can keep it in the data sent from R, thus allowing you to quickly alter the Stan model without having to also change the calling information in your R script.

As in R2WinBUGS and R2jags, after running the Stan model, you can quickly
summarize using `plot()` and `print()`. You can access the simulations themselves using various extractor functions, as described in the RStan documentation.

Various information about the sampler, such as number of leapfrog steps, log probability, and step size, is available through extractor functions. These can be useful for understanding what is going wrong when the algorithm is slow to converge.

### 33.5. The Stan community
Stan, like WinBUGS, OpenBUGS, and JAGS, has an active community, which you can access via the user’s mailing list and the developer’s mailing list; see the Stan web site for information on subscribing and posting and to look at archives.
References


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(1): 59–73.


