

Iterative Solution of Linear Systems

with Improved Arithmetic and Result Verification

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Meinen Eltern, Inge und Armin

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Introduction

⁶⁶ Calvin: You can't just turn on creativity like a faucet. You have to be in the right mood. Hobbes: What mood is that? Calvin: Last-minute panic.

Calvin and Hobbes (by Bill Watterson), 1991

During the last decades, we have had an exponential growth of processor speed and storage capacity. Due to Moore's law, these quantities increase by a factor of two every 18 months. This means, we have about one thousand times the computing power today than we had in 1985, the year when the IEEE floating point standard 754 [4] was released. This standard proposes a 64 bit arithmetic for floating-point operations and up to now there are hardly any improvements innovated by hardware manufactures. The complete computing power is still utilized to increase the size of problems that can be handled.

"There will still be people who say not all the relevant physics is in the models, but that's still a much smaller criticism than not being able to establish firm results for the physics that is being modeled." (John Gustafson, 1998 [54])

One of the largest consumers of floating-point arithmetics are iterative solvers for linear systems of equations. Most of todays big problems in scientific computing, e.g, in structural engineering or fluid dynamics are modeled by differential equations which lead to large linear systems after discretization. Therefore it is an important task to develop accurate and reliable algorithms for this purpose.

Frequently, preconditioned Krylov subspace methods are used to solve these large and often sparse linear systems. Theoretically, Krylov methods have many favorable properties concerning convergence rates, accuracy, computing time, and storage requirements. Unfortunately, these properties prevalently do not hold in the presence of roundoff errors. Computing time increases due to unnecessarily many iterations and expensively computed stopping criteria and more storage is needed, e.g, because of reorthogonalization strategies. Moreover, convergence does sometimes not happen at all or stagnates without delivering the desired accuracy.

There are many investigations which try to quantize the attainable accuracy or convergence rates in finite precision arithmetic. However, the purpose of this work is not to try to get the best result with the given arithmetic. We aim to fix our requirements on final accuracy and then choose an appropriate arithmetic that enables us to meet these requirements.

After identifying the critical parts which mostly suffer from finite precision arithmetic, we selectively introduce arithmetical improvements to reduce the propagated errors and thereby reducing the number of needed iterations.

The work in this thesis is based on recent developments on *state of the art* linear system solvers, on arithmetical tools for verification and highly accurate computing, as well as on high performance object oriented programming. Starting from there, we develop powerful algorithms which are capable to deliver almost any desired accuracy. Additionally, if the system is not too large, we are often able to prove the correctness of our results, i.e., we can give a rigorous upper bound for the error norm. The thesis is structured as follows.

Chapter 1 gives an introduction into the theory of preconditioned Krylov subspace methods. We start with a unified description of subspace methods in general and various Krylov subspace generation techniques. After summarizing the most important results of *infinite precision* convergence theory, we give a detailed introduction into generic preconditioners. Particularly, we focus on splitting techniques and incomplete factorizations with a special emphasis on suitable modifications made for the requirements of our verification methods. Based on these fundamentals, we give a broad overview over Krylov subspace solvers and describe some of the most important variants in more details. For each method we prove the important short recurrence properties and give pseudo-programming language formulation of the preconditioned algorithm.

In **Chapter 2**, we investigate the behavior of preconditioned Krylov Methods in the presence of finite precision arithmetic. After introducing the basic concepts of floating-point numbers and arithmetic with these numbers, we present the central results of *finite precision* theory of Krylov methods. Based on the error analysis of C. Paige, we proved the direct dependency of the level of orthogonality among the Krylov basis vectors on the used arithmetic. Subsequently, we give some examples, demonstrating the influence of rounding errors on preconditioning and solving.

To narrow the gap between exact precision behavior as described in Chapter 1 and finite precision behavior stated in Chapter 2, **Chapter 3** introduces several important techniques which increase the precision and reliability of computer arithmetic. Namely, we describe high precision arithmetics, interval arithmetic, and the exact scalar product.

Chapter 4 summarizes important techniques for computing error bounds for solutions of large linear systems. Introducing the basic concepts of classical verification methods, such as interval Gaussian elimination or interval fixed point methods we pass over into recently developed verification techniques based on perturbation theory. In particular, we describe a fundamental method to bound error norms via residual norms and then deduce a more advanced technique improving these bounds by exploiting the Lanczos-Gauß connection.

The next two chapters focus on implementation techniques on a computer. Chapter 5 gives a broad overview of several powerful programming techniques for writing high performance object oriented numerical linear algebra routines.

Almost all ideas and concepts presented in this thesis are implemented in the **v**ariable precision \mathbf{k} rylov solver $\mathbf{v}\mathbf{k}$. Chapter 6 describes the structure of the code and how to use it. Additionally there is a graphical user interface $\mathbf{x}\mathbf{v}\mathbf{k}$ which is also described in this chapter.

Finally, **Chapter 7** exemplifies the techniques and methods, described in this thesis. Particularly, we show that improved arithmetics are not only capable to deliver more accurate results but also can accelerate convergence significantly. Moreover we present highly precise verified solutions for systems with up to 2000000 unknowns or condition numbers of approximately 10^{62} .

Notation

⁶⁶ Notation is everything. ??

Charles F. van Loan

Throughout this thesis, all matrices are denoted by bold capital letters (\mathbf{A}) , vectors by bold lowercase letters (\mathbf{a}) , and scalar variables by ordinary lowercase letters $(a, or \alpha)$. Interval variables are enclosed in square brackets $([\mathbf{A}], [\mathbf{a}], [\mathbf{a}], or [\alpha])$.

If not mentioned explicitly, all matrices are square with dimension n and the vectors are n-vectors.

By default the column vectors of a matrix are denoted with the same but lowercase letter and the elements are printed with the same letter, too, but in medium weight. For example we have

$$\boldsymbol{A} = (\boldsymbol{a}_1 | \cdots | \boldsymbol{a}_n) = (a_{i,j})_{i,j=1}^n$$
.

This notational convention is also used in the other direction, i.e., when we have a sequence of vectors and need them collected in a matrix. Calligraphic letters (\mathcal{K}) denote (affine) vector-spaces or sets.

Some letters have a predefined sense in this thesis. That is the system matrix A, the right hand side vector b, the exact solution

$$\boldsymbol{x}^* = \boldsymbol{A}^{-1} \boldsymbol{b},$$

any approximation $\tilde{\boldsymbol{x}}$ to the solution, the residual vector

$$r=b-A\tilde{x},$$

the standard basis vectors $\boldsymbol{e}_i = (0, \ldots, 0, 1, 0, \ldots, 0)^T$ with the '1' at the *i*th place, and the identity matrix \boldsymbol{I} . The letter ϵ always denotes the machine precision, but since we deal with different number formats, we also have different values of ϵ . Therefore the actual size is always given in the context if necessary. Eigenvalues are always called λ and singular values σ . Particularly the smallest singular value is denoted with σ_{\min} .

- 5 -

We tried not to use variable names twice while simultaneously respecting traditionally used notational conventions. There is only one exception, where we preferred convention over uniqueness. That are the scalar variables in various Krylov algorithms (α and β) which conflict with scalar coefficients in the (Bi-)Lanczos algorithm.

Table 1 shows the most important used functions and operators with the according definition.

Symbol	Definition
$\langle oldsymbol{x} \mid oldsymbol{y} angle$	scalar product of \boldsymbol{x} and \boldsymbol{y}
$x \perp y$	$oldsymbol{x}$ and $oldsymbol{y}$ are orthogonal, i.e., $\langle oldsymbol{x} \mid oldsymbol{y} angle = 0$
$\operatorname{span}\{oldsymbol{x}_1,\ldots,oldsymbol{x}_n\}$	linear hull of $\{ \boldsymbol{x}_1, \dots, \boldsymbol{x}_n \}$
$\operatorname{rank}({oldsymbol{A}})$	rank of A
$ ho(oldsymbol{A})$	spectral radius of \boldsymbol{A}
$\operatorname{cond}({oldsymbol{A}})$	condition number of \boldsymbol{A}
$\triangle(\cdot), \bigtriangledown(\cdot), \diamondsuit(\cdot), \Box(\cdot)$	round upwards, downwards, to interval, to nearest
$\mathbb{A}, \mathbb{A}, \mathbb{A}, \mathbb{A}$	upward rounded arithmetic operations
$\forall , \forall , \forall , \forall$	downward rounded arithmetic operations
$\diamondsuit,\diamondsuit,\diamondsuit,\diamondsuit$	interval arithmetic operations

Table 1: Functions and operators with the according definition.

CHAPTER

Preconditioned Krylov Subspace Methods

⁶⁶ Ich empfehle Ihnen diesen Modus zur Nachahmung. Schwerlich werden Sie je wieder direkt eliminieren, wenigstens nicht, wenn Sie mehr als zwei Unbekannte haben. Das indirekte [iterative] Verfahren läßt sich halb im Schlaf ausführen oder man kann während desselben an andere Dinge denken.¹ ??

> Carl Friedrich Gauß to Christoph Ludwig Gerling, December 26, 1823

Krylov subspace methods are used both to solve systems of linear equations Ax = band to find eigenvalues of A [21, 22]. In this work we focus on linear system solving, however particularly in investigating theoretical properties of Krylov methods, we also need some facts from eigenvalue theory [96].

Krylov algorithms assume that \boldsymbol{A} is accessible only via a *black-box* subroutine that returns $\boldsymbol{y} = \boldsymbol{A}\boldsymbol{z}$ for any \boldsymbol{z} (and perhaps $\boldsymbol{y} = \boldsymbol{A}^T\boldsymbol{z}$ if \boldsymbol{A} is nonsymmetric). This is an important assumption for several reasons. First, the cheapest non-trivial

¹ "I recommend this method for your imitation. You will hardly ever again eliminate directly, at least not when you have more than two unknowns. The indirect [iterative] procedure can be done while half asleep, or while thinking about other things."

operation that one can perform on a sparse matrix is to multiply it by a vector — if \boldsymbol{A} has *nnz* nonzero entries, a matrix-vector multiplication costs *nnz* multiplications and (at most) *nnz* additions. Secondly, \boldsymbol{A} may not be represented explicitly as a matrix but may be available only as a subroutine for computing \boldsymbol{Az} .

This chapter is organized as follows. In Section 1.1 we give some basic facts about subspace solvers in general. Section 1.2 describes how information about A is extracted via matrix-vector multiplication. In Krylov subspace methods, this information is stored in the so called Krylov subspaces. In Section 1.3 we present some convergence theory for symmetric and nonsymmetric Lanczos-like algorithms. Since for real life problems, one will hardly ever solve a linear system of equations without preconditioning, we describe the basic facts of preconditioning and introduce the most important generic preconditioners in Section 1.4. Based on these introductory sections, Section 1.5 gives an overview about several Krylov solvers and describes some of the most important variants in more detail.

1.1 Subspace Methods

The basic idea of subspace methods is generating a sequence of subspaces \mathcal{V}_m with increasing dimension m and finding a vector \boldsymbol{x}_m within each of these subspaces that is in some sense an *optimal* approximation in \mathcal{V}_m to the solution \boldsymbol{x}^* of the entire problem. Clearly, this optimality measure has to guarantee that we choose $\boldsymbol{x}_m = \boldsymbol{x}^*$ if $\boldsymbol{x}^* \in \mathcal{V}_m$ (at the latest if m = n). Therefore, designing a subspace method is subdivided in two tasks. First we have to define the sequence of subspaces and secondly we have to decide in which way we select a vector out of each subspace, i.e., which condition we provide to $\boldsymbol{x}_m \in \mathcal{V}_m$ in order to get a *good* approximation for \boldsymbol{x}^* [20].

Let us assume for the moment that we already have chosen these subspaces \mathcal{V}_m and now look for a criterion to select $\boldsymbol{x}_m \in \mathcal{V}_m$. From approximation theory we know that an optimal subspace approximation \boldsymbol{x}_m is characterized by the fact that the error $\boldsymbol{x}^* - \boldsymbol{x}_m$ stays orthogonal on the subspace where \boldsymbol{x}_m is chosen from, i.e.,

$$\boldsymbol{x}^* - \boldsymbol{x}_m \perp \boldsymbol{\mathcal{V}}_m \tag{1.1}$$

must hold. Unfortunately, we do not know $\boldsymbol{x}^* - \boldsymbol{x}_m$ since we do not know the exact solution \boldsymbol{x}^* . However, we can compute the residual $\boldsymbol{r}_m := \boldsymbol{b} - \boldsymbol{A}\boldsymbol{x}_m$ which in some sense also is a measure for the *quality* of \boldsymbol{x}_m . Thus, a first idea might be simply to replace $\boldsymbol{x}^* - \boldsymbol{x}_m$ by \boldsymbol{r}_m in (1.1).

Since we have

$$x^* - x_m = A^{-1}b - A^{-1}Ax_m = A^{-1}r_m$$

it might be advantageous to choose $\boldsymbol{x}_m \in \mathcal{V}_m$ satisfying

$$\boldsymbol{A}^{-1}\boldsymbol{r}_m \perp \boldsymbol{\mathcal{V}}_m \quad \Leftrightarrow \quad \boldsymbol{r}_m \perp \boldsymbol{A}\boldsymbol{\mathcal{V}}_m, \tag{1.2}$$

that is, \boldsymbol{r}_m may not stay orthogonal on \mathcal{V}_m but on another subspace, say \mathcal{W}_m .

We fix these ideas in the following definition.

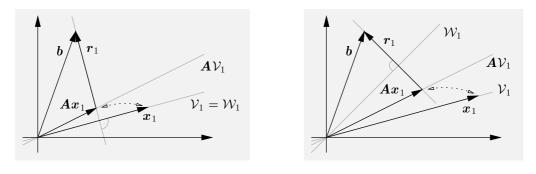
Definition 1.1 A projecting method for solving a linear system Ax = b is a procedure that constructs approximate solutions $x_m \in \mathcal{V}_m$ under the constraint

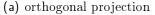
$$\boldsymbol{r}_m = \boldsymbol{b} - \boldsymbol{A} \boldsymbol{x}_m \perp \mathcal{W}_m \tag{1.3}$$

where \mathcal{V}_m and \mathcal{W}_m are m-dimensional subspaces of \mathbb{R}^n .

In the case $\mathcal{W}_m = \mathcal{V}_m$ we have an orthogonal projecting method and (1.3) is called a Galerkin condition whereas the general case $(\mathcal{W}_m \neq \mathcal{V}_m)$ is called a skew or oblique projecting method with a Petrov-Galerkin condition in equation (1.3).

In Figure 1.1 we illustrate the case n = 2, m = 1. Given \mathcal{V}_1 we select $\boldsymbol{x}_1 \in \mathcal{V}_1$ to satisfy $\boldsymbol{r}_1 = \boldsymbol{b} - \boldsymbol{A}\boldsymbol{x}_1 \perp \mathcal{V}_1$ or $\boldsymbol{r}_1 \perp \mathcal{W}_1$ in the case of skew projections.





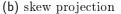


Figure 1.1: Projecting methods. Given \mathcal{V}_1 we compute $A\mathcal{V}_1 = \{Ax \mid x \in \mathcal{V}_1\}$ and then select $x_1 \in \mathcal{V}_1$ such that $r_1 = b - Ax_1 \perp \mathcal{V}_1$, respectively $r_1 \perp \mathcal{W}_1$ in the case of skew projections.

Another way to characterize an optimal approximation is by its error norm. That means $\boldsymbol{x}_m \in \mathcal{V}_m$ is called optimal if $\|\boldsymbol{x}^* - \boldsymbol{x}_m\|$ minimizes $\|\boldsymbol{x}^* - \boldsymbol{x}\|$ for all $\boldsymbol{x} \in \mathcal{V}_m$. Again we have to replace $\boldsymbol{x}^* - \boldsymbol{x}_m$ by \boldsymbol{r}_m (because we generally do not know \boldsymbol{x}^*) and fix our ideas in the following definition.

Definition 1.2 A norm minimizing method for solving a linear system Ax = b is a procedure that constructs approximate solutions $x_m \in \mathcal{V}_m$ under the constraint

$$\|\boldsymbol{r}_m\|_2 = \|\boldsymbol{b} - \boldsymbol{A}\boldsymbol{x}_m\|_2 = \min_{\boldsymbol{x}\in\mathcal{V}_m} \{\|\boldsymbol{b} - \boldsymbol{A}\boldsymbol{x}\|_2\}$$
(1.4)

where \mathcal{V}_m is an m-dimensional subspace of \mathbb{R}^n .

We show the case n = 2, m = 1 in Figure 1.2. Given \mathcal{V}_1 we select $\boldsymbol{x}_1 \in \mathcal{V}_1$ minimizing $\|\boldsymbol{b} - \boldsymbol{A}\boldsymbol{x}\|_2$ for all $\boldsymbol{x} \in \mathcal{V}_1$.

1.2 Generating Krylov Spaces

In this section we focus on the question of how to select the subspaces \mathcal{V}_m . To take a possibly given initial guess \boldsymbol{x}_0 into account, we allow \mathcal{V}_m to be an affine subspace,

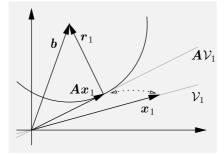


Figure 1.2: Norm minimizing methods. Given \mathcal{V}_1 we compute $A\mathcal{V}_1 = \{Ax \mid x \in \mathcal{V}_1\}$ and then select $x_1 \in \mathcal{V}_1$ to satisfy $\|r_1\|_2 = \|b - Ax_1\|_2 = \min_{x \in \mathcal{V}_1} \{\|b - Ax\|_2\}$.

i.e., $\mathcal{V}_m = \boldsymbol{x}_0 + \tilde{\mathcal{V}}_m$. There are various possibilities to choose $\tilde{\mathcal{V}}_m$ but it turns out that using Krylov subspaces has several advantageous properties as we will see later on.

Definition 1.3 A Krylov subspace method is a projecting or a norm minimizing method (see Definitions 1.1 and 1.2) to solve a linear system Ax = b where the subspaces $\tilde{\mathcal{V}}_m$ are chosen as Krylov subspaces

$$\tilde{\mathcal{V}}_m = \mathcal{K}_m(\boldsymbol{A}, \boldsymbol{r}_0) := \operatorname{span}\{\boldsymbol{r}_0, \boldsymbol{A}\boldsymbol{r}_0, \dots, \boldsymbol{A}^{m-1}\boldsymbol{r}_0\}, \quad m = 1, 2, \dots$$
(1.5)

with $\boldsymbol{r}_0 = \boldsymbol{b} - \boldsymbol{A}\boldsymbol{x}_0$.

Since we intend to work with vectors out of $\mathcal{K}_m(\mathbf{A}, \mathbf{r}_0)$ we have to find a handy representation for them. One of the most powerful tricks in linear algebra that often simplifies problems significantly, is to find a *suitable* basis of the vector space we have to work with. Often orthonormal bases are a good choice but it is generally an enormous amount of work to compute one. The Arnoldi algorithm (Section 1.2.1) and the Lanczos algorithm (Section 1.2.2) are methods to compute orthonormal bases of Krylov spaces. In Section 1.2.3 we will see that there is a cheaper way to get a useful basis, although it is not orthonormal anymore.

1.2.1 Arnoldi Algorithm

The Krylov space $\mathcal{K}_m(\boldsymbol{A}, \boldsymbol{r}_0)$ is given as the linear hull of $\{\boldsymbol{r}_0, \boldsymbol{A}\boldsymbol{r}_0, \ldots, \boldsymbol{A}^{m-1}\boldsymbol{r}_0\}$ and a well known technique for orthonormalizing a sequence of vectors is the Gram-Schmidt algorithm. In the case m = 1, we simply have $\mathcal{K}_1(\boldsymbol{A}, \boldsymbol{r}_0) = \operatorname{span}\{\boldsymbol{r}_0\}$ and thus $\boldsymbol{v}_1 := \boldsymbol{r}_0 / \|\boldsymbol{r}_0\|$ is a orthonormal basis (ONB) for this one dimensional Krylov space. Now suppose we have already an ONB $\boldsymbol{V}_m := \{\boldsymbol{v}_1, \ldots, \boldsymbol{v}_m\}$ for $\mathcal{K}_m(\boldsymbol{A}, \boldsymbol{r}_0)$ and look for \boldsymbol{v}_{m+1} to extend \boldsymbol{V}_m to \boldsymbol{V}_{m+1} . With Gram-Schmidt we compute \boldsymbol{v}_{m+1} as follows

$$\begin{split} \tilde{\boldsymbol{v}}_{m+1} &= \boldsymbol{A}\boldsymbol{v}_m & (\text{compute a prototype for } \boldsymbol{v}_{m+1}) \\ \tilde{\boldsymbol{v}}_{m+1} &\leftarrow \tilde{\boldsymbol{v}}_{m+1} - \sum_{i=1}^m \underbrace{\langle \tilde{\boldsymbol{v}}_{m+1} \mid \boldsymbol{v}_i \rangle}_{=:h_{im}} \boldsymbol{v}_i & (\text{orthogonalize it against } \boldsymbol{V}_m) & (1.6) \\ \boldsymbol{v}_{m+1} &= \underbrace{\tilde{\boldsymbol{v}}_{m+1}}_{=:h_{m+1,m}} &. & (\text{and finally normalize it}) \end{split}$$

In the context of Krylov spaces this algorithm is called Arnoldi algorithm [6] or full orthogonalization method (FOM) because we orthogonalize $\tilde{\boldsymbol{v}}_{m+1}$ against all previous basis vectors. Collecting the coefficients $h_{i,j}$ to vectors $\boldsymbol{h}_i := (h_{i,1}|\cdots|h_{i,m})^T$ we get

$$oldsymbol{v}_{m+1}h_{m+1,m}=oldsymbol{A}oldsymbol{v}_m-\sum_{i=1}^moldsymbol{v}_ih_{i,m}=oldsymbol{A}oldsymbol{v}_m-oldsymbol{V}_moldsymbol{h}_m$$

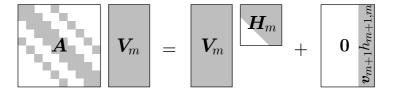
and further $\operatorname{arranging}^2 \boldsymbol{H}_m := (\boldsymbol{h}_1 \dots \boldsymbol{h}_m)$ yields

$$\boldsymbol{A}\boldsymbol{V}_{m} = \boldsymbol{V}_{m}\boldsymbol{H}_{m} + h_{m+1,m} \cdot \boldsymbol{v}_{m+1}\boldsymbol{e}_{m}^{T}, \qquad (1.7)$$

where \boldsymbol{H}_m is the upper Hessenberg matrix of recurrence coefficients

$$\boldsymbol{H}_{m} = \begin{pmatrix} h_{1,1} & h_{1,2} & \cdots & h_{1,m-1} & h_{1,m} \\ h_{2,1} & h_{2,2} & & \vdots & & \vdots \\ & & h_{3,2} & & \vdots & & \vdots \\ & & & \ddots & & \vdots & & \vdots \\ & & & & h_{m,m-1} & h_{m,m} \end{pmatrix}.$$

Pictorially, this matrix equation looks like



Representing $\boldsymbol{x}_m \in \mathcal{V}_m$ as $\boldsymbol{x}_m = \boldsymbol{x}_0 + \boldsymbol{V}_m \boldsymbol{\xi}_m$, i.e., \boldsymbol{x}_m is a shifted linear combination of $\boldsymbol{v}_1, \ldots, \boldsymbol{v}_m$, the Galerkin condition $\boldsymbol{b} - \boldsymbol{A}\boldsymbol{x}_m \perp \mathcal{V}_m$ now writes as

$$V_m^T(\boldsymbol{b} - \boldsymbol{A}\boldsymbol{x}_m) = V_m^T(\boldsymbol{b} - \boldsymbol{A}\boldsymbol{x}_0 - \boldsymbol{A}V_m\boldsymbol{\xi}_m) = 0$$

$$\Leftrightarrow \underbrace{V_m^T \boldsymbol{A}V_m}_{\boldsymbol{A}|_{\boldsymbol{k}_m(\boldsymbol{A},\boldsymbol{r}_0)}} \boldsymbol{\xi}_m = V_m^T \boldsymbol{r}_0.$$

 $^{^2{\}rm Here}$ and further on we collect vectors of different dimensions by extending them with trailing zeros to common length.

This also clarifies the name orthogonal projecting method because the restricted system matrix $A|_{\mathcal{K}_m(\boldsymbol{A},\boldsymbol{r}_0)}$ is obtained by projection with the orthogonal projector V_m . Multiplying equation (1.7) with V_m^T from left yields

$$V_m^T A V_m = \underbrace{V_m^T V_m}_I H_m + h_{m+1,m} \cdot \underbrace{V_m^T v_{m+1}}_0 \cdot e_m^T = H_m.$$

Remembering $\boldsymbol{v}_1 = \boldsymbol{r}_0 / \|\boldsymbol{r}_0\|$ we finally get

$$\boldsymbol{b} - \boldsymbol{A} \boldsymbol{x}_m \perp \mathcal{V}_m \quad \Leftrightarrow \quad \boldsymbol{H}_m \boldsymbol{\xi}_m = \| \boldsymbol{r}_0 \| \boldsymbol{e}_1.$$
 (1.8)

Hence, the projected system matrix is not only of smaller dimension m but is also particularly structured (upper Hessenberg).

Note that the Arnoldi algorithm can terminate before m = n if $\|\tilde{\boldsymbol{v}}_{m+1}\| = 0$. Fortunately, in this case $\mathcal{K}_m(\boldsymbol{A}, \boldsymbol{r}_0)$ is an \boldsymbol{A} -invariant subspace, i.e., $\mathcal{K}_m(\boldsymbol{A}, \boldsymbol{r}_0) = \boldsymbol{A}\mathcal{K}_m(\boldsymbol{A}, \boldsymbol{r}_0)$. Therefore we have

$$\begin{aligned} \boldsymbol{r}_{0} \in \mathcal{K}_{m}(\boldsymbol{A}, \boldsymbol{r}_{0}) &= \boldsymbol{A}\mathcal{K}_{m}(\boldsymbol{A}, \boldsymbol{r}_{0}) \\ \Leftrightarrow \quad \boldsymbol{A}^{-1}\boldsymbol{r}_{0} \in \mathcal{K}_{m}(\boldsymbol{A}, \boldsymbol{r}_{0}) \\ \Leftrightarrow \quad \boldsymbol{x}^{*} - \boldsymbol{x}_{0} \in \mathcal{K}_{m}(\boldsymbol{A}, \boldsymbol{r}_{0}) \\ \Leftrightarrow \quad \boldsymbol{x}^{*} \in \boldsymbol{x}_{0} + \mathcal{K}_{m}(\boldsymbol{A}, \boldsymbol{r}_{0}) = \mathcal{V}_{m}. \end{aligned}$$

$$(1.9)$$

That is, in this *break down* situation we can already find the solution x^* in the shifted Krylov space computed so far.

Since the Gram-Schmidt algorithm tends to be unstable if angles between Av_m and v_1, \ldots, v_m are small, we use a computationally more robust variant, the so called modified Gram-Schmidt algorithm. Here Av_m is successively orthogonalized. In a pseudo programming language we can write the Arnoldi procedure with modified Gram-Schmidt as shown in Algorithm 1.1.

Given
$$\boldsymbol{x}_0$$

 $\tilde{\boldsymbol{v}}_1 = \boldsymbol{b} - \boldsymbol{A} \boldsymbol{x}_0$
 $\boldsymbol{v}_1 = \tilde{\boldsymbol{v}}_1 / \| \tilde{\boldsymbol{v}}_1 \|$
for $m = 1, 2, \dots$
 $\tilde{\boldsymbol{v}}_{m+1} = \boldsymbol{A} \boldsymbol{v}_m$
for $j = 1, \dots, m$
 $h_{j,m} = \langle \tilde{\boldsymbol{v}}_{m+1} \mid \boldsymbol{v}_j \rangle$
 $\tilde{\boldsymbol{v}}_{m+1} = \tilde{\boldsymbol{v}}_{m+1} - h_{j,m} \boldsymbol{v}_j$
 $h_{m+1,m} = \| \tilde{\boldsymbol{v}}_{m+1} \|$
 $\boldsymbol{v}_{m+1} = \tilde{\boldsymbol{v}}_{m+1} / h_{m+1,m}$

Algorithm 1.1: Arnoldi algorithm with a modified Gram-Schmidt procedure.

Unfortunately, the Arnoldi algorithm is expensive in memory and computing time since we need access to all m previous basis vectors and have to perform O(nnz + nm) operations in the mth iteration.

1.2.2 Lanczos Algorithm

If A is symmetric the situation becomes much more favorable. We have

$$\boldsymbol{H}_m = \boldsymbol{V}_m^T \boldsymbol{A} \boldsymbol{V}_m = (\boldsymbol{V}_m^T \boldsymbol{A}^T \boldsymbol{V}_m)^T = \boldsymbol{H}_m^T,$$

that is H_m turns out to be symmetric, too. In this case H_m is a symmetric upper Hessenberg and therefore a symmetric tridiagonal matrix which is denoted with T_m , where

$$\boldsymbol{T}_{m} = \begin{pmatrix} \alpha_{1} & \beta_{1} & & \\ \beta_{1} & \alpha_{2} & \beta_{2} & & \\ & \ddots & \ddots & \ddots & \\ & & \beta_{m-2} & \alpha_{m-1} & \beta_{m-1} \\ & & & & \beta_{m-1} & \alpha_{m} \end{pmatrix}.$$

The recursion of depth m in (1.6) reduces to one of depth three.

$$\begin{split} \tilde{\boldsymbol{v}}_{m+1} &= \boldsymbol{A}\boldsymbol{v}_m - \beta_{m-1}\boldsymbol{v}_{m-1} & \text{(compute a prototype for } \boldsymbol{v}_{m+1} \\ & \text{and orthogonalize it against } \boldsymbol{v}_{m-1} \text{)} \\ \tilde{\boldsymbol{v}}_{m+1} &\leftarrow \tilde{\boldsymbol{v}}_{m+1} - \underbrace{\langle \tilde{\boldsymbol{v}}_{m+1} \mid \boldsymbol{v}_m \rangle}_{=:\alpha_m} \boldsymbol{v}_m & \text{(orthogonalize it against } \boldsymbol{v}_m \text{)} & (1.10) \\ \boldsymbol{v}_{m+1} &= \underbrace{\tilde{\boldsymbol{v}}_{m+1}}_{=:\beta_m} \text{.} & \text{(and finally normalize it)} \end{split}$$

This special case of the Arnoldi algorithm for symmetric systems is called Lanczos algorithm [77]. Similar to the nonsymmetric case we have

$$AV_{m} = V_{m}T_{m} + \beta_{m}v_{m+1}e_{m}^{T}$$

$$\Leftrightarrow V_{m}^{T}AV_{m} = \underbrace{V_{m}^{T}V_{m}}_{I}T_{m} + \beta_{m} \cdot \underbrace{V_{m}^{T}v_{m+1}}_{0} \cdot e_{m}^{T} = T_{m}.$$

$$(1.11)$$

and thus

$$\boldsymbol{b} - \boldsymbol{A}\boldsymbol{x}_m \perp \mathcal{V}_m \quad \Leftrightarrow \quad \boldsymbol{T}_m \boldsymbol{\xi}_m = \|\boldsymbol{r}_0\| \boldsymbol{e}_1.$$
 (1.12)

That is, the Galerkin condition reduces to a symmetric tridiagonal system of dimension m.

Algorithm 1.2 shows the Lanczos procedure in a pseudo programming language.

In the Lanczos algorithm we only need access to the last three basis vectors (no increasing memory requirements per iteration) and the number of operation is O(nnz). Unfortunately, this three-term-recurrence which makes the Lanczos algorithm so favorable, only exists for symmetric matrices A, at least if we use orthogonal projecting methods.

1.2.3 Bi-Lanczos Algorithm

Analyzing the Arnoldi and Lanczos algorithm we find that the most important property was the simple structure of $A_{\mathcal{K}_m(\mathcal{A}, r_0)}$. We now try to retain the short recurrences of the Lanczos algorithm which led to the tridiagonal shape of $A_{\mathcal{K}_m(\mathcal{A}, r_0)}$.

Given \boldsymbol{x}_0 $\tilde{\boldsymbol{v}}_1 = \boldsymbol{b} - \boldsymbol{A} \boldsymbol{x}_0$ for $m = 1, 2, \dots$ $\beta_{m-1} = \| \tilde{\boldsymbol{v}}_m \|$ $\boldsymbol{v}_m = \tilde{\boldsymbol{v}}_m / \beta_{m-1}$ $\tilde{\boldsymbol{v}}_{m+1} = \boldsymbol{A} \boldsymbol{v}_m$ if m > 1 $\tilde{\boldsymbol{v}}_{m+1} = \tilde{\boldsymbol{v}}_{m+1} - \beta_{m-1} \boldsymbol{v}_{m-1}$ $\alpha_m = \langle \tilde{\boldsymbol{v}}_{m+1} \mid \boldsymbol{v}_m \rangle$ $\tilde{\boldsymbol{v}}_{m+1} = \tilde{\boldsymbol{v}}_{m+1} - \alpha_m \boldsymbol{v}_m$

Algorithm 1.2: Lanczos algorithm for symmetric system matrices A.

but without the necessity of symmetry [112]. To achieve this, we construct a skew projecting method, that is, we choose $\boldsymbol{x}_m \in \mathcal{V}_m$ as $\boldsymbol{x}_m = \boldsymbol{x}_0 + \boldsymbol{V}_m \boldsymbol{\xi}_m$ to satisfy the Petrov-Galerkin condition

$$egin{aligned} m{W}_m^T(m{b}-m{A}m{x}_m) &= m{W}_m^T(m{b}-m{A}m{x}_0-m{A}m{V}_mm{\xi}_m) &= 0 \ &\Leftrightarrow m{W}_m^Tm{A}m{V}_mm{\xi}_m &= m{W}_m^Tm{r}_0, \end{aligned}$$

Here we have to compute two sets of vectors: $V_m = (v_1 \dots v_m)$ and simultaneously $W_m = (w_1 \dots w_m)$, providing a *simple* structure of $W_m^T A V_m$. This can be achieved by defining V_m and W_m via a pair of coupled three-term-recurrences

$$\alpha_{m} = \langle \boldsymbol{A}\boldsymbol{v}_{m} \mid \boldsymbol{w}_{m} \rangle \qquad \dots = \langle \boldsymbol{v}_{m} \mid \boldsymbol{A}^{T}\boldsymbol{w}_{m} \rangle$$

$$\tilde{\boldsymbol{v}}_{m+1} = \boldsymbol{A}\boldsymbol{v}_{m} - \alpha_{m}\boldsymbol{v}_{m} - \beta_{m-1}\boldsymbol{v}_{m-1} \qquad \tilde{\boldsymbol{w}}_{m+1} = \boldsymbol{A}^{T}\boldsymbol{w}_{m} - \alpha_{m}\boldsymbol{w}_{m} - \gamma_{m-1}\boldsymbol{w}_{m-1}$$

$$\gamma_{m} = \|\tilde{\boldsymbol{v}}_{m+1}\| \qquad \beta_{m} = \langle \tilde{\boldsymbol{v}}_{m+1} \mid \tilde{\boldsymbol{w}}_{m+1} \rangle / \gamma_{m} \qquad (1.13)$$

$$\boldsymbol{v}_{m+1} = \tilde{\boldsymbol{v}}_{m+1} / \gamma_{m} \qquad \boldsymbol{w}_{m+1} = \tilde{\boldsymbol{w}}_{m+1} / \beta_{m} ,$$

starting with $\boldsymbol{v}_1 = \boldsymbol{r}_0/\|\boldsymbol{r}_0\|$ and $\boldsymbol{w}_1 = \boldsymbol{r}_0^{dual}/\|\boldsymbol{r}_0^{dual}\|$ with $\langle \boldsymbol{r}_0 \mid \boldsymbol{r}_0^{dual} \rangle \neq 0$, e.g. $\boldsymbol{r}_0 = \boldsymbol{r}_0^{dual}$.

In matrix form these recurrences can be written as

$$AV_m = V_m T_m + \gamma_m v_{m+1} e_m^T$$

$$A^T W_m = W_m T_m^T + \beta_m w_{m+1} e_m^T,$$
(1.14)

where T_m is the *m*-by-*m* tridiagonal matrix of recurrence coefficients

$$\boldsymbol{T}_{m} = \begin{pmatrix} \alpha_{1} & \beta_{1} & & & \\ \gamma_{1} & \alpha_{2} & \beta_{2} & & \\ & \ddots & \ddots & \ddots & \\ & & \gamma_{m-2} & \alpha_{m-1} & \beta_{m-1} \\ & & & \gamma_{m-1} & \alpha_{m} \end{pmatrix}$$

It turns out that V_m is a basis for $\mathcal{K}_m(\boldsymbol{A}, \boldsymbol{r}_0)$ and W_m is a basis for $\mathcal{K}_m(\boldsymbol{A}^T, \boldsymbol{r}_0)$. Multiplying the upper equation in (1.14) with W_m^T from left yields

$$\boldsymbol{W}_{m}^{T}\boldsymbol{A}\boldsymbol{V}_{m} = \boldsymbol{W}_{m}^{T}\boldsymbol{V}_{m}\boldsymbol{T}_{m} + \gamma_{m}\boldsymbol{W}_{m}^{T}\boldsymbol{v}_{m+1}\boldsymbol{e}_{m}^{T}.$$
(1.15)

To obtain the desired tridiagonal structure of $\boldsymbol{W}_m^T \boldsymbol{A} \boldsymbol{V}_m$ we need the so called biorthogonality condition for \boldsymbol{V}_m and \boldsymbol{W}_m , i.e., $\boldsymbol{W}_m^T \boldsymbol{V}_m = \boldsymbol{V}_m^T \boldsymbol{W}_m = \boldsymbol{I}$.

Since it is not clear in advance that this mutual orthogonality between W_m and V_m holds, we prove it in the following theorem (compare [48]).

Theorem 1.1 Suppose $\mathbf{v}_1, \ldots, \mathbf{v}_{m+1}$ and $\mathbf{w}_1, \ldots, \mathbf{w}_{m+1}$ exist, that is, $\langle \mathbf{v}_j | \mathbf{w}_j \rangle \neq 0$ for $j = 1, \ldots, m+1$. Then $\mathbf{V}_m^T \mathbf{W}_m = \mathbf{W}_m^T \mathbf{V}_m = \mathbf{I}$.

Proof: Since $\boldsymbol{v}_j = \tilde{\boldsymbol{v}}_j/\gamma_{j-1}$ and $\boldsymbol{w}_j = \tilde{\boldsymbol{w}}_j/\beta_{j-1}$ we have

$$\langle \boldsymbol{v}_j \mid \boldsymbol{w}_j \rangle = \frac{1}{\gamma_{j-1}\beta_{j-1}} \langle \tilde{\boldsymbol{v}}_j \mid \tilde{\boldsymbol{w}}_j \rangle = \frac{\langle \tilde{\boldsymbol{v}}_j \mid \tilde{\boldsymbol{w}}_j \rangle}{\|\langle \tilde{\boldsymbol{v}}_j \mid \tilde{\boldsymbol{w}}_j \rangle\|} = 1$$

We prove the orthogonality of \boldsymbol{v}_i and \boldsymbol{w}_j for $i \neq j$ with $i, j \leq m+1$ by induction.

We have $\langle \boldsymbol{v}_1 | \boldsymbol{w}_1 \rangle = \|\boldsymbol{r}_0\|^{-1} \|\boldsymbol{r}_0^{dual}\|^{-1} \langle \boldsymbol{r}_0 | \boldsymbol{r}_0^{dual} \rangle \neq 0$. Assume that $\langle \boldsymbol{v}_i | \boldsymbol{w}_j \rangle = 0$ holds for $i \neq j$ with $i, j \leq m$. Because of the symmetry of (1.14) we only have to show $\langle \boldsymbol{v}_{m+1} | \boldsymbol{w}_j \rangle = 0$ for $j \leq m$. Then we get for j = m

$$\langle \boldsymbol{v}_{m+1} \mid \boldsymbol{w}_{m} \rangle = \frac{1}{\gamma_{m}} \langle \boldsymbol{A} \boldsymbol{v}_{m} - \alpha_{m} \boldsymbol{v}_{m} - \beta_{m-1} \boldsymbol{v}_{m-1} \mid \boldsymbol{w}_{m} \rangle$$

$$= \frac{1}{\gamma_{m}} \left(\underbrace{\langle \boldsymbol{A} \boldsymbol{v}_{m} \mid \boldsymbol{w}_{m} \rangle}_{=\alpha_{m}} - \alpha_{m} \underbrace{\langle \boldsymbol{v}_{m} \mid \boldsymbol{w}_{m} \rangle}_{=1} - \beta_{m-1} \underbrace{\langle \boldsymbol{v}_{m-1} \mid \boldsymbol{w}_{m} \rangle}_{=0} \right)$$

$$= \frac{1}{\gamma_{m}} (\alpha_{m} - \alpha_{m}) = 0,$$

for j = m - 1 we get

$$\langle \boldsymbol{v}_{m+1} \mid \boldsymbol{w}_{m-1} \rangle = \frac{1}{\gamma_m} \langle \boldsymbol{A} \boldsymbol{v}_m - \alpha_m \boldsymbol{v}_m - \beta_{m-1} \boldsymbol{v}_{m-1} \mid \boldsymbol{w}_{m-1} \rangle$$

$$= \frac{1}{\gamma_m} (\langle \boldsymbol{A} \boldsymbol{v}_m \mid \boldsymbol{w}_{m-1} \rangle - \alpha_m \underbrace{\langle \boldsymbol{v}_m \mid \boldsymbol{w}_{m-1} \rangle}_{=0} - \beta_{m-1} \underbrace{\langle \boldsymbol{v}_{m-1} \mid \boldsymbol{w}_{m-1} \rangle}_{=1})$$

$$= \frac{1}{\gamma_m} (\langle \boldsymbol{v}_m \mid \boldsymbol{A}^T \boldsymbol{w}_{m-1} \rangle - \beta_{m-1})$$

$$= \frac{1}{\gamma_m} (\langle \boldsymbol{v}_m \mid \tilde{\boldsymbol{w}}_m + \alpha_{m-1} \boldsymbol{w}_{m-1} + \gamma_{m-2} \boldsymbol{w}_{m-2} \rangle - \beta_{m-1})$$

$$= \frac{1}{\gamma_m} (\beta_{m-1} \underbrace{\langle \boldsymbol{v}_m \mid \boldsymbol{w}_m \rangle}_{=1} + \alpha_{m-1} \underbrace{\langle \boldsymbol{v}_m \mid \boldsymbol{w}_{m-1} \rangle}_{=0} + \gamma_{m-2} \underbrace{\langle \boldsymbol{v}_m \mid \boldsymbol{w}_{m-2} \rangle}_{=0} - \beta_{m-1})$$

$$= \frac{1}{\gamma_m} (\beta_{m-1} - \beta_{m-1}) = 0,$$

and finally for j < m - 1 we get

$$\langle \boldsymbol{v}_{m+1} \mid \boldsymbol{w}_j \rangle = \frac{1}{\gamma_m} \langle \boldsymbol{A} \boldsymbol{v}_m \mid \boldsymbol{w}_j \rangle = \frac{1}{\gamma_m} \langle \boldsymbol{v}_m \mid \boldsymbol{A}^T \boldsymbol{w}_j \rangle = \frac{\beta_j}{\gamma_m} \langle \boldsymbol{v}_m \mid \boldsymbol{w}_{j+1} \rangle = 0.$$

This means, the Bi-Lanczos algorithm provides a short recurrence formula to compute bases for Krylov spaces for nonsymmetric matrices A. Although these bases are not (self) orthonormal, they suffice to obtain Lanczos like short recurrences.

However, we stress that the Bi-Lanczos process can terminate in two different situations [114]. First, if $\|\tilde{\boldsymbol{v}}_{m+1}\| = 0$ or $\|\tilde{\boldsymbol{w}}_{m+1}\| = 0$, then the algorithm has found an \boldsymbol{A} -invariant subspace with $\boldsymbol{x}^* \in \mathcal{V}_m$ or an \boldsymbol{A}^T -invariant subspace with $\boldsymbol{x}^* \in \mathcal{W}_m$, respectively. This is referred to as *regular termination* because we can find the solution in the Krylov spaces of the previous step (compare equation (1.9) on page 12).

The second case, called *serious breakdown*, occurs when $\langle \tilde{\boldsymbol{v}}_{m+1} | \tilde{\boldsymbol{w}}_{m+1} \rangle = 0$ but neither $\tilde{\boldsymbol{v}}_{m+1} = 0$ nor $\tilde{\boldsymbol{w}}_{m+1} = 0$. Hence we have no invariant subspace and thus cannot guarantee to find \boldsymbol{x}^* . However, in some later step, say m + l, there might exist nonzero vectors $\tilde{\boldsymbol{v}}_{m+l} \in \mathcal{K}_{m+l}(\boldsymbol{A}, \boldsymbol{r}_0)$ and $\tilde{\boldsymbol{w}}_{m+l} \in \mathcal{K}_{m+l}(\boldsymbol{A}^T, \boldsymbol{r}_0)$ with $(\boldsymbol{w}_1 | \cdots | \boldsymbol{w}_m | \boldsymbol{w}_{m+l})^T \cdot (\boldsymbol{v}_1 | \cdots | \boldsymbol{v}_m | \boldsymbol{v}_{m+l}) = \boldsymbol{I}$. Thus we have to skip these l-1intermediate steps. For practical implementations, it turns out that we also have to cover *near breakdown situations* where $\langle \tilde{\boldsymbol{v}}_{m+1} | \tilde{\boldsymbol{w}}_{m+1} \rangle$ is sufficiently small to cause numerical instabilities. This technique is called *look-ahead* and is described in further details in several papers including [14, 15, 63, 99, 102].

In Algorithm 1.3 we formulate the Bi-Lanczos process without look-ahead in a pseudo programming language.

Given
$$\boldsymbol{x}_{0}$$

 $\tilde{\boldsymbol{v}}_{1} = \boldsymbol{b} - \boldsymbol{A}\boldsymbol{x}_{0}$
 $\boldsymbol{v}_{1} = \boldsymbol{w}_{1} = \tilde{\boldsymbol{v}}_{1}/\|\tilde{\boldsymbol{v}}_{1}\|$
for $m = 1, 2, ...$
 $\tilde{\boldsymbol{v}}_{m+1} = \boldsymbol{A}\boldsymbol{v}_{m}$
 $\tilde{\boldsymbol{w}}_{m+1} = \boldsymbol{A}^{T}\boldsymbol{w}_{m}$
 $\alpha_{m} = \langle \tilde{\boldsymbol{v}}_{m+1} \mid \boldsymbol{w}_{m} \rangle$
 $\tilde{\boldsymbol{v}}_{m+1} = \tilde{\boldsymbol{v}}_{m+1} - \alpha_{m}\boldsymbol{v}_{m}$
if $m > 1$
 $\tilde{\boldsymbol{v}}_{m+1} = \tilde{\boldsymbol{v}}_{m+1} - \beta_{m-1}\boldsymbol{v}_{m-1}$
 $\tilde{\boldsymbol{w}}_{m+1} = \tilde{\boldsymbol{w}}_{m+1} - \gamma_{m-1}\boldsymbol{w}_{m-1}$
 $\gamma_{m} = \|\tilde{\boldsymbol{v}}_{m+1}\|$
 $\boldsymbol{v}_{m+1} = \tilde{\boldsymbol{v}}_{m+1}/\gamma_{m}$
 $\beta_{m} = \langle \boldsymbol{v}_{m+1} \mid \tilde{\boldsymbol{w}}_{m+1} \rangle$
 $\boldsymbol{w}_{m+1} = \tilde{\boldsymbol{w}}_{m+1}/\beta_{m}$

Algorithm 1.3: The Bi-Lanczos algorithm without look-ahead.

1.3 Convergence Properties

The biggest part of convergence theory and error estimates is for Lanczos procedures used as eigenvalue solvers. Many results are collected under the name Kaniel-Paige-theory [66, 93–95, 142] concerning the relations between the eigenvalues of T and A

as well as convergence of the Ritz values (compare Theorem 2.1).

Many of these results are applicable to Lanczos procedures used to solve linear systems and thus we have a good knowledge about convergence at least for symmetric systems [51, 62, 128, 132], see Section 1.3.1.

Unfortunately, the situation becomes much less clear for nonsymmetric systems, because the proofs in the Lanczos theory are principally based on the symmetry of A. However, there are some error estimates but they are neither as sharp as in the symmetric case nor practically useful at all [5, 47, 50, 64, 129, 141], see Section 1.3.2.

Strakos shows that, practically, the behavior of symmetric and nonsymmetric Krylov solvers is very similar. Unsatisfyingly, up to now nobody managed to prove this [130].

1.3.1 Symmetric Case

Convergence rates and also the quality of the iterated solutions of iterative linear system solvers depend strongly on the *good nature* of the system matrix A. It can be observed that system matrices, close to the identity, are easier to solve. Closeness to I in this sense could be expressed, for example by

- A = I + B with rank(B) is small (small rank perturbation) or
- $\operatorname{cond}(\boldsymbol{A}) \approx 1.$

Theorem 1.2 [17] If A = I + B is an *n* by *n* matrix and rank(B) = m, then the Lanczos algorithm terminates after at most m + 1 steps.

Proof: The dimension of

$$\mathcal{K}_k(oldsymbol{A},oldsymbol{r}_0) = ext{span}\{oldsymbol{r}_0,oldsymbol{A}oldsymbol{r}_0,\dots,oldsymbol{A}^{k-1}oldsymbol{r}_0\} = ext{span}\{oldsymbol{r}_0,oldsymbol{B}oldsymbol{r}_0,\dots,oldsymbol{B}^{\min\{m,k-1\}}oldsymbol{r}_0\}$$

cannot exceed m + 1. Therefore at least $\mathcal{K}_m(\boldsymbol{A}, \boldsymbol{r}_0)$ is an \boldsymbol{A} -invariant subspace of \mathbb{R}^n and thus $\boldsymbol{A}^{-1}\boldsymbol{b} \in \mathcal{K}_m(\boldsymbol{A}, \boldsymbol{r}_0)$ (compare Section 1.2.2).

An error bound of a different manner can be obtained in terms of the **A**-norm $(||\boldsymbol{z}||_{\boldsymbol{A}} = \sqrt{\langle \boldsymbol{z} | \boldsymbol{A} \boldsymbol{z} \rangle})$. This norm is well defined if **A** is s.p.d. Therefore, the following theorem is restricted to the CG algorithm (see Section 1.5.2).

Theorem 1.3 Suppose A is an n by n s.p.d. matrix and b is an n vector. Then for the CG-iterates x_k there holds

$$\|oldsymbol{x} - oldsymbol{x}_k\|_{oldsymbol{A}} \leq 2\sqrt{\kappa} \left(rac{\sqrt{\kappa}-1}{\sqrt{\kappa}+1}
ight)^k \|oldsymbol{x} - oldsymbol{x}_0\|_{oldsymbol{A}}$$

for any $\boldsymbol{x} \in \mathbb{R}^n$ or

$$\|oldsymbol{x}_k-oldsymbol{x}^*\|_2\leq 2\sqrt{\kappa}\left(rac{\sqrt{\kappa}-1}{\sqrt{\kappa}+1}
ight)^k\|oldsymbol{x}_0-oldsymbol{x}^*\|_2,$$

where $\kappa = \operatorname{cond}(A)$.

See [83] for a proof. This means, the nearer κ is to one, the faster the error will decrease.

1.3.2 Nonsymmetric Case

If A is a low rank perturbation of the identity then A^T has obviously the same property (since B is square, it has equal column and row rank). Thus, Theorem 1.2 is also applicable to nonsymmetric systems.

Theorem 1.4 If $\mathbf{A} = \mathbf{I} + \mathbf{B}$ is an *n* by *n* matrix and rank $(\mathbf{B}) = m$, then the Arnoldi and Bi-Lanczos algorithm terminate after at most m + 1 steps.

Proof: The dimensions of

$$egin{aligned} \mathcal{K}_k(oldsymbol{A},oldsymbol{r}_0) &= ext{span}\{oldsymbol{r}_0,oldsymbol{A}oldsymbol{r}_0,oldsymbol{A}oldsymbol{r}_0,\dots,oldsymbol{A}^{\min\{m,k-1\}}oldsymbol{r}_0\} \ &= ext{span}\{oldsymbol{r}_0,oldsymbol{B}oldsymbol{r}_0,\dots,oldsymbol{B}^{\min\{m,k-1\}}oldsymbol{r}_0\} \end{aligned}$$

and

$$egin{aligned} \mathcal{K}_k(oldsymbol{A}^T,oldsymbol{r}_0)&= ext{span}\{oldsymbol{r}_0,oldsymbol{A}^Toldsymbol{r}_0,\ldots,(oldsymbol{A}^{k-1})^Toldsymbol{r}_0\}\ &= ext{span}\{oldsymbol{r}_0,oldsymbol{B}^Toldsymbol{r}_0,\ldots,(oldsymbol{B}^{\min\{m,k-1\}})^Toldsymbol{r}_0\} \end{aligned}$$

cannot exceed m + 1. Therefore, at least $\mathcal{K}_m(\boldsymbol{A}, \boldsymbol{r}_0)$ or $\mathcal{K}_m(\boldsymbol{A}^T, \boldsymbol{r}_0)$ is an \boldsymbol{A} -invariant (respectively \boldsymbol{A}^T -invariant) subspace of \mathbb{R}^n and therefore $\boldsymbol{A}^{-1}\boldsymbol{b}$ is either in $\mathcal{K}_m(\boldsymbol{A}, \boldsymbol{r}_0)$ or in $\mathcal{K}_m(\boldsymbol{A}^T, \boldsymbol{r}_0)$ (compare Sections 1.2.1 and 1.2.3).

To illustrate the difficulties with error bounds for nonsymmetric matrices we present some results for GMRES (see Section 1.5.5). The 2-norm of the kth GMRES-residual \mathbf{r}_k satisfies

$$\|\boldsymbol{r}_{k}\|_{2} = \min_{\substack{\phi_{k} \in \mathcal{P}_{k} \\ \phi_{k}(0)=1}} \|\phi_{k}(\boldsymbol{A})\boldsymbol{r}_{0}\|_{2}$$
(1.16)

where \mathcal{P}_k is set set of polynomials of degree k or less [64]. Suppose \boldsymbol{A} is diagonalizable then there exists an eigen-decomposition $\boldsymbol{A} = \boldsymbol{S} \boldsymbol{\Lambda} \boldsymbol{S}^{-1}$ where $\boldsymbol{\Lambda} = \text{diag}(\lambda_1, \ldots, \lambda_n)$ and the columns of \boldsymbol{S} are the eigenvectors of \boldsymbol{A} . From (1.16) we obtain

$$egin{aligned} \|oldsymbol{r}_k\|_2 &= \min_{\substack{\phi_k\in\mathcal{P}_k\ \phi_k(0)=1}} \|oldsymbol{S}\phi_k(oldsymbol{\Lambda})oldsymbol{S}^{-1}oldsymbol{r}_0\|_2 &\leq ext{cond}_2(oldsymbol{S}) \min_{\substack{\phi_k\in\mathcal{P}_k\ \phi_k(0)=1}} \|oldsymbol{\phi}_k(oldsymbol{\Lambda})\|_2 \|oldsymbol{r}_0\|_2 \ &\Leftrightarrow \quad rac{\|oldsymbol{r}_k\|_2}{\|oldsymbol{r}_0\|_2} &\leq ext{cond}_2(oldsymbol{S}) \min_{\substack{\phi_k\in\mathcal{P}_k\ \phi_k(0)=1}} \left\{ \min_{i=1}^n |\phi_k(\lambda_i)|
ight\} \end{aligned}$$

If A is non-normal, then S does not need not to be unitary and thus $\operatorname{cond}_2(S) > 1$. Consequently, convergence of GMRES, or at least this bound of the residual norm does not solely depend on the eigenvalues of A. Additionally, it can be shown that

Theorem 1.5 Given a non-increasing positive sequence $r_0 \ge r_1 \ge \ldots \ge r_{n-1} > 0$ and an arbitrary set of nonzero complex numbers $\{\lambda_1, \ldots, \lambda_n\}$, there exists a matrix \boldsymbol{A} with eigenvalues $\lambda_1, \ldots, \lambda_n$ and an initial residual \boldsymbol{r}_0 with $\|\boldsymbol{r}_0\|_2 = r_0$ such that the residual vectors \boldsymbol{r}_k at each step of the GMRES method applied to \boldsymbol{A} and \boldsymbol{r}_0 satisfy $\|\boldsymbol{r}_k\|_2 = r_k$ for $k = 1, 2, \ldots, n-1$. See [49] for a proof.

The situation becomes even more difficult, if we have to take rounding errors into consideration. However, the aim of this work is not to fix the arithmetic and then try to get the best result, but it is to fix our requirements on accuracy and then to choose an appropriate arithmetic that enables us to reach this needed accuracy.

1.4 Preconditioners

The idea of preconditioning, i.e., *pre*liminary reduction of the *condition* number, simply is to replace the original linear system Ax = b by a modified linear system $\tilde{Ax} = \tilde{b}$, where this second system has to fulfill two properties [7,8]:

- Solving $\tilde{A}x = \tilde{b}$ should be (more) easy and
- $\tilde{A}x = \tilde{b}$ and Ax = b must have the same solution, i.e., $\tilde{A}^{-1}\tilde{b} = A^{-1}b$.

Here, we demonstrate preconditioning in context of the simplest algorithm for generating Krylov subspaces — the Lanczos algorithm. Since this algorithm only works for symmetric systems, we have to retain the symmetry in \tilde{A} . Therefore we make the ansatz $\tilde{A} = L^{-1}AL^{-T}$ with a nonsingular matrix L, where $M := LL^{T}$ shall in some sense be near to A. To ensure the equivalence of the preconditioned and non-preconditioned system, we have to define $\tilde{b} = L^{-1}b$ and $\tilde{x} = L^{T}x$ to get

$$egin{array}{lll} ilde{A} ilde{x} &= ilde{b} \ \Leftrightarrow & oldsymbol{L}^{-1}oldsymbol{A}oldsymbol{L}^{-T}oldsymbol{L}^Toldsymbol{x} &= oldsymbol{L}^{-1}oldsymbol{b} \ \Leftrightarrow & oldsymbol{A}oldsymbol{x} &= oldsymbol{b}. \end{array}$$

Simply replacing A by $\tilde{A} = L^{-1}AL^{-T}$ and b by $\tilde{b} = L^{-1}b$ and for technical reasons also renaming v by z in the Lanczos algorithm (compare Algorithm 1.2) leads to

1 Given
$$\boldsymbol{x}_{0}, \, \tilde{\boldsymbol{x}}_{0} = \boldsymbol{L}^{T}\boldsymbol{x}_{0}$$

2 $\tilde{\boldsymbol{z}}_{1} = \boldsymbol{L}^{-1}\boldsymbol{b} - \boldsymbol{L}^{-1}\boldsymbol{A}\boldsymbol{L}^{-T}\boldsymbol{L}^{T}\boldsymbol{x}_{0} = \boldsymbol{L}^{-1}(\boldsymbol{b} - \boldsymbol{A}\boldsymbol{x}_{0})$
3 for $m = 1, 2, ...$
4 $\beta_{m-1} = \|\tilde{\boldsymbol{z}}_{m}\|$
5 $\boldsymbol{z}_{m} = \tilde{\boldsymbol{z}}_{m}/\beta_{m-1}$
6 $\tilde{\boldsymbol{z}}_{m+1} = \boldsymbol{L}^{-1}\boldsymbol{A}\boldsymbol{L}^{-T}\boldsymbol{z}_{m}$
7 if $m > 1$
8 $\tilde{\boldsymbol{z}}_{m+1} = \tilde{\boldsymbol{z}}_{m+1} - \beta_{m-1}\boldsymbol{z}_{m-1}$
9 $\alpha_{m} = \langle \tilde{\boldsymbol{z}}_{m+1} \mid \boldsymbol{z}_{m} \rangle$
10 $\tilde{\boldsymbol{z}}_{m+1} = \tilde{\boldsymbol{z}}_{m+1} - \alpha_{m}\boldsymbol{z}_{m}$

From line 2 and 5 we see, comparing with Algorithm 1.2,

$$\boldsymbol{v}_m = \boldsymbol{L} \boldsymbol{z}_m, \quad ext{and} \quad ilde{\boldsymbol{v}}_m = \boldsymbol{L} ilde{\boldsymbol{z}}_m.$$

To avoid the explicit use of L^{-1} and L^{-T} we introduce two auxiliary vectors \boldsymbol{p}_m and $\tilde{\boldsymbol{p}}_m$, defined by

$$\boldsymbol{p}_m = \boldsymbol{M}^{-1} \boldsymbol{v}_m = \boldsymbol{L}^{-T} \boldsymbol{L}^{-1} \boldsymbol{v}_m, \text{ and } \tilde{\boldsymbol{p}}_m = \boldsymbol{M}^{-1} \tilde{\boldsymbol{v}}_m.$$

Now we substitute the vectors \boldsymbol{z}_m and $\tilde{\boldsymbol{z}}_m$ by expressions with \boldsymbol{v}_m , $\tilde{\boldsymbol{v}}_m$, \boldsymbol{p}_m , and $\tilde{\boldsymbol{p}}_m$. With line 4 we get

$$egin{aligned} eta_{m-1} &= \|m{ ilde{z}}_m\| = \langle m{L}^{-1}m{ ilde{v}}_m \mid m{L}^{-1}m{ ilde{v}}_m
angle^{1/2} \ &= \langlem{ ilde{v}}_m \mid m{L}^{-T}m{L}^{-1}m{ ilde{v}}_m
angle^{1/2} \ &= \langlem{ ilde{v}}_m \mid m{ ilde{p}}_m
angle^{1/2}. \end{aligned}$$

Using $\tilde{\boldsymbol{z}}_{m+1} = \boldsymbol{L}^{-1} \tilde{\boldsymbol{v}}_{m+1}$ and $\boldsymbol{L}^{-T} \boldsymbol{z}_m = \boldsymbol{p}_m$ we get in line 6

$$ilde{oldsymbol{z}}_{m+1} = oldsymbol{L}^{-1}oldsymbol{A}oldsymbol{p}_m \quad \Leftrightarrow \quad oldsymbol{L} ilde{oldsymbol{z}}_{m+1} = oldsymbol{A}oldsymbol{p}_m \quad \Leftrightarrow \quad oldsymbol{ ilde{oldsymbol{v}}}_{m+1} = oldsymbol{A}oldsymbol{p}_m \cdot$$

Multiplying lines 8 and 10 with \boldsymbol{L} from left yields $\tilde{\boldsymbol{v}}_{m+1} = \tilde{\boldsymbol{v}}_{m+1} - \beta_{m-1}\boldsymbol{v}_{m-1}$ and $\tilde{\boldsymbol{v}}_{m+1} = \tilde{\boldsymbol{v}}_{m+1} - \alpha_m \boldsymbol{v}_m$. With line 9 we have

$$\alpha_m = \langle \tilde{\boldsymbol{z}}_{m+1} \mid \boldsymbol{z}_m \rangle = \langle \boldsymbol{L}^{-1} \tilde{\boldsymbol{v}}_{m+1} \mid \boldsymbol{L}^T \boldsymbol{p}_m \rangle = \langle \tilde{\boldsymbol{v}}_{m+1} \mid \boldsymbol{p}_m \rangle.$$

We collect our results in Algorithm 1.4 (left)

Given
$$\boldsymbol{x}_0$$
Given \boldsymbol{x}_0 Given \boldsymbol{x}_0 $\tilde{\boldsymbol{v}}_1 = \boldsymbol{b} - \boldsymbol{A} \boldsymbol{x}_0$ $\tilde{\boldsymbol{v}}_1 = \boldsymbol{b} - \boldsymbol{A} \boldsymbol{x}_0$ for $m = 1, 2, \ldots$ for $m = 1, 2, \ldots$ $\tilde{\boldsymbol{p}}_m = \boldsymbol{M}^{-1} \tilde{\boldsymbol{v}}_m$ $\boldsymbol{\delta}_{m-1} = || \tilde{\boldsymbol{v}}_m ||$ $\boldsymbol{\omega}_m = \tilde{\boldsymbol{v}}_m / \beta_{m-1}$ $\boldsymbol{\omega}_m = \tilde{\boldsymbol{v}}_m / \beta_{m-1}$ $\boldsymbol{v}_m = \tilde{\boldsymbol{p}}_m / \beta_{m-1}$ $\boldsymbol{v}_{m+1} = \boldsymbol{A} \boldsymbol{p}_m$ $\tilde{\boldsymbol{v}}_{m+1} = \boldsymbol{A} \boldsymbol{p}_m$ $\tilde{\boldsymbol{v}}_{m+1} = \boldsymbol{A} \boldsymbol{v}_m$ if $m > 1$ $\tilde{\boldsymbol{v}}_{m+1} = \tilde{\boldsymbol{v}}_{m+1} - \beta_{m-1} \boldsymbol{v}_{m-1}$ $\boldsymbol{\omega}_m = \langle \tilde{\boldsymbol{v}}_{m+1} \mid \boldsymbol{p}_m \rangle$ $\boldsymbol{\omega}_{m+1} = \tilde{\boldsymbol{v}}_{m+1} - \beta_{m-1} \boldsymbol{v}_{m-1}$ $\boldsymbol{v}_{m+1} = \tilde{\boldsymbol{v}}_{m+1} - \alpha_m \boldsymbol{v}_m$ $\tilde{\boldsymbol{v}}_{m+1} = \tilde{\boldsymbol{v}}_{m+1} - \alpha_m \boldsymbol{v}_m$

Algorithm 1.4: A (symmetrically) preconditioned Lanczos algorithm with preconditioner $M := LL^T$ (left) and its non-preconditioned variant (right).

As we can see from Algorithm 1.4, the essential modification is the computation of the solution of $M\tilde{p}_m = \tilde{v}_m$ with a matrix M similar to A. In other words, we need an approximate solution of $A\tilde{p}_m = \tilde{v}_m$. At a first glance it seems to make no sense to solve Ax = b by repeatedly solving $A\tilde{p}_m = \tilde{v}_m$ but the emphasis is on *approximate solution*, that is, we only need a fast approximation even though it is a rough one.

In this sense, every linear system solver can be applied as a preconditioner. Due to the nature of these solvers, preconditioners can be divided roughly into three categories:

• Preconditioners based on simple iterative solvers, e.g., Jacobi, Gauß-Seidel, or SOR preconditioners.

- Preconditioners based on direct solvers, modified for fast but approximative solving, e.g., incomplete Cholesky or incomplete LU (ILU) or modified variants [11].
- Problem specific preconditioners, either designed for a broad class of underlying problems or even for one specific matrix or problem. For example, there are preconditioners for elliptic PDE's, namely multigrid or domain decomposition preconditioners [26], or special preconditioners as the diffusion synthetic acceleration preconditioner (DSA), solely designed for the transport equation (see [88]).

In the following two sections, we give an introductory overview about the first two categories. Since this work focuses on generic linear system solving, we won't consider the special preconditioners, described in the last category.

1.4.1 Splitting Techniques

Historically, the first class of iterative solvers for linear systems of equations was based on so called splitting techniques [55, 127, 135]. There we split the matrix \boldsymbol{A} in a sum of two matrices, say \boldsymbol{B} and $\boldsymbol{A}-\boldsymbol{B}$ and then write $\boldsymbol{B}\boldsymbol{x} = (\boldsymbol{B}-\boldsymbol{A})\boldsymbol{x}+\boldsymbol{b}$ instead of $\boldsymbol{A}\boldsymbol{x} = \boldsymbol{b}$. If \boldsymbol{B} is nonsingular, we obtain the following fixed point formulation

$$x = B^{-1}(B - A)x + B^{-1}b$$

Substituting the left hand side \boldsymbol{x} by \boldsymbol{x}_{k+1} and the right hand side \boldsymbol{x} by \boldsymbol{x}_k , we get the iteration scheme

$$x_{k+1} = B^{-1}(B - A)x_k + B^{-1}b = x_k - B^{-1}Ax_k + B^{-1}b$$
 (1.17)

which is convergent if and only if the spectral radius of $B^{-1}(B-A)$ is less than one. In the case of convergence, i.e., with $x^* = \lim_{k\to\infty} (x_k)$ we have $B^{-1}Ax^* + B^{-1}b = 0$ or $Ax^* = b$.

To utilize the iteration scheme (1.17) as a preconditioner, we only perform one iteration step. The main effort is solving the linear system Bz = (B - A)x for z, that is, we should chose B to assure that this solution is easily computable. According to different choices of B we have different algorithms.

• Jacobi Preconditioners

For the Jacobi algorithm, we choose $\boldsymbol{B} = \text{diag}(\boldsymbol{A})$ (see Figure 1.3). This leads to the simple preconditioner

$$z_{i} = x_{i} - \frac{1}{a_{i,i}} \sum_{j=1}^{n} a_{i,j} x_{j} + \frac{b_{i}}{a_{i,i}}$$
$$= \frac{1}{a_{i,i}} \left(b_{i} - \sum_{\substack{j=1\\j\neq i}}^{n} a_{i,j} x_{j} \right), \text{ for } i = 1, \dots, n.$$

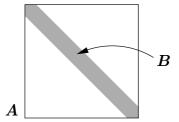


Figure 1.3: Splitting scheme for the Jacobi iteration. B = diag(A).

• Gauß-Seidel Preconditioners

Here we set $\boldsymbol{B} = \text{lowerTriangle}(\boldsymbol{A})$ (see Figure 1.4), This leads to the Gauß-Seidel algorithm which is related to the solution of a triangular system

$$z_{i} = x_{i} - \frac{1}{a_{i,i}} \left(\sum_{j=1}^{i-1} a_{i,j} z_{j} + \sum_{j=i}^{n} a_{i,j} x_{j} \right) + \frac{b_{i}}{a_{i,i}}$$
$$= \frac{1}{a_{i,i}} \left(b_{i} - \sum_{j=1}^{i-1} a_{i,j} z_{j} - \sum_{j=i+1}^{n} a_{i,j} x_{j} \right), \quad \text{for } i = 1, \dots, n.$$

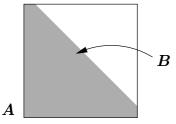


Figure 1.4: Splitting scheme for the Gauß-Seidel iteration. B = lowerTriangle(A).

• Relaxation Methods

For both, the Jacobi and the Gauß-Seidel algorithm, one can scale the matrix \boldsymbol{B} by a so called relaxation parameter ω . This leads to the Jacobi relaxation preconditioner

$$z_i = x_i - \frac{\omega}{a_{i,i}} \sum_{j=1}^n a_{i,j} x_j + \omega \frac{b_i}{a_{i,i}}$$
$$= (1-\omega)x_i - \frac{\omega}{a_{i,i}} \left(b_i - \sum_{\substack{j=1\\j\neq i}}^n a_{i,j} x_j \right), \quad \text{for } i = 1, \dots, n.$$

or the Gauß-Seidel relaxation preconditioner, also known as SOR (successive over relaxation) preconditioner

$$z_{i} = x_{i} - \frac{\omega}{a_{i,i}} \left(\sum_{j=1}^{i-1} a_{i,j} z_{j} + \sum_{j=i}^{n} a_{i,j} x_{j} \right) + \omega \frac{b_{i}}{a_{i,i}}$$
$$= (1 - \omega) x_{i} - \frac{\omega}{a_{i,i}} \left(\sum_{j=1}^{i-1} a_{i,j} z_{j} + \sum_{j=i+1}^{n} a_{i,j} x_{j} \right), \quad \text{for } i = 1, \dots, n.$$

These algorithms can be interpreted as a weighted sum of the non-preconditioned vector \boldsymbol{x} and the *update* vector $\boldsymbol{z} - \boldsymbol{x}$. To ensure positive weights we must choose $\omega \in (0, 2)$. A good relaxation parameter can improve the preconditioner significantly but in general it is hard to determine an optimal relaxation parameter ω . However, there are several works (see e.g. [2, 135]) dealing with this question depending on special properties of the matrix \boldsymbol{A} .

1.4.2 Incomplete Decompositions

This important class of generic preconditioners is based on direct solvers, i.e., on a multiplicative decomposition, say $\boldsymbol{L} \cdot \boldsymbol{U}$ of \boldsymbol{A} . Favorably, we deal with triangular factors \boldsymbol{L} and \boldsymbol{U} as delivered, for example, by the LU-decomposition. Since solving $\boldsymbol{L}\boldsymbol{U}\boldsymbol{z} = \boldsymbol{x}$ for \boldsymbol{z} is generally neither really fast nor very approximative, we actually do not compute the entire factors \boldsymbol{L} and \boldsymbol{U} but only incomplete factors $\tilde{\boldsymbol{L}}$ and $\tilde{\boldsymbol{U}}$. That means, we compute only a subset of the elements of these triangular matrices. There are basically two strategies how to decide, whether an element of the complete triangular factor is to be taken up into the sparse triangular factor or whether it can be dropped:

- Compute an element $l_{i,j}$ of $\tilde{\boldsymbol{L}}$ (or $u_{i,j}$ of $\tilde{\boldsymbol{U}}$, respectively) depending on memory management considerations. Usually we compute an element at place (i, j) if and only if $a_{i,j} \neq 0$. In this case we can use the storage scheme of \boldsymbol{A} to store $\tilde{\boldsymbol{L}}$ and $\tilde{\boldsymbol{U}}$ in.
- Compute an element $l_{i,j}$ of \tilde{L} (or $u_{i,j}$ of \tilde{U} , respectively) depending on its *importance*. Usually this importance is measured in the following sense. The sparse triangular factors are computed columnwise. An element is dropped if it is smaller than a given *drop-tolerance* times the norm of the corresponding column of A.

There exist various modifications [11], one, for example, tries to save some of the dropped information by adding the dropped elements to the diagonal element of the upper diagonal factor to retain the column-norms of A.

The LU-decomposition works for arbitrary nonsingular matrices A. However, there are some variants exploiting special properties of A such as symmetry or positive definiteness. Since we make extensive use of these preconditioners (see Section 4.3.3, we describe some important LU variants here.

• LDM^T Decomposition of General Matrices

For this variant, we actually do not need a special structure in A. In the usual LU factorization L tends to be well-conditioned whereas the condition number of A moves into U [41]. Here we factorize A into a three-matrices product LDM^T where D is diagonal and L and M both are unit lower triangular. Subsequently we distribute D among L and M, i.e., we define

$$oldsymbol{D}_1 := \sqrt{|oldsymbol{D}|}, \quad oldsymbol{D}_2 := \mathrm{sign}(oldsymbol{D})\sqrt{|oldsymbol{D}|}, \quad oldsymbol{\hat{L}} := oldsymbol{L}oldsymbol{D}_1, \quad oldsymbol{\hat{U}} := oldsymbol{M}oldsymbol{D}_2.$$

This leads to a $\hat{\boldsymbol{L}}\hat{\boldsymbol{U}}^T$ factorization and heuristically, $\hat{\boldsymbol{L}}$ and $\hat{\boldsymbol{U}}$ have a more or less equal condition number $\sqrt{\text{cond}(\boldsymbol{A})}$. This modification is important for verifying an approximate solution of a linear system as described in Section 4.3.3. There we need the smallest singular value of \boldsymbol{A} which we estimate by $\sigma_{\min}(\hat{\boldsymbol{L}}) \cdot \sigma_{\min}(\hat{\boldsymbol{U}})$. Since these singular values are the square root of the eigenvalues of $\hat{\boldsymbol{L}}\hat{\boldsymbol{L}}^T$ and $\hat{\boldsymbol{U}}\hat{\boldsymbol{U}}^T$ we have to compute eigenvalues of matrices with condition numbers $\operatorname{cond}(\hat{\boldsymbol{L}})^2$ respectively $\operatorname{cond}(\hat{\boldsymbol{U}})^2$. Without the balancing, i.e. with $\operatorname{cond}(\boldsymbol{U}) \approx \operatorname{cond}(\boldsymbol{A})$, this would limit us to linear systems with moderate condition numbers less than $\epsilon^{-1/2}$.

To obtain sparse or incomplete triangular factors, obviously all modifications described for the LU-factorization can be applied.

• LDL^T Decomposition of Symmetric Matrices

Here we suppose A to be symmetric, then we have redundancy in the LDM^{T} algorithm since in this case L = M. This can be seen by multiplying $A = LDM^{T}$ with M^{-1} from left and M^{-T} from right. This yields

$$M^{-1}AM^{-T} = M^{-1}LDM^{T}M^{-T} = M^{-1}LD.$$

The left hand side is symmetric and the right hand side is lower triangular and thus $M^{-1}LD$ is diagonal. Since D is nonsingular, this implies $M^{-1}L$ is also diagonal. But $M^{-1}L$ is unit lower triangular and so $M^{-1}L = I$ (see [41]). Thus we can omit computing M. Again distributing D yields

$$oldsymbol{D}_1 := \sqrt{|oldsymbol{D}|}, \quad oldsymbol{D}_2 := \mathrm{sign}(oldsymbol{D})\sqrt{|oldsymbol{D}|}, \quad oldsymbol{\hat{L}} := oldsymbol{L}oldsymbol{D}_1, \quad oldsymbol{\hat{U}} := oldsymbol{L}oldsymbol{D}_2.$$

In this symmetric case we even have

$$\operatorname{cond}(\hat{\boldsymbol{L}}) = \operatorname{cond}(\boldsymbol{L}|\boldsymbol{D}|^{1/2}) = \operatorname{cond}(\boldsymbol{L}|\boldsymbol{D}|^{1/2}\operatorname{sign}(\boldsymbol{D})) = \operatorname{cond}(\hat{\boldsymbol{U}}).$$

• Cholesky or $\boldsymbol{L}\boldsymbol{L}^T$ Decomposition of S.P.D. Matrices

Moreover, if \boldsymbol{A} is symmetric positive definite (s.p.d.), i.e., if $\boldsymbol{x}^T \boldsymbol{A} \boldsymbol{x} > 0$ for all $\boldsymbol{x} \neq \boldsymbol{o}$ then we have

$$0 < (\boldsymbol{L}^{-T}\boldsymbol{e}_i)^T \boldsymbol{A}(\boldsymbol{L}^{-T}\boldsymbol{e}_i) = \boldsymbol{e}_i^T \boldsymbol{L}^{-1} \boldsymbol{A} \boldsymbol{L}^{-T} \boldsymbol{e}_i = \boldsymbol{e}_i^T \boldsymbol{D} \boldsymbol{e}_i = d_{i,i}.$$

Thus all elements of D are positive which enables us to define

$$\hat{D} := \sqrt{D}, \quad \hat{L} := LD.$$

Together we get $\boldsymbol{A} = \hat{\boldsymbol{L}} \hat{\boldsymbol{L}}^{T}$, that is $\hat{\boldsymbol{L}}$ is the Cholesky factor of \boldsymbol{A} .

• Pivoting and Reordering Algorithms

Suppose $\boldsymbol{L}\boldsymbol{U}^T$ is a triangular factorization of \boldsymbol{A} , computed in finite precision. Then we have $\boldsymbol{L}\boldsymbol{U}^T = \tilde{\boldsymbol{A}} \approx \boldsymbol{A}$. For the error matrix $\tilde{\boldsymbol{A}} - \boldsymbol{A}$ we have (see [104])

$$|\tilde{\boldsymbol{A}} - \boldsymbol{A}| \le 3(n-1)\epsilon \cdot (|\boldsymbol{A}| + |\boldsymbol{L}||\boldsymbol{U}|) + O(\epsilon^2).$$

Consequently, this error might be very large if we encounter a small pivot during factorizing A because this leads to large elements in L and U. We stress that small pivots are not necessarily due to ill-conditioning as the example

$$\boldsymbol{A} = \begin{pmatrix} \epsilon & 1 \\ 1 & 0 \end{pmatrix} = \begin{pmatrix} 1 & 0 \\ 1/\epsilon & 1 \end{pmatrix} \begin{pmatrix} \epsilon & 1 \\ 0 & -1/\epsilon \end{pmatrix} = \boldsymbol{L} \boldsymbol{U}^{T}$$

demonstrates. To avoid large elements in L and U we must allow some kind of pivoting [30], i.e., permuting of the rows of A:

$$\boldsymbol{P}\boldsymbol{A} = \begin{pmatrix} 0 & 1 \\ 1 & 0 \end{pmatrix} \begin{pmatrix} \epsilon & 1 \\ 1 & 0 \end{pmatrix} = \begin{pmatrix} 1 & 0 \\ \epsilon & 1 \end{pmatrix} = \begin{pmatrix} 1 & 0 \\ \epsilon & 1 \end{pmatrix} \begin{pmatrix} 1 & 0 \\ 0 & 1 \end{pmatrix} = \boldsymbol{L}\boldsymbol{U}^{T}.$$

Usually there are two pivoting strategies: column pivoting and complete pivoting. Both limit the norm of \boldsymbol{L} and \boldsymbol{U} , but unfortunately, they destroy a possibly given band structure. In particular we have the following situation. Suppose \boldsymbol{A} to have lower bandwidth p and upper bandwidth q. Without pivoting, the original bandwidths remain unchanged, with column pivoting, \boldsymbol{U} has bandwidth p+q, while \boldsymbol{L} 's band-structure is completely lost, and finally with complete pivoting we loose the structure of \boldsymbol{L} and \boldsymbol{U} . Thus, dealing with sparse matrices, we have to trade off between loosing accuracy and saving memory.

Usually, pivoting is done in each step of an LU factorization. However, there are some useful strategies how to compute permutation matrices \boldsymbol{P} and \boldsymbol{Q} in advance, such that an LU factorization of $\boldsymbol{P}\boldsymbol{A}\boldsymbol{Q}$ has *advantageous* properties [23,31]. Advantageous in this sense means smaller bandwidths or less nonzero elements, for example. One of the most successful algorithms is the so called 'reverse Cuthill-McKee' algorithm. This powerful graph theoretic algorithm often leads to a dramatical reduction of the numbers of nonzeros in \boldsymbol{L} and \boldsymbol{U} and therefore to a large speedup in solving and particularly in verifying a sparse linear system of equations.

1.5 Krylov Type Linear System Solver

In this section we first give an overview about various Krylov type linear system solver [44, 60]. Since our improvements in convergence, speed, and accuracy as well as our verification methods do not depend on the particular method, we only describe some of the most important variants in more detail. These are CG (Section 1.5.2), BiCG (Section 1.5.3), CGS (Section 1.5.4), and GMRES (Section 1.5.5). Additionally, we present the basic ideas of residual norm smoothing (BiCGStab, QMRCGStab) and quasi minimization (QMR, TFQMR), see Section 1.5.6.

For each described method, we give a preconditioned algorithm formulated in a pseudo programming language, taken from [9]

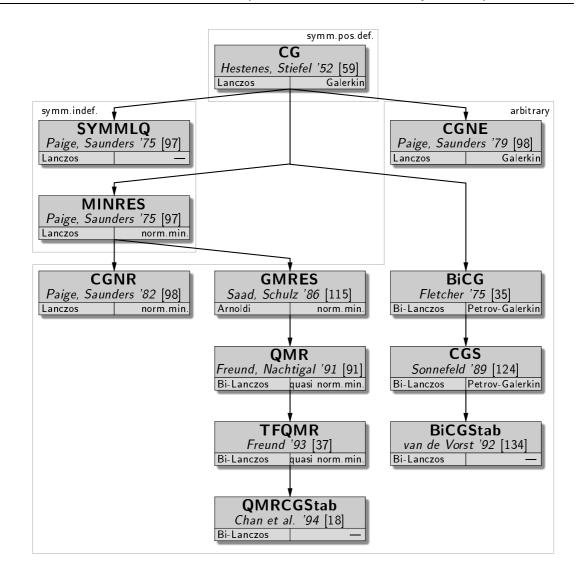


Figure 1.5: Krylov type linear system solver

1.5.1 Overview

The oldest and probably best known Krylov type method is the Conjugate Gradient (CG) method, developed by Hestenes and Stiefel in 1952 [59]. It was designed for solving systems of linear equations with symmetric positive definite coefficient matrices. Possibly because matrix dimensions were small at this time and CG was considered as a direct solver, there was no much attention to this algorithm. This changed in the middle of the 70's, where the iterative character of CG was spotted mainly by Paige and Saunders (see Section 1.5.2).

In 1975, the first remarkable variants of CG were developed: MINRES and SYMMLQ by Paige and Saunders [97] and BiCG by Fletcher [35]. Since CG is based on a Lanczos algorithm for generating orthonormal bases of the Krylov spaces and a subsequent LDL^{T} factorization of the symmetric tridiagonal matrix T, it is potentially unstable if A, and consequently T is indefinite (T is unitarily similar to A), see Section 1.5.2.

MINRES avoids this LDL^{T} factorization switching to the norm **min**imizing condition for the **res**idual vectors (see Definition 1.2). This yields to a least squares problem which does not depend on the definiteness of A [32].

SYMMLQ solves the projected system with system matrix T via an LQ factorization instead of LDL^{T} , but does not minimize anything. However, it keeps the residuals orthogonal to all previous ones.

While these two variants retained the Lanczos process and were therefore bound to symmetric matrices A, Fletcher generalized CG by switching to the Bi-Lanczos algorithm in combination with the Petrov-Galerkin condition. This leads to the **bi-c**onjugate gradient (BiCG) algorithm, which works for arbitrary square (and nonsingular) matrices A. Unfortunately, BiCG needs matrix-vector products with the transposed system matrix, which is often a problem for large sparse matrices (compare Section 6.1.2). Due to this lack, the development of Bi-Lanczos based algorithms stagnated for several years.

The next two variants, again based on CG, avoiding the assumption of symmetry of A. One obvious trick is to apply CG to the **n**ormal **e**quations (CGNE) [98], i.e., to $A^T A x = A^T b$. While the convergence rate of CG depends on the condition number of the system matrix (see Section 1.3) it now depends on the square of cond(A) and thus might be relatively slow.

Several proposals have been made to improve the numerical stability of this method. The best known is by Paige and Saunders [98] and is based upon applying the Lanczos method to the auxiliary 2n by 2n system

$$egin{pmatrix} oldsymbol{I} & oldsymbol{A} \ oldsymbol{A}^T & oldsymbol{0} \end{pmatrix} egin{pmatrix} oldsymbol{r} \ oldsymbol{x} \end{pmatrix} = egin{pmatrix} oldsymbol{b} \ oldsymbol{o} \end{pmatrix} egin{pmatrix} oldsymbol{A} \ oldsymbol{A} \end{pmatrix} egin{pmatrix} oldsymbol{r} \ oldsymbol{r} oldsymbol{r} \ oldsymbol{r} \ oldsymbol{r} \end{pmatrix} egin{pmatrix} oldsymbol{r} \ oldsymbol{r} \$$

A clever execution of this scheme delivers the $\boldsymbol{L}\boldsymbol{D}\boldsymbol{L}^{T}$ factorization of the tridiagonal matrix that would have been computed by carrying out the Lanczos procedure with $\boldsymbol{A}^{T}\boldsymbol{A}$ but without squaring the condition number.

Applying this ideas to MINRES leads to a CG like algorithm applied to the **n**ormal equations and minimizing the **r**esidual norm. The resulting algorithm was called CGNR [98].

Another important extension of the MINRES algorithm, called GMRES (generalized minimal residuals), was developed in 1986 by Saad and Schulz [115]. They avoided the need of symmetry by interchanging the underlying Lanczos algorithm with the Arnoldi algorithm. The disadvantage of this approach is that it needs increasing time and memory with each iteration due to the Arnoldi method which orthogonalizes every new Krylov basis vector against all previous ones. Several proposals have been made to get this mathematically excellent algorithm computationally more attractive. The best known is the restart technique GMRES(m), restarting GMRES every m iterations with the best approximation computed so far as the new starting vector. Beside this computational penalty, however, it is the only Krylov algorithm for arbitrary matrices with a norm minimizing property of the generated residuals (see Section 1.5.5).

Three years later, Sonnefeld [124] improved the meanwhile 14 years old BiCG algorithm to work without access to the transposed of \boldsymbol{A} . Substantially, this improvement was based on replacing scalar products like $\langle \boldsymbol{A}\boldsymbol{p} \mid \boldsymbol{A}^T\boldsymbol{p} \rangle$ with $\langle \boldsymbol{A}^2\boldsymbol{p} \mid \boldsymbol{p} \rangle$ and therefore was called CGS (conjugate gradients squared). This squaring can be

interpreted as performing two *minimization* steps at once while computing only one search direction. Sometimes we can observe a doubled convergence rate but since the second 'minimization' step uses the old (and maybe completely wrong) search direction, we often have a quite irregular convergence behavior (see Section 1.5.4).

In 1992, van de Vorst [134] introduced an additional parameter into the Petrov-Galerkin condition and used this parameter to smooth this irregular behavior of the residual norms in CGS. Due to this **stab**ilizing parameter, he called his algorithm BiCGStab, again reminding on the underlying Bi-Lanczos procedure (see Section 1.5.6).

Since norm minimizing of the residuals for arbitrary matrices depends strongly on the Arnoldi algorithm (GMRES), there seemed to be no possibility to develop a GMRES like algorithm with short recurrences. However, in 1991, Freund and Nachtigal [91] managed to bound the residual norm with a product of two norms, where one of them can be minimized even by using a Bi-Lanczos procedure for generating the needed Krylov spaces and the other can be bound independently of A. Minimizing only the first of these two norms, they introduced a **q**uasi **m**inimization of the **r**esidual norms (QMR) based on short recurrence formulas (see Section 1.5.6).

Similar to the step from BiCG to CGS, Freund [37] improves this algorithm 1993 to work without transposed matrix-vector products. The resulting procedure was called transpose free QMR — TFQMR. Compared with QMR, this algorithm again shows a more irregular behavior in the computed residual norms for the same reason as the CGS algorithm does. Applying the ideas of van de Vorst, Chan *et al.* [18] stabilized TFQMR and developed his so called QMRCGStab algorithm.

1.5.2 Conjugate Gradients (CG)

There are various ways to derive CG. Usually one starts with an obvious steepest descent approach to minimize the function

$$\boldsymbol{\phi}(\boldsymbol{x}) = \frac{1}{2} \boldsymbol{x}^T \boldsymbol{A} \boldsymbol{x} - \boldsymbol{x}^T \boldsymbol{b}$$

Minimizing ϕ is equivalent to finding the zero of its gradient $\nabla \phi(x) = Ax - b$ if A is positive definite (note: for the second derivate of ϕ we have $\nabla^2 \phi \equiv A$).

The resulting algorithm often shows a prohibitively slow convergence rate and heavy oscillating residuals. Modifying this steepest descent algorithm to get conjugate search directions avoids these pitfalls and leads to the Conjugate Gradient algorithm. Using this approach is fairly intuitive and has a good geometrical interpretation. Unfortunately, this interpretation gets lost for most of the other more advanced Krylov algorithms. Therefore we try to give a uniform derivation based on the generating process of the used Krylov spaces (Section 1.2) and based on the conditions to choose the current iterate from the Krylov space (Section 1.1), compare [41]. The main attention thereby is on deriving the important short update formulas.

Suppose A to be positive definite. After m steps of the Lanczos algorithm (compare Section 1.2.2) we obtain the factorization

$$\boldsymbol{A} \boldsymbol{V}_m = \boldsymbol{V}_m \boldsymbol{T}_m + \boldsymbol{v}_{m+1} \cdot \boldsymbol{h}_{m+1,m} \boldsymbol{e}_m^T.$$

Representing $\boldsymbol{x}_m \in \mathcal{V}_m$ as $\boldsymbol{x}_m = \boldsymbol{x}_0 + \boldsymbol{V}_m \boldsymbol{\xi}_m$, the Galerkin condition $\boldsymbol{b} - \boldsymbol{A} \boldsymbol{x}_m \perp \mathcal{V}_m$ now writes as

$$egin{aligned} oldsymbol{V}_m^T(oldsymbol{b}-oldsymbol{A}oldsymbol{x}_m) &= oldsymbol{0} \ &\Leftrightarrow oldsymbol{V}_m^Toldsymbol{A}oldsymbol{V}_moldsymbol{\xi}_m) &= oldsymbol{0} \ &\Leftrightarrow oldsymbol{V}_m^Toldsymbol{A}oldsymbol{V}_moldsymbol{\xi}_m) &= oldsymbol{0} \ &\Leftrightarrow oldsymbol{V}_moldsymbol{\xi}_m &= oldsymbol{V}_m^Toldsymbol{r}_n \ &\Leftrightarrow oldsymbol{T}_moldsymbol{\xi}_m &= oldsymbol{W}_m^Toldsymbol{p}_moldsymbol{a} \ &oldsymbol{V}_m^Toldsymbol{A}oldsymbol{v}_moldsymbol{\xi}_m) = oldsymbol{0} \ &\Leftrightarrow oldsymbol{V}_moldsymbol{\xi}_m &= oldsymbol{V}_m^Toldsymbol{r}_n \ &\Leftrightarrow oldsymbol{T}_moldsymbol{\xi}_m &= oldsymbol{W}_m^Toldsymbol{p}_moldsymbol{a} \ &oldsymbol{T}_moldsymbol{\xi}_m = oldsymbol{V}_m^Toldsymbol{p}_moldsymbol{e} \ &oldsymbol{A}$$

With this approach, computing the *m*th iterated approximation \boldsymbol{x}_m becomes equivalent to solving a positive definite tridiagonal system with system matrix

$$\boldsymbol{T}_{m} = \begin{pmatrix} \alpha_{1} & \beta_{1} & & & \\ \beta_{1} & \alpha_{2} & \beta_{2} & & \\ & \ddots & \ddots & \ddots & \\ & & \beta_{m-2} & \alpha_{m-1} & \beta_{m-1} \\ & & & & \beta_{m-1} & \alpha_{m} \end{pmatrix},$$

which may be solved via the LDL^{T} factorization. In particular, by setting

$$\boldsymbol{L}_{m} = \begin{pmatrix} 1 & 0 & \cdots & \cdots & 0 \\ l_{1} & 1 & & & \vdots \\ 0 & l_{2} & \ddots & & \vdots \\ \vdots & & \ddots & 1 & 0 \\ 0 & \cdots & 0 & l_{m-1} & 1 \end{pmatrix} \quad \text{and} \quad \boldsymbol{D}_{m} = \begin{pmatrix} d_{1} & 0 & \cdots & \cdots & 0 \\ 0 & d_{2} & & & \vdots \\ \vdots & & \ddots & & \vdots \\ \vdots & & \ddots & & \vdots \\ \vdots & & d_{m-1} & 0 \\ 0 & \cdots & \cdots & 0 & d_{m} \end{pmatrix}$$

we find by comparing entries in $\mathbf{T}_m = \mathbf{L}_m \mathbf{D}_m \mathbf{L}_m^T$ that $d_1 = \alpha_1$, $l_{j-1} = \beta_{j-1}/d_{j-1}$, and $d_j = \alpha_j - \beta_{j-1}l_{j-1}$ for $j = 2, \ldots, m$. Note that since the computation of l_{j-1} and d_j depends solely on d_{j-1} , we can update $(\mathbf{L}_m, \mathbf{D}_m)$ from $(\mathbf{L}_{m-1}, \mathbf{D}_{m-1})$ by computing

$$l_{m-1} = \beta_{m-1}/d_{m-1} d_m = \alpha_m - \beta_{m-1}l_{m-1}.$$

Defining $\tilde{\boldsymbol{P}}_m \in {\rm I\!R}^{n \times m}$ and $\boldsymbol{y}_m \in {\rm I\!R}^m$ by the equations

$$\tilde{\boldsymbol{P}}_{m}\boldsymbol{L}_{m}^{T} = \boldsymbol{V}_{m} \text{ and } \boldsymbol{L}_{m}\boldsymbol{D}_{m}\boldsymbol{y}_{m} = \boldsymbol{V}_{m}^{T}\boldsymbol{r}_{0}$$
 (1.18)

we get

$$egin{aligned} oldsymbol{x}_m &= oldsymbol{x}_0 + oldsymbol{V}_m \cdot oldsymbol{T}_m^{-1} oldsymbol{V}_m^T oldsymbol{r}_0 = oldsymbol{x}_0 + oldsymbol{V}_m oldsymbol{L}_m^{-T} \cdot (oldsymbol{L}_m oldsymbol{D}_m)^{-1} \cdot oldsymbol{V}_m^T oldsymbol{r}_0 \ oldsymbol{L}_m oldsymbol{D}_m oldsymbol{y}_m \ &= oldsymbol{x}_0 + oldsymbol{ ilde{P}}_m oldsymbol{y}_m. \end{aligned}$$

Due to the simple structure of L_m and D_m we get the following short update formulas for \tilde{P}_m and y_m :

$$\begin{split} \tilde{\boldsymbol{P}}_{m} &= (\tilde{\boldsymbol{P}}_{m-1} | \tilde{\boldsymbol{p}}_{m}) \quad \text{with} \quad \tilde{\boldsymbol{p}}_{m} = \boldsymbol{v}_{m} - l_{m-1} \tilde{\boldsymbol{p}}_{m-1} \\ \boldsymbol{y}_{m} &= (\boldsymbol{y}_{m-1} | y_{m})^{T} \quad \text{with} \quad y_{m} = (\boldsymbol{v}_{m}^{T} \boldsymbol{r}_{0} - l_{m-1} d_{m-1} y_{m-1})/d_{m-1} \end{split}$$

Given
$$\boldsymbol{x}_{0}$$

 $\boldsymbol{r}_{0} = \boldsymbol{b} - \boldsymbol{A}\boldsymbol{x}_{0}$
 $\beta_{0} = \|\boldsymbol{r}_{0}\|_{2}$
 $\boldsymbol{q}_{0} = \boldsymbol{0}$
for $m = 1, 2, ...$
 $\boldsymbol{v}_{m} = \boldsymbol{r}_{m-1}/\beta_{m-1}$
 $\alpha_{m} = \boldsymbol{v}_{m}^{T}\boldsymbol{A}\boldsymbol{v}_{m}$
 $\boldsymbol{r}_{m} = (\boldsymbol{A} - \alpha_{m}\boldsymbol{I})\boldsymbol{v}_{m} - \beta_{m-1}\boldsymbol{v}_{m-1}$
 $\beta_{m} = \|\boldsymbol{r}_{m}\|_{2}$
if $m = 1$
 $d_{1} = \alpha_{1}$
 $y_{1} = \beta_{0}/\alpha_{1}$
 $\tilde{\boldsymbol{p}}_{1} = \boldsymbol{v}_{1}$
 $\boldsymbol{x}_{1} = y_{1}\boldsymbol{v}_{1}$
else
 $l_{m-1} = \beta_{m-1}/d_{m-1}$
 $d_{m} = \alpha_{m} - \beta_{m-1}l_{m-1}$
 $y_{m} = -l_{m-1}d_{m-1}y_{m-1}/d_{m}$
 $\tilde{\boldsymbol{p}}_{m} = \boldsymbol{v}_{m} - l_{m-1}\tilde{\boldsymbol{p}}_{m-1}$
 $\boldsymbol{x}_{m} = \boldsymbol{x}_{m-1} + y_{m}\tilde{\boldsymbol{p}}_{m}$

Algorithm 1.5: Conjugate Gradient algorithm derived from a Lanczos process with Petrov condition.

and thus

$$\boldsymbol{x}_m = \boldsymbol{x}_0 + \boldsymbol{P}_m \boldsymbol{y}_m = \boldsymbol{x}_0 + \boldsymbol{P}_{m-1} \boldsymbol{y}_{m-1} + \tilde{\boldsymbol{p}}_m y_m = \boldsymbol{x}_{m-1} + \tilde{\boldsymbol{p}}_m y_m.$$

Doing a lot of algebraic substitutions and transformations (compare [41]) one can prove that Algorithm 1.5 is equivalent to the CG Algorithm 1.6 (with M = I). However, we can see that both algorithms require one matrix vector product per iteration (Av_m respectively Ap_m) and update their approximate solution both with a short recurrence.

1.5.3 Bi-Conjugate Gradients (BiCG)

The Conjugate Gradient method is not suitable for nonsymmetric systems because the residual vectors cannot be made orthogonal with short recurrences (for a proof of this see Faber and Manteuffel [33]). The GMRES method (see Section 1.5.5) retains orthogonality of the residuals by using long recurrences, at the cost of a larger storage demand. The Bi-Conjugate Gradient (BiCG) method takes another approach, replacing the orthogonal sequence of residuals by two mutually orthogonal sequences, at the price of no longer providing a minimization.

BiCG is based on a Bi-Lanczos process and a Petrov-Galerkin condition. That is, two sequences of Krylov subspace basis vectors are generated: $V_m = (v_1 | \cdots | v_m)$ and $W_m = (w_1 | \cdots | w_m)$.

Given
$$\boldsymbol{x}_0$$

 $\boldsymbol{r}_0 = \boldsymbol{b} - \boldsymbol{A} \boldsymbol{x}_0$
for $m = 1, 2, ...$
solve $\boldsymbol{M} \boldsymbol{z}_{m-1} = \boldsymbol{r}_{m-1}$
 $\rho_{m-1} = \langle \boldsymbol{r}_{m-1} \mid \boldsymbol{z}_{m-1} \rangle$
if $m = 1$
 $\boldsymbol{p}_1 = \boldsymbol{z}_0$
else
 $\beta_{m-1} = \rho_{m-1} / \rho_{m-2}$
 $\boldsymbol{p}_m = \boldsymbol{z}_{m-1} + \beta_{m-1} \boldsymbol{p}_{m-1}$
 $\boldsymbol{q}_m = \boldsymbol{A} \boldsymbol{p}_m$
 $\alpha_m = \rho_{m-1} / \langle \boldsymbol{p}_m \mid \boldsymbol{q}_m \rangle$
 $\boldsymbol{x}_m = \boldsymbol{x}_{m-1} + \alpha_m \boldsymbol{p}_m$
 $\boldsymbol{r}_m = \boldsymbol{r}_{m-1} - \alpha_m \boldsymbol{q}_m$

Algorithm 1.6: Preconditioned Conjugate Gradient algorithm (CG).

Representing $\boldsymbol{x}_m \in \mathcal{V}_m$ as $\boldsymbol{x}_m = \boldsymbol{x}_0 + \boldsymbol{V}_m \boldsymbol{\xi}_m$, the Petrov-Galerkin condition $\boldsymbol{b} - \boldsymbol{A} \boldsymbol{x}_m \perp \mathcal{W}_m$ now writes as

$$W_m^T(\boldsymbol{b} - \boldsymbol{A}\boldsymbol{x}_m) = W_m^T(\boldsymbol{b} - \boldsymbol{A}\boldsymbol{x}_0 - \boldsymbol{A}\boldsymbol{V}_m\boldsymbol{\xi}_m) = 0$$

$$\Leftrightarrow \quad W_m^T\boldsymbol{A}\boldsymbol{V}_m\boldsymbol{\xi}_m = W_m^T\boldsymbol{r}_0,$$

$$\Leftrightarrow \quad \boldsymbol{T}_m\boldsymbol{\xi}_m = \|\boldsymbol{r}_0\|\boldsymbol{e}_1$$

Contrary to the Lanczos algorithm, the matrix

$$\boldsymbol{T}_{m} = \boldsymbol{W}_{m}^{T} \boldsymbol{A} \boldsymbol{V}_{m} = \begin{pmatrix} \alpha_{1} & \beta_{1} & & & \\ \gamma_{1} & \alpha_{2} & \beta_{2} & & \\ & \ddots & \ddots & \ddots & \\ & & \gamma_{m-2} & \alpha_{m-1} & \beta_{m-1} \\ & & & \gamma_{m-1} & \alpha_{m} \end{pmatrix}$$

is no more symmetric but still tridiagonal. Thus, instead of LDL^{T} we have to apply a LU factorization to solve this tridiagonal system:

$$\boldsymbol{L}_{m} = \begin{pmatrix} 1 & 0 & \cdots & \cdots & 0 \\ l_{1} & 1 & & & \vdots \\ 0 & l_{2} & \ddots & & \vdots \\ \vdots & & \ddots & 1 & 0 \\ 0 & \cdots & 0 & l_{m-1} & 1 \end{pmatrix} \quad \text{and} \quad \boldsymbol{U}_{m} = \begin{pmatrix} u_{1,1} & u_{1,2} & 0 & \cdots & 0 \\ 0 & u_{2,2} & u_{2,3} & & \vdots \\ \vdots & & \ddots & \ddots & 0 \\ \vdots & & & u_{m-1,m-1} & u_{m-1,m} \\ 0 & \cdots & & 0 & u_{m,m} \end{pmatrix}$$

Similar to CG, we have simple update formulas for \boldsymbol{L}_m and \boldsymbol{U}_m . With $\tilde{\boldsymbol{P}}_m \in \mathbb{R}^{n \times m}$ and $\boldsymbol{y}_m \in \mathbb{R}^m$ defined by

$$\tilde{\boldsymbol{P}}_{m}\boldsymbol{U}_{m}=\boldsymbol{V}_{m} \quad ext{and} \quad \boldsymbol{L}_{m}\boldsymbol{y}_{m}=\boldsymbol{V}_{m}^{T}\boldsymbol{r}_{0}$$

$$(1.19)$$

we get

$$egin{aligned} oldsymbol{x}_m &= oldsymbol{x}_0 + oldsymbol{V}_m \cdot oldsymbol{T}_m^{-1} oldsymbol{V}_m^T oldsymbol{r}_0 &= oldsymbol{x}_0 + oldsymbol{V}_m oldsymbol{U}_m^{-1} \cdot oldsymbol{L}_m^{-1} \cdot oldsymbol{V}_m^T oldsymbol{r}_0 \ &= oldsymbol{x}_0 + oldsymbol{ ilde{P}}_m oldsymbol{y}_m. \end{aligned}$$

Due to the particular structure of L_m and U_m we get the following short update formulas for \tilde{P}_m and y_m :

$$egin{aligned} & egin{split} & e$$

and thus

$$oldsymbol{x}_m = oldsymbol{x}_0 + oldsymbol{ ilde{P}}_m oldsymbol{y}_m = oldsymbol{x}_0 + oldsymbol{ ilde{P}}_{m-1} oldsymbol{y}_{m-1} + oldsymbol{ ilde{p}}_m y_m
onumber \ = oldsymbol{x}_{m-1} + oldsymbol{ ilde{p}}_m y_m$$

and

$$oldsymbol{r}_m = oldsymbol{b} - oldsymbol{A} oldsymbol{x}_m = oldsymbol{b} - oldsymbol{A} oldsymbol{x}_{m-1} - oldsymbol{A} oldsymbol{ ilde{p}}_m y_m \ = oldsymbol{r}_{m-1} - y_m oldsymbol{A} oldsymbol{ ilde{p}}_m.$$

Similar we get the residuals and search directions of the dual problem $A^T x^{dual} = b$:

$$\begin{split} \tilde{\boldsymbol{p}}_{m}^{dual} &= \boldsymbol{w}_{m} - l_{m} \tilde{\boldsymbol{p}}_{m-1}^{dual} \\ y_{m}^{dual} &= \boldsymbol{w}_{m}^{T} \boldsymbol{r}_{0}^{dual} - u_{m-1,m} / u_{m,m} \cdot y_{m-1} \\ \boldsymbol{r}_{m}^{dual} &= \boldsymbol{r}_{m-1}^{dual} - y_{m}^{dual} \boldsymbol{A}^{T} \tilde{\boldsymbol{p}}_{m}^{dual}. \end{split}$$

Together we have short recurrence formulas for all quantities. Again, we do not present all of the algebraic substitutions (see [89]) but only show the BiCG algorithm as it is usually formulated in Algorithm 1.7.

1.5.4 Conjugate Gradient Squared (CGS)

Let \mathcal{P}_m be the set of polynomials with maximum degree m. Then we can write

$$egin{aligned} m{r}_m &= \phi_m(m{A})m{r}_0, & m{r}_m^{dual} &= \phi_m(m{A}^T)m{r}_0^{dual} \ m{p}_m &= \psi_m(m{A})m{r}_0, & m{p}_m^{dual} &= \psi_m(m{A}^T)m{r}_0^{dual} \end{aligned}$$

with $\phi_m, \psi_m \in \mathcal{P}_m$. To satisfy the definitions in the BiCG algorithm, we have to define these polynomials via the following recurrence relations (see [89]):

$$\psi_m(\lambda) = \phi_m(\lambda) + \beta_{m-1}\psi_{m-1}(\lambda)$$

$$\phi_{m+1}(\lambda) = \phi_m(\lambda) - \alpha_m\lambda\psi_m(\lambda)$$

with $\psi_0(\lambda) = \phi_0(\lambda) \equiv 1$. The basic idea of the CGS algorithm is exploiting the fact

$$\langle \pi(\boldsymbol{A})\boldsymbol{r}_0 \mid \pi(\boldsymbol{A}^T)\boldsymbol{r}_0 \rangle = \langle \pi^2(\boldsymbol{A})\boldsymbol{r}_0 \mid \boldsymbol{r}_0 \rangle,$$

Given
$$\boldsymbol{x}_{0}$$

 $\boldsymbol{r}_{0} = \boldsymbol{r}_{0}^{dual} = \boldsymbol{b} - \boldsymbol{A}\boldsymbol{x}_{0}$
 $\boldsymbol{p}_{1} = \boldsymbol{p}_{1}^{dual} = \boldsymbol{r}_{0}$
for $m = 1, 2, \dots$
solve $\boldsymbol{M}\boldsymbol{z}_{m-1} = \boldsymbol{r}_{m-1}$
solve $\boldsymbol{M}^{T}\boldsymbol{z}_{m-1}^{dual} = \boldsymbol{r}_{m-1}^{dual}$
 $\rho_{m-1} = \langle \boldsymbol{r}_{m-1}^{dual} \mid \boldsymbol{z}_{m-1} \rangle$
if $\rho_{m-1} = 0$ method fails
if $m = 1$
 $\boldsymbol{p}_{1} = \boldsymbol{z}_{0}$
 $\boldsymbol{p}_{1}^{dual} = \boldsymbol{z}_{0}^{dual}$
else
 $\beta_{m-1} = \rho_{m-1}/\rho_{m-2}$
 $\boldsymbol{p}_{m} = \boldsymbol{z}_{m-1} + \beta_{m-1}\boldsymbol{p}_{m-1}$
 $\boldsymbol{p}_{m}^{dual} = \boldsymbol{z}_{m-1}^{dual} + \beta_{m-1}\boldsymbol{p}_{m-1}^{dual}$
 $\boldsymbol{q}_{m} = \boldsymbol{A}\boldsymbol{p}_{m}$
 $\boldsymbol{q}_{m}^{dual} = \boldsymbol{A}^{T}\boldsymbol{p}_{m}^{dual}$
 $\alpha_{m} = \rho_{m-1}/\langle \boldsymbol{p}_{m}^{dual} \mid \boldsymbol{q}_{m} \rangle$
 $\boldsymbol{x}_{m} = \boldsymbol{x}_{m-1} + \alpha_{m}\boldsymbol{p}_{m}$
 $\boldsymbol{r}_{m} = \boldsymbol{r}_{m-1} - \alpha_{m}\boldsymbol{q}_{m}^{dual}$

Algorithm 1.7: Preconditioned Bi-Conjugate Gradient algorithm (BiCG).

which holds for all polynomials $\pi \in \mathcal{P}_m$. Defining $\hat{\boldsymbol{r}}_m := \phi_m^2(\boldsymbol{A})\boldsymbol{r}_0$ and $\hat{\boldsymbol{p}}_m := \psi_m^2(\boldsymbol{A})\boldsymbol{r}_0$ yields

$$\begin{aligned} \alpha_m &= \frac{\langle \phi_m(\boldsymbol{A}) \boldsymbol{r}_0 \mid \phi_m(\boldsymbol{A}^T) \boldsymbol{r}_0 \rangle}{\langle \boldsymbol{A} \psi_m(\boldsymbol{A}) \boldsymbol{r}_0 \mid \psi_m(\boldsymbol{A}^T) \boldsymbol{r}_0 \rangle} = \frac{\langle \phi_m^2(\boldsymbol{A}) \boldsymbol{r}_0 \mid \boldsymbol{r}_0 \rangle}{\langle \boldsymbol{A} \psi_m^2(\boldsymbol{A}) \boldsymbol{r}_0 \mid \boldsymbol{r}_0 \rangle} \\ &= \frac{\langle \hat{\boldsymbol{r}}_m \mid \boldsymbol{r}_0 \rangle}{\langle \boldsymbol{A} \hat{\boldsymbol{p}}_m \mid \boldsymbol{r}_0 \rangle} \end{aligned}$$

and

$$\beta_m = \frac{\langle \phi_{m+1}(\boldsymbol{A})\boldsymbol{r}_0 \mid \phi_{m+1}(\boldsymbol{A}^T)\boldsymbol{r}_0 \rangle}{\langle \phi_m(\boldsymbol{A})\boldsymbol{r}_0 \mid \phi_m(\boldsymbol{A}^T)\boldsymbol{r}_0 \rangle} = \frac{\langle \hat{\boldsymbol{r}}_{m+1} \mid \boldsymbol{r}_0 \rangle}{\langle \hat{\boldsymbol{r}}_m \mid \boldsymbol{r}_0 \rangle}$$

Applying the recurrence relations we get

$$\psi_m^2(\lambda) = \phi_m^2(\lambda) + 2\beta_{m-1}\phi_m(\lambda)\psi_{m-1}(\lambda) + \beta_{m-1}^2\psi_{m-1}^2(\lambda),$$

$$\phi_{m+1}^2(\lambda) = \phi_m^2(\lambda) - \alpha_m\lambda\left(2\phi_m^2(\lambda) + 2\beta_{m-1}\phi_m(\lambda)\psi_{m-1}(\lambda) - \alpha_m\lambda\psi_{m-1}^2(\lambda)\right), \text{ and }$$

$$\phi_{m+1}(\lambda)\psi_m(\lambda) = \phi_m^2(\lambda) + \beta_{m-1}\phi_m(\lambda)\psi_{m-1}(\lambda) - \alpha_m\lambda\psi_{m-1}^2(\lambda).$$

Now we introduce two auxiliary vectors $\hat{\boldsymbol{q}}_m$ and $\hat{\boldsymbol{u}}_m$ as

$$\hat{\boldsymbol{q}}_m := \phi_{m+1}(\boldsymbol{A})\psi_m(\boldsymbol{A})\boldsymbol{r}_0$$

and

$$egin{aligned} \hat{oldsymbol{u}}_m &:= \phi_m(oldsymbol{A})\psi_m(oldsymbol{A})oldsymbol{r}_0 \ &= \phi_m^2(oldsymbol{A})oldsymbol{r}_0 + eta_{m-1}\phi_m(oldsymbol{A})\psi_{m-1}(oldsymbol{A})oldsymbol{r}_0 \ &= oldsymbol{\hat{r}}_m + eta_{m-1}oldsymbol{\hat{q}}_{m-1}. \end{aligned}$$

This yields

$$\hat{\boldsymbol{q}}_{m} := \phi_{m+1}^{2}(\boldsymbol{A})\boldsymbol{r}_{0} + \beta_{m-1}\phi_{m}(\boldsymbol{A})\psi_{m-1}(\boldsymbol{A})\boldsymbol{r}_{0} - \alpha_{m}\boldsymbol{A}\psi_{m-1}^{2}(\boldsymbol{A})$$
$$= \underbrace{\hat{\boldsymbol{r}}_{m} + \beta_{m-1}\hat{\boldsymbol{q}}_{m-1}}_{\hat{\boldsymbol{u}}_{m}} - \alpha_{m}\boldsymbol{A}\hat{\boldsymbol{p}}_{m}.$$

Using these auxiliary vectors we obtain the desired short recurrences for \hat{r}_{m+1} and \hat{p}_{m+1} :

$$\hat{\boldsymbol{r}}_{m+1} = \phi_{m+1}^2(\boldsymbol{A})\boldsymbol{r}_0$$

$$= \phi_m^2(\boldsymbol{A}) - \alpha_m \boldsymbol{A} \left(2\phi_m^2(\boldsymbol{A}) + 2\beta_{m-1}\phi_m(\boldsymbol{A})\psi_{m-1}(\boldsymbol{A}) - \alpha_m \boldsymbol{A}\psi_{m-1}^2(\boldsymbol{A}) \right)$$

$$= \hat{\boldsymbol{r}}_m - \alpha_m \boldsymbol{A} \left(\underbrace{\hat{\boldsymbol{r}}_m + \beta_{m-1}\hat{\boldsymbol{q}}_{m-1}}_{\hat{\boldsymbol{u}}_m} + \underbrace{\hat{\boldsymbol{r}}_m + \beta_{m-1}\hat{\boldsymbol{q}}_{m-1} - \alpha_m \boldsymbol{A}\hat{\boldsymbol{p}}_m}_{\hat{\boldsymbol{q}}_m} \right)$$

and

$$\begin{split} \hat{\boldsymbol{p}}_{m+1} &= \psi_{m+1}^2(\boldsymbol{A})\boldsymbol{r}_0 \\ &= \phi_{m+1}^2(\boldsymbol{A}) + 2\beta_m \phi_{m+1}(\boldsymbol{A})\psi_m(\boldsymbol{A}) + \beta_m^2 \psi_m^2(\boldsymbol{A}), \\ &= \hat{\boldsymbol{r}}_{m+1} + 2\beta_m \hat{\boldsymbol{q}}_m + \beta_m^2 \hat{\boldsymbol{p}}_m. \end{split}$$

Omitting the hats $(^)$ and doing some algebraic reformulation to save memory and computing time, we get the CGS algorithm as shown in Algorithm 1.8.

1.5.5 Generalized Minimal Residuals (GMRES)

GMRES is based on the Arnoldi algorithm to generate \mathcal{V}_m and a Petrov-Galerkin condition with $\mathcal{W}_m = \mathbf{A}\mathcal{V}_m$ to choose $\mathbf{x}_m \in \mathcal{V}_m$. That is $\mathbf{x}_m = \mathbf{x}_0 + \mathbf{V}_m \boldsymbol{\xi}_m$ has to satisfy

$$\boldsymbol{b} - \boldsymbol{A}\boldsymbol{x}_m \perp \boldsymbol{W}_m = \boldsymbol{A}\boldsymbol{V}_m. \tag{1.20}$$

With this particular \mathcal{W}_m , the Petrov-Galerkin condition is equivalent to a norm minimizing condition. Therefore, $\boldsymbol{\xi}_m \in \mathbb{R}^m$ minimizes the function

$$\phi_m(\boldsymbol{\xi}_m) := \| \boldsymbol{b} - \boldsymbol{A} \boldsymbol{x}_m \|_2 = \| \boldsymbol{b} - \boldsymbol{A} \boldsymbol{x}_0 - \boldsymbol{A} \boldsymbol{V}_m \boldsymbol{\xi}_m \|_2 = \| \boldsymbol{r}_0 - \boldsymbol{A} \boldsymbol{V}_m \boldsymbol{\xi}_m \|_2$$

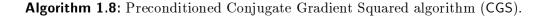
if and only if $\boldsymbol{x}_m = \boldsymbol{x}_0 + \boldsymbol{V}_m \boldsymbol{\xi}_m$ satisfies (1.20).

With the Arnoldi factorization (equation (1.7) on page 11) we get

$$\begin{split} \phi_{m}(\boldsymbol{\xi}_{m}) &= \|\boldsymbol{r}_{0} - \boldsymbol{A}\boldsymbol{V}_{m}\boldsymbol{\xi}_{m}\|_{2} \\ &= \|\|\boldsymbol{r}_{0}\|_{2}\boldsymbol{v}_{1} - (\boldsymbol{V}_{m}\boldsymbol{H}_{m} + h_{m+1,m} \cdot \boldsymbol{v}_{m+1}\boldsymbol{e}_{m}^{T})\boldsymbol{\xi}_{m}\|_{2} \\ &= \|\|\boldsymbol{r}_{0}\|_{2}\boldsymbol{V}_{m+1}\boldsymbol{e}_{1} - \boldsymbol{V}_{m+1}\tilde{\boldsymbol{H}}_{m}\boldsymbol{\xi}_{m}\|_{2} \\ &= \|\|\boldsymbol{V}_{m+1}(\|\boldsymbol{r}_{0}\|_{2}\boldsymbol{e}_{1} - \tilde{\boldsymbol{H}}_{m}\boldsymbol{\xi}_{m})\|_{2} \\ &= \|\|\boldsymbol{r}_{0}\|_{2}\boldsymbol{e}_{1} - \tilde{\boldsymbol{H}}_{m}\boldsymbol{\xi}_{m}\|_{2} \end{split}$$
(1.21)

Given
$$\boldsymbol{x}_{0}$$

 $\boldsymbol{r}_{0} = \boldsymbol{r}_{0}^{dual} = \boldsymbol{b} - \boldsymbol{A}\boldsymbol{x}_{0}$
choose $\tilde{\boldsymbol{r}}$, for example $\tilde{\boldsymbol{r}} = \boldsymbol{r}_{0}$
for $m = 1, 2, ...$
 $\rho_{m-1} = \langle \tilde{\boldsymbol{r}} \mid \boldsymbol{r}_{m-1} \rangle$
if $\rho_{m-1} = 0$ method fails
if $m = 1$
 $\boldsymbol{u}_{1} = \boldsymbol{r}_{0}$
 $\boldsymbol{p}_{1} = \boldsymbol{u}_{1}$
else
 $\beta_{m-1} = \rho_{m-1} / \rho_{m-2}$
 $\boldsymbol{u}_{m} = \boldsymbol{r}_{m-1} + \beta_{m-1}\boldsymbol{q}_{m-1}$
 $\boldsymbol{p}_{m} = \boldsymbol{u}_{m} + \beta_{m-1}(\boldsymbol{q}_{m-1} + \beta_{m-1}\boldsymbol{p}_{m-1})$
solve $\boldsymbol{M}\boldsymbol{z}_{m} = \boldsymbol{p}_{m}$
 $\hat{\boldsymbol{z}}_{m} = \boldsymbol{A}\boldsymbol{z}_{m}$
 $\alpha_{m} = \rho_{m-1} / \langle \hat{\boldsymbol{z}}_{m} \mid \tilde{\boldsymbol{r}} \rangle$
 $\boldsymbol{q}_{m} = \boldsymbol{u}_{m} - \alpha_{m}\hat{\boldsymbol{z}}_{m}$ solve $\boldsymbol{M}\boldsymbol{z}_{m} = \boldsymbol{u}_{m} + \boldsymbol{q}_{m}$
 $\hat{\boldsymbol{z}}_{m} = \boldsymbol{A}\boldsymbol{z}_{m}$
 $\hat{\boldsymbol{z}}_{m} = \boldsymbol{A}\boldsymbol{z}_{m}$
 $\hat{\boldsymbol{z}}_{m} = \boldsymbol{A}\boldsymbol{z}_{m}$



where

$$\tilde{\boldsymbol{H}}_{m} = \begin{pmatrix} \boldsymbol{H}_{m} \\ \boldsymbol{e}_{m}^{T} \cdot \boldsymbol{h}_{m+1,m} \end{pmatrix} = \begin{pmatrix} h_{1,1} & h_{1,2} & \cdots & h_{1,m-1} & h_{1,m} \\ h_{2,1} & h_{2,2} & \vdots & \vdots \\ & h_{3,2} & \vdots & \vdots \\ & & \ddots & \vdots & \vdots \\ & & & h_{m,m-1} & h_{m,m} \\ \hline 0 & \cdots & 0 & h_{m+1,m} \end{pmatrix} \in \mathbb{R}^{(m+1) \times m}$$

Thus, minimizing ϕ_m is equivalent to solving the least squares problem

$$\min_{oldsymbol{\xi}_m\in\mathbb{R}^m}\|\|oldsymbol{r}_0\|_2oldsymbol{e}_1-oldsymbol{ ilde{H}}_moldsymbol{\xi}_m\|_2.$$

A standard method for solving least squares problems is to factorize the m + 1 by m matrix \tilde{H}_m into an upper triangular m + 1 by m matrix R_m (with last row 0) and an m + 1 by m + 1 unitary matrix Q_m . Exploiting the special structure of \tilde{H}_m this generalized QR factorization can efficiently be computed with Givens rotations. The solution $\boldsymbol{\xi}_m$ is then obtained by solving the upper triangular m by m subsystem of

$$\boldsymbol{R}_{m}\boldsymbol{\xi}_{m} = \|\boldsymbol{r}_{0}\|_{2}\boldsymbol{Q}_{m}\boldsymbol{e}_{1}. \tag{1.23}$$

Suppose we have such a generalized QR factorization of \tilde{H}_m , i.e., we have m

Givens rotations \boldsymbol{G}_i with

$$\underbrace{\boldsymbol{G}_{m}\boldsymbol{G}_{m-1}\cdots\boldsymbol{G}_{1}}_{\boldsymbol{Q}_{m}}\tilde{\boldsymbol{H}}_{m}=\boldsymbol{R}_{m}=\begin{pmatrix}r_{1,1} & r_{1,2} & \cdots & r_{1,m}\\ 0 & r_{2,2} & \vdots \\ & \ddots & \ddots & \vdots \\ & & \ddots & & \vdots \\ & & & \ddots & r_{m,m}\\ & & & & 0\end{pmatrix}.$$

Applying \boldsymbol{Q}_m in the next step to $ilde{\boldsymbol{H}}_{m+1}$ yields

$$\boldsymbol{G}_{m}\boldsymbol{G}_{m-1}\cdots\boldsymbol{G}_{1}\cdot\tilde{\boldsymbol{H}}_{m+1} = \begin{pmatrix} r_{1,1} & r_{1,2} & \cdots & r_{1,m} \\ 0 & r_{2,2} & \vdots & & \vdots \\ & \ddots & \ddots & \vdots & & \vdots \\ & & \ddots & r_{m,m} & & r_{m,m+1} \\ \hline & & 0 & & \tilde{r}_{m+1,m+1} \\ \hline 0 & \cdots & 0 & & \tilde{r}_{m+2,m+1} \end{pmatrix}$$

Thus we just have to apply one more Givens rotation G_{m+1} to eliminate $\tilde{r}_{m+2,m+1}$:

$$\boldsymbol{G}_{m+1} := \begin{pmatrix} \boldsymbol{I} & \boldsymbol{0} \\ \boldsymbol{0}^T & \tilde{\boldsymbol{G}}_{m+1} \end{pmatrix} \text{ with } \tilde{\boldsymbol{G}}_{m+1} \begin{pmatrix} \tilde{r}_{m+1,m+1} \\ \tilde{r}_{m+2,m+1} \end{pmatrix} = \begin{pmatrix} r_{m+1,m+1} \\ 0 \end{pmatrix}$$

Collecting our results we can update \mathbf{R}_{m+1} from \mathbf{R}_m by applying \mathbf{Q}_m to the last column of $\tilde{\mathbf{H}}_{m+1}$, computing \mathbf{G}_{m+1} , and setting

$$r_{m+1,m+1} := \begin{cases} \operatorname{sign}(r_{m+1,m+1}) \sqrt{\tilde{r}_{m+1,m+1}^2 + \tilde{r}_{m+2,m+1}^2} & \text{if } \tilde{r}_{m+1,m+1} \neq 0\\ \tilde{r}_{m+2,m+1} & \text{if } \tilde{r}_{m+1,m+1} = 0 \end{cases}$$

and

 $r_{m+2,m+1} := 0.$

The right hand side vector in (1.23) is computed by applying all Givens rotations to the first unit vector e_1 . For the residual norm, we have

$$\|m{r}_m\|_2 = \|m{b} - m{A}m{x}_m\|_2 = \|\|m{r}_0\|_2m{e}_1 - m{Q}_mm{R}_mm{\xi}_m\|_2 = \|\|m{r}_0\|_2m{Q}_m^Tm{e}_1 - m{R}_mm{\xi}_m\|_2.$$

The first *m* elements of $\|\boldsymbol{r}_0\|_2 \boldsymbol{Q}_m^T \boldsymbol{e}_1 - \boldsymbol{R}_m \boldsymbol{\xi}_m$ are 0, because the *m* by *m* upper triangular subsystem in (1.23) is nonsingular and therefore the least squares problem is solved with error 0. Therefore, $\|\boldsymbol{r}_0\|_2 \boldsymbol{Q}_m^T \boldsymbol{e}_1 - \boldsymbol{R}_m \boldsymbol{\xi}_m = (0, \dots, 0, \|\boldsymbol{r}_0\|_2 \cdot \boldsymbol{e}_{m+1}^T \boldsymbol{Q}_m^T \boldsymbol{e}_1)^T$ and thus $\|\boldsymbol{r}_m\|_2 = \|\boldsymbol{r}_0\|_2 \cdot |\boldsymbol{e}_1^T \boldsymbol{Q}_m \boldsymbol{e}_{m+1}|$.

This enables us to check the residual norm quickly and only to compute the approximate solution \boldsymbol{x}_m if $\|\boldsymbol{r}_m\|_2$ is sufficiently small.

Since we have to store all Krylov basis vectors anyway, it might be advantageous to replace the Gram-Schmidt orthogonalization by the robust Householder process [140]. However, Rozložník showed, that GMRES with modified Gram-Schmidt is backward stable [50, 105] and thus the higher computational effort is not necessary.

```
Given \boldsymbol{x}_0
for j = 1, 2, ...
     \tilde{\boldsymbol{v}}_1 = \boldsymbol{b} - \boldsymbol{A}\boldsymbol{x}_0
     solve M \boldsymbol{z}_1 = \tilde{\boldsymbol{v}}_1
     m{v}_1 = m{z}_1 / \| m{z}_1 \|
     s_1 = \| \boldsymbol{z}_1 \| \boldsymbol{e}_1
     for m = 1, ..., M
          \tilde{\boldsymbol{v}}_{m+1} = \boldsymbol{A} \boldsymbol{v}_m
         solve M \boldsymbol{z}_{m+1} = \tilde{\boldsymbol{v}}_{m+1}
          for k = 1, ..., m
               h_{k,m} = \langle \boldsymbol{z}_{m+1} \mid \boldsymbol{v}_k \rangle
               \boldsymbol{z}_{m+1} = \boldsymbol{z}_{m+1} - h_{k,m} \boldsymbol{v}_k
          h_{m+1,m} = \|\boldsymbol{z}_{m+1}\|
          \boldsymbol{v}_{m+1} = \boldsymbol{z}_{m+1} / h_{m+1,m}
          apply G_1, \ldots, G_{m-1} to (h_{1,m} \ldots h_{m+1,m})^T
          compute G_m
          s_{m+1} = G_m s_m
         if s_{m+1,m+1} is small enough
               compute \boldsymbol{x}_{m+1}
               quit
     compute \boldsymbol{x}_{m+1}
     if convergence
          quit
     else
          \boldsymbol{x}_0 = \boldsymbol{x}_{m+1}
```

Algorithm 1.9: Preconditioned Generalized Minimal Residual algorithm with restart $(\mathsf{GMRES}(m))$.

1.5.6 Stabilized Variants and Quasi Minimization

• Smoothing Residual Norms

Inspecting the run of the curve of CGS residual norms, we often observe an irregular, oscillating behavior. This is due to the fact that CGS does two 'minimization' steps at once, but computes only once the search direction. Therefore CGS sometimes overshoots locally. To smooth these oscillations, van de Vorst [133, 134] coupled CGS with a repeatedly applied GMRES(1) iteration. This leads to a local minimization of the residual norms and therefore to a considerably smoother convergence behavior. Chan *et. al.* applied this idea to TFQMR and developed the so called QMRCGStab algorithm (see below).

• Quasi Minimization

There are various quasi minimization methods which try to retain the advantages of GMRES but only need short recurrences from Bi-Lanczos like algorithms. The most important are QMR (**Q**uasi Minimal Residuals) which combines BiCG with

GMRES, TFQMR (Transpose Free QMR) which is based on CGS and GMRES, and QMRCGStab (BiCGStab with GMRES).

We illustrate only the basic ideas of quasi minimization at the example of QMR. For further references see [18, 37, 91].

As in BiCG the *m*th iterate has the form $\boldsymbol{x}_m = \boldsymbol{x}_0 + \boldsymbol{V}_m \boldsymbol{\xi}_m$ and we have

$$oldsymbol{A}oldsymbol{V}_m = oldsymbol{V}_{m+1} ilde{oldsymbol{T}}_m$$

where $\tilde{\boldsymbol{T}}_m \in \mathbb{R}^{m+1 \times m}$ is tridiagonal. It follows that if $\boldsymbol{v}_1 = \boldsymbol{r}_0 / \|\boldsymbol{r}_0\|_2$, then

$$egin{aligned} \phi_m(m{\xi}_m) &:= \|m{b} - m{A}(m{x}_0 + m{V}_mm{\xi}_m)\|_2 \ &= \|m{r}_0 - m{A}m{V}_mm{\xi}_m\|_2 \ &= \|m{r}_0 - m{V}_{m+1}m{ ilde{T}}_mm{\xi}_m\|_2 \ &= \|m{V}_{m+1}(\|m{r}_0\|_2m{e}_1 - m{ ilde{T}}_mm{\xi}_m)\|_2 \end{aligned}$$

Unfortunately, V_{m+1} is not orthogonal and thus we cannot conclude $\phi_m(\boldsymbol{\xi}_m) = \|\|\boldsymbol{r}_0\|_2 \boldsymbol{e}_1 - \boldsymbol{T}_m \boldsymbol{\xi}_m\|_2$ as we did in GMRES. However, introducing the scaling matrix $\boldsymbol{S}_{m+1} = \text{diag}(\boldsymbol{V}_{m+1})$ we have

$$\phi_m(\boldsymbol{\xi}_m) \le \|\boldsymbol{V}_{m+1}\boldsymbol{S}_{m+1}^{-1}\|_2 \cdot \|\boldsymbol{S}_{m+1}(\|\boldsymbol{r}_0\|_2 \boldsymbol{e}_1 - \tilde{\boldsymbol{T}}_m \boldsymbol{\xi}_m)\|_2$$
(1.24)

with $\|V_{m+1}S_{m+1}^{-1}\|_2 \leq \sqrt{m+1}$. In the QMR algorithm we simply neglect this first factor in (1.24) and only minimize the second.

Applying these ideas to CGS/BiCGStab leads to the TFQMR respectively QMR-CGStab algorithm.

CHAPTER

Krylov Methods and Floating-Point Arithmetic

⁶⁶ Accuracy and precision are the same for the scalar computation c = a * b, but accuracy can be much worse than precision in the solution of a linear system of equations, for example. ⁹

Nicholas J. Higham, 1996

It makes me nervous to fly on airplanes since I know they are designed using floating-point arithmetic??

Alston S. Householder

Preconditioned Krylov subspace solvers are frequently used for solving large sparse linear systems. There are many advantageous properties concerning convergence rates and error estimates but unfortunately, if we implement such a solver on a computer, we often observe an unexpected and even contrary behavior (see e.g. [48,65,128]). The basic reason for this is that standard computer arithmetic is based on floating-point numbers, i.e., numbers with a finite precision. In Section 2.1 we give a short overview about floating-point arithmetics, Section 2.3.1 describes possible failures during preconditioning and Section 2.3.2 gives some deeper insight into the effect of finite precision arithmetic to Krylov methods.

2.1 Floating-Point Arithmetics

2.1.1 Floating-Point Numbers

Nearly all important computer architectures used today are providing so called floating-point numbers for numerical computations. The floating-point number space $\mathcal{F} = \mathcal{F}(b, l, e_{\min}, e_{\max})$ is a finite subset of IR, characterized by its base b, the length of its mantissa l, the minimal exponent e_{\min} , and maximal exponent e_{\max} .

A floating-point number itself consists of three parts: a sign s (+ or -), a fixed length mantissa $0.m_1m_2...m_l$ with digits $m_i \in \{1,...,b-1\}$, and an exponent ewith $e_{\min} \leq e \leq e_{\max}$, given to the base b. These define the floating-point number $s m \cdot b^e$. To get unique representations, the first mantissa digit m_1 is required to be nonzero (otherwise, e.g., we had $0.01b^e = 0.10b^{e-1}$). Numbers with this property are called normalized. For $e = e_{\min}$ it is not necessary to require $m_1 \neq 0$ in order to retain the uniqueness of the representation. That is, we may allow non-normalized numbers¹ between 0 and $\pm 0.1b^{e_{\min}}$.

		normalized mantissas				
		0.100_{2}	0.101_{2}	0.110_{2}	0.111_{2}	
exponent	3	2.0	2.5	3.0	3.5	
	2	1.0	1.25	1.5	1.75	
	0	0.5	0.625	0.75	0.875	
e.	-1	0.25	0.3125	0.375	0.4375	
		denormalized mantissas				
		0.000_{2}	0.001_{2}	0.010_{2}	0.011_{2}	
	-1	0.0	0.0625	0.125	0.1875	

Table 2.1 and Figure 2.1 illustrate the floating-point system $\mathcal{F}(2,3,-1,2)$.

Table 2.1: The floating-point system $\mathcal{F}(2,3,-1,2)$ with denormalized numbers



Figure 2.1: The floating-point system $\mathcal{F}(2, 3, -1, 2)$ with denormalized numbers

The two most common floating-point systems are defined in the IEEE-754 standard [4]. The first is called 'single precision' which is a $\mathcal{F}(2, 24, -125, 128)$ system and the second is called 'double precision' which is $\mathcal{F}(2, 53, -1021, 1024)^2$. Both systems provide the special numbers $-0, \pm \inf(infinity)$, and nan (not-a-number).

¹Having non-normalized numbers also ensures the existence of unique additive inverses in \mathcal{F} .

 $^{^2\}mathrm{Note}$ that this definition differs a little from standard due to our definition of normalized numbers.

2.1.2 Roundings

Obviously, (\mathcal{F}, \circ) with $\circ \in \{+, -, *, /\}$ is not closed, i.e., with $a, b \in \mathcal{F}$ we generally don't have $a \circ b \in \mathcal{F}$. Therefore we need a *rounding* after each operation, to map $a \circ b$ back into the floating-point screen \mathcal{F} .

Usually, one is interested to obtain the nearest number out of \mathcal{F} . In this case we define $\Box(a \circ b) =: x \in \mathcal{F}$ with³ $|x - (a \circ b)| = \min_{\xi \in \mathcal{F}} \{|\xi - (a \circ b)|\}$. Unfortunately, with this rounding we neither know the exact size nor the sign of the error we made by substituting the mathematically correct $a \circ b$ with the computer representable $\Box(a \circ b)$. However, at least for the relative error we have

$$\frac{|\Box(a \circ b) - a \circ b|}{|a \circ b|} \le \frac{b^{-l-1}}{2} := \epsilon/2 \quad \text{if } a \circ b \neq 0.$$

This ϵ is called *machine precision*. It is the smallest possible relative error in \mathcal{F} , i.e., $(1 - \min_{x \in \mathcal{F}} \{x > 1\}).$

To retain at least the sign of the error, we may use directed roundings:

$$\nabla(a \circ b) = \max_{\xi \in \mathcal{F}} \{\xi \le a \circ b\}, \qquad \triangle(a \circ b) = \min_{\xi \in \mathcal{F}} \{\xi \ge a \circ b\}$$

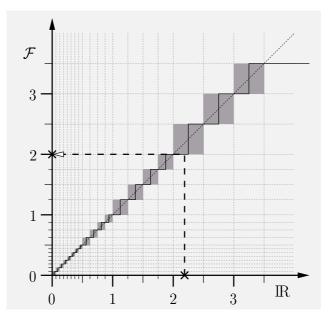


Figure 2.2: Round to nearest in $\mathcal{F}(2, 3, -1, 2)$ with denormalized numbers

Figure 2.2 illustrates 'round to nearest' in our example floating-point screen $\mathcal{F}(2,3,-1,2)$. Since a rounding can only take values in \mathcal{F} , it is completely defined by its saltus in each of the grey boxes ([75]). That is the smallest possible rounding is ∇ , with the saltus on the right edge in each box and \triangle is the largest one, with the saltus on the left edge.

³If $a \circ b$ lies exactly in the middle between two floating-point numbers, then the one with even mantissa, i.e. with $m_l = 0$, is chosen ('round to nearest even').

2.2 Finite Precision Behavior of Lanczos Procedures

Rounding errors greatly affect the behavior of the Lanczos iteration [12, 64, 122, 143]. The basic difficulty with finite precision arithmetic is the loss of orthogonality among the generated basis vectors of the Krylov subspaces. The central results about Lanczos error analysis are mostly based on the fundamental and excellent work of Paige, see e.g. [93, 94].

Implementing the Lanczos procedure with modified Gram-Schmidt orthogonalization in finite precision arithmetic we have to take rounding errors into consideration. That is, each value, computed in finite precision, may differ from the exact one and thus would need a different notation. However, for simplicity we retain the notation unchanged in this section and explicitly point out where we use exact quantities.

Paige shows that in finite precision equation (1.11) on page 13 has to be extended to

$$\boldsymbol{A}\boldsymbol{V}_{m} = \boldsymbol{V}_{m}\boldsymbol{T}_{m} + \boldsymbol{v}_{m+1}\boldsymbol{\beta}_{m}\boldsymbol{e}_{m}^{T} + \boldsymbol{F}_{m}, \qquad (2.1)$$

where \boldsymbol{F}_m contains the rounding errors.

We define

$$\epsilon_0 := 2(n+4)\epsilon \quad ext{and} \quad \epsilon_1 := 2(7+nnz \cdot \| \| \boldsymbol{A} \| \|_2 / \| \boldsymbol{A} \|_2)\epsilon$$

where ϵ denotes the machine precision and nnz is the maximum number of nonzero elements in any row of A. Under the assumptions that

$$\epsilon_0 < \frac{1}{12}, \qquad m(3\epsilon_0 + \epsilon_1) < 1,$$

and ignoring higher order terms in ϵ , Paige shows that

$$\|\boldsymbol{F}_m\|_2 \leq \sqrt{m}\epsilon_1 \|\boldsymbol{A}\|_2$$

or, a little bit more sloppy, $\|\boldsymbol{F}_m\|_2$ is approximately of size $\epsilon \|\boldsymbol{A}\|_2$ [100, 119]. This means, the tri-diagonalization holds up to machine precision. Unfortunately, the situation becomes much more difficult concerning the orthogonality condition. This fact is stated in the following theorem.

Theorem 2.1 Suppose $(\mathbf{S}_m, \mathbf{\Lambda}_m)$ with $\mathbf{\Lambda}_m := \text{diag}(\lambda_1, \ldots, \lambda_m)$ is the exact spectral factorization of the computed \mathbf{T}_m (see equation. (2.1)), i.e., the $(\mathbf{z}_j, \lambda_j)$ are the computed Ritz-pairs of \mathbf{A} if we define $\mathbf{z}_j := \mathbf{V}_m \mathbf{s}_j$. Then, we have

$$\angle(\boldsymbol{z}_j, \boldsymbol{v}_{m+1}) \approx Arccos\left(\frac{\epsilon \|\boldsymbol{A}\|_2}{\|\boldsymbol{A}\boldsymbol{z}_j - \lambda_j \boldsymbol{z}_j\|_2 - \epsilon \|\boldsymbol{A}\|_2\|}\right).$$

Notice that \boldsymbol{v}_{m+1} should stay orthogonal to $\mathcal{K}_m = \operatorname{span}(\boldsymbol{v}_1, \ldots, \boldsymbol{v}_m)$ and $\boldsymbol{z}_j \in \mathcal{K}_m$, because it is a linear combination of the columns of \boldsymbol{V}_m — the basis vectors. So \boldsymbol{v}_{m+1} should stay orthogonal to \boldsymbol{z}_j , too. If $(\boldsymbol{z}_j, \lambda_j)$ is a good approximation to any eigenpair of \boldsymbol{A} , i.e., $\|\boldsymbol{A}\boldsymbol{z}_j - \lambda_j\boldsymbol{z}_j\|_2$ is small, then $\angle(\boldsymbol{z}_j, \boldsymbol{v}_{m+1}) \approx \operatorname{Arccos}(1) = 0$. This means, ironically, if a Ritz-pair is converging (and this is exactly what we want in eigenvalue computations) we lose our orthogonality in \boldsymbol{V}_m completely. Nevertheless we can easily delay this loss of orthogonality using a higher precision arithmetic, i.e., a small machine ϵ .

Proof: Separating the last column of (2.1) by multiplying it with e_m from the right we get

$$\beta_m \boldsymbol{v}_{m+1} = \boldsymbol{A} \boldsymbol{v}_m - \alpha_m \boldsymbol{v}_m - \beta_{m-1} \boldsymbol{v}_{m-1} - \boldsymbol{f}_m$$

Now multiplication with V_m^T from left yields

$$\beta_{m} \underbrace{\mathbf{V}_{m}^{T} \mathbf{v}_{m+1}}_{=:\mathbf{q}_{m+1}} = \underbrace{\mathbf{V}_{m}^{T} \mathbf{A}}_{\mathbf{T}_{m}^{T} \mathbf{V}_{m}^{T} + \beta_{m} \mathbf{e}_{m} \mathbf{v}_{m}^{T} \mathbf{v}_{m}} - \beta_{m-1} \mathbf{V}_{m}^{T} \mathbf{v}_{m-1} - \mathbf{V}_{m}^{T} \mathbf{f}_{m}$$

$$\Leftrightarrow \qquad \beta_{m} \mathbf{q}_{m+1} = \mathbf{T}_{m} \overleftarrow{\mathbf{q}}_{m} - \alpha_{m} \overleftarrow{\mathbf{q}}_{m} - \beta_{m-1} \overleftarrow{\mathbf{q}}_{m-1} + \beta_{m} \mathbf{e}_{m} \mathbf{v}_{m+1}^{T} \mathbf{v}_{m}$$

$$+ \underbrace{\mathbf{F}_{m}^{T} \mathbf{v}_{m} - \mathbf{V}_{m}^{T} \mathbf{f}_{m}}_{=:\mathbf{g}_{m}}$$

$$\Leftrightarrow \qquad \beta_{m} \mathbf{q}_{m+1} \mathbf{e}_{m}^{T} = \mathbf{T}_{m} \overleftarrow{\mathbf{Q}}_{m} - \overleftarrow{\mathbf{Q}}_{m} \mathbf{T}_{m} + \overleftarrow{\mathbf{G}}_{m}. \qquad (2.2)$$

Where, to facilitate additions of vectors with different lengths and the collection of vectors with different lengths as columns of a matrix, we introduced the $\stackrel{\leftrightarrow}{\cdot}$ -operator which extends a vector to the appropriate size by adding some zero-elements at the bottom.

Let now the vectors \mathbf{s}_j , \mathbf{z}_j , and the scalars λ_j be defined in the same way as in the theorem. Multiplying (2.2) with \mathbf{s}_j^T from left and \mathbf{s}_j from right for any $j \leq m$ yields

$$\beta_{m} \underbrace{\boldsymbol{s}_{j}^{T} \boldsymbol{V}_{m}^{T}}_{\boldsymbol{z}_{j}^{T}} \boldsymbol{v}_{m+1} \underbrace{\boldsymbol{e}_{m}^{T} \boldsymbol{s}_{j}}_{\boldsymbol{s}_{mj}} = \underbrace{\boldsymbol{s}_{j}^{T} \boldsymbol{T}_{m}}_{\boldsymbol{\lambda}_{j} \boldsymbol{s}_{j}^{T}} \overleftrightarrow{\boldsymbol{Q}}_{m} \boldsymbol{s}_{j} - \boldsymbol{s}_{j}^{T} \overleftrightarrow{\boldsymbol{Q}}_{m} \underbrace{\boldsymbol{T}_{m} \boldsymbol{s}_{j}}_{\boldsymbol{\lambda}_{j} \boldsymbol{s}_{j}} + \boldsymbol{s}_{j}^{T} \overleftrightarrow{\boldsymbol{G}}_{m} \boldsymbol{s}_{j}$$

$$\Leftrightarrow \qquad \boldsymbol{z}_{j}^{T} \boldsymbol{v}_{m+1} = \frac{\boldsymbol{s}_{j}^{T} \overleftrightarrow{\boldsymbol{G}}_{m} \boldsymbol{s}_{j}}{\beta_{m} \boldsymbol{s}_{mj}}.$$

$$(2.3)$$

The numerator of this fraction is approximately of size $\epsilon \|\boldsymbol{A}\|_2 \|\boldsymbol{z}_j\|_2$. In order to estimate the denominator we look at the *quality* of the Ritz-pair $(\lambda_j, \boldsymbol{z}_j)$:

$$\|\boldsymbol{A}\boldsymbol{z}_{j} - \lambda_{j}\boldsymbol{z}_{j}\|_{2} = \|\underbrace{\boldsymbol{A}\boldsymbol{V}_{m}}_{\boldsymbol{V}_{m}\boldsymbol{T}_{m}+\beta_{m}\boldsymbol{v}_{m+1}\boldsymbol{e}_{m}^{T}+\boldsymbol{F}_{m}} \\ \leq \|\beta_{m}\boldsymbol{v}_{m+1}s_{mj}\|_{2} + \|\boldsymbol{F}_{m}\|_{2} \\ \approx |\beta_{m}s_{mj}| + \epsilon \|\boldsymbol{A}\|_{2}.$$
(2.4)

Collecting our results in (2.3) and (2.4) we finally obtain

$$\angle(\boldsymbol{z}_j, \boldsymbol{v}_{m+1}) = \operatorname{Arccos}\left(\frac{|\boldsymbol{z}_j^T \boldsymbol{v}_{m+1}|}{\|\boldsymbol{z}_j\|_2 \|\boldsymbol{v}_{m+1}\|_2}\right) \approx \operatorname{Arccos}\left(\frac{\epsilon \|\boldsymbol{A}\|_2}{\|\|\boldsymbol{A}\boldsymbol{z}_j - \lambda_j \boldsymbol{z}_j\|_2 - \epsilon \|\boldsymbol{A}\|_2\|}\right)$$

An obvious way to retain the orthogonality among the Lanczos vectors is to orthogonalize each newly computed vector to all its predecessors. In fact, this is almost equivalent to applying the Arnoldi procedure to a symmetric matrix. With this approach, the computed Lanczos vectors are orthogonal up to machine precision but we have to pay a high price concerning computing time and storage demands.

A more carefully inspection of Theorem 2.1 shows that re-orthogonalizing against all previous Lanczos vectors is not necessary [120]. The *bad guys* are only the (almost) converged Ritz vectors \boldsymbol{z} (with Ritz value λ), since $\angle(\boldsymbol{z}, \boldsymbol{v}_k)$ tends to zero only if $\|\boldsymbol{A}\boldsymbol{z} - \lambda\boldsymbol{z}\|_2$ is sufficiently small. Exploiting this fact leads to so called selective orthogonalization [101] where only *good* Ritz vectors are stored and used for re-orthogonalizing of newly computed Lanczos vectors. In this sense, a Ritz pair $(\boldsymbol{z}, \lambda)$ is called *good* if it satisfies

$$\|\boldsymbol{A}\boldsymbol{z} - \lambda \boldsymbol{z}\|_2 \leq \sqrt{\epsilon} \|\boldsymbol{A}\|_2.$$

An even more refined reorthogonalization strategy called partial reorthogonalization can be found in [121].

2.3 Examples

2.3.1 Preconditioning

Preconditioning is very important in solving linear systems because well conditioned systems are much easier and particularly faster to solve. However, preconditioning in finite precision can cause a drastically perturbed solution. Thus an important question is: *How does preconditioning affect the solution of a linear system?*

Since in modern Krylov subspace solvers preconditioning is no separate step but an inherent part to the solver itself (see Algorithm 1.2 on page 14), it cannot be distinguished which part of the deviation between the solutions of a preconditioned and a non-preconditioned system is caused by the preconditioning itself and which part is induced by various other error sources.

To give an idea of the magnitude of the perturbation that can be caused by preconditioning we consider the following example. We use a Jacobi preconditioner, which is so easy to apply to an entire linear system, that it is often used to transform A and b in advance. Afterwards a non-preconditioning solver is applied. This means we scale A to get a unit diagonal (we just perform one division for each element of A and b). This operation is done in IEEE double-precision. In order to identify the error caused by this scaling operation, we apply a verifying solver to the scaled system. Here we solve systems $\mathsf{GK4.16}(n)$ which result from a 5-point discretization of a fourth order ODE, see (6.1) on page 89. The resulting verified solutions of the scaled systems are shown in Figure 2.3.

For comparison we also plotted the verified solutions of the non-preconditioned systems (compare [34]). They differ so little that they appear as a single line (x). Additionally, we drew the solution of the underlying continuous problem. The discretization errors are so small that this curve is also not distinguishable from the non-preconditioned solutions (x).

This example demonstrates that preconditioning in this traditional way may introduce unacceptably large errors which can be significantly larger than the discretization error of an underlying continuous problem (which is less than 10^{-6} in our examples). It is possible to avoid these problems by doing verified preconditioning

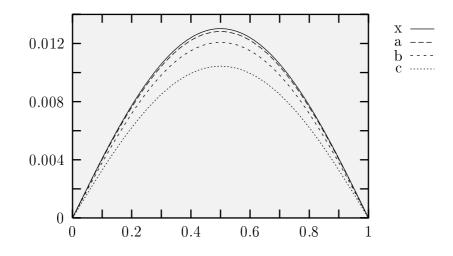


Figure 2.3: The effect of preconditioning on ill-conditioned linear systems. The curves represent the solutions of the preconditioned systems a: GK(8191), b: GK(12287), and c: GK(16383) with cond $\approx 10^{14}$, 10^{15} , and 10^{16} , resp. The exact solutions (compare [34]) of the non-preconditioned systems differ so little, that they appear as a single line (x).

but this leads to an interval matrix and therefore to an often unacceptable computational effort to solve these systems. Another, maybe more practical way is to use a more accurate arithmetic to avoid significant propagation of the various errors.

2.3.2 Convergence

One well known Krylov subspace method is the Conjugate Gradient algorithm. It can be interpreted as a Lanczos procedure and a subsequent LDL^{T} -factorization to solve the tridiagonal system $T\tilde{x} = \tilde{b}$ (see Section 1.5.2) [41]. Then the residuals turn out to be scaled versions of the Krylov basis vectors and hence should stay orthogonal to their preceding residuals.

In Figure 2.4 the Euclidean norms of the residual and the error in each step are plotted during solving a $\mathsf{GK}(1023)$ system. Additionally we show the level of orthogonality of the new residual-vector \mathbf{r}_{m+1} to the previous ones: $\max_{k=1}^{m} \{\langle \mathbf{r}_k |$ $\mathbf{r}_{m+1} \rangle / (||\mathbf{r}_k||_2 ||\mathbf{r}_{m+1}||_2)\}$. As we can see, there is no convergence at all up to step m = 1.5n and particularly no convergence at the theoretically guaranteed step m = n. One reason is easy to identify (see Theorem 2.1 or [51, 128, 129]): the basis of the Krylov subspace loses its orthogonality completely at $m \approx 400$ and the basisvectors may even become linearly dependent. So CG can't minimize the residual in the entire \mathbb{R}^m but only in a smaller subspace.

Further we can observe that the error norm runs into saturation at a level of approximately 10^{-6} . This matches with the well known rule of thumb saying that we may lose up to $\log(\text{cond}(\mathbf{A}))$ (≈ 10 in this case) digits from the 16 decimal digits we have in IEEE double precision.

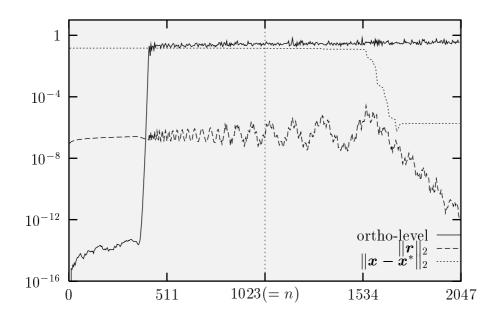


Figure 2.4: The Euclidean norms of the residual (dashed) and the error (dotted), and the level of orthogonality (solid) during solving the GK(1023) system. (ortho-level = $\max_{k=1}^{m} \{ \langle \boldsymbol{r}_{k} \mid \boldsymbol{r}_{m+1} \rangle / (||\boldsymbol{r}_{k}||_{2} ||\boldsymbol{r}_{m+1}||_{2}) \})$

Chapter

Improved Arithmetics

⁶⁶ Numerical precision is the very soul of science ⁹⁹

Sir D'Arcy Wentworth Thompson, 1942

In many numerical algorithms there is a large gap between the theoretical, i.e., mathematical, behavior on the one hand and the finite precision behavior on the other hand. In cases where the accuracy of a result is insufficient or no results can be obtained at all due to poorly conditioned problems, it is desirable to have a *better* arithmetic. Particularly, iterative algorithms sometimes even speed up because a higher precision arithmetic produces fewer errors that have to be minimized by the algorithm and therefore often iterations can be saved. In fact, it is nearly always possible to save iterations, but since the computing time per iteration increases with higher precision, we only sometimes really save time. However, we should always take into consideration that the higher computing time per iteration is often due to missing hardware support. For example the *exact scalar product* (Section 3.1), suitably supported in hardware, can be computed at least as fast as the standard scalar product (see [74]).

To narrow the gap between exact and finite precision arithmetic, often some minor arithmetic improvements suffice to get the desired results. One uses a more precise arithmetic (see Section 3.2). A second possibility is to leave the data type but control the rounding errors introduced by arithmetic operations performed on these numbers as described in Section 3.3. Further on we may pick out some frequently used operations or functions and improve their accuracy. In particular, we focus on the scalar product (Section 3.1) which is a fundamental operation in numerical linear algebra. We start with this *exact scalar product* because this concept is needed for one of the higher precision arithmetics we describe in Section 3.2.

3.1 The Exact Scalar Product

Since scalar products occur very frequently and are important basic operations in numerical linear algebra, it is advantageous to perform this operation with the same precision as the other basic operations like addition or subtraction. Usually, scalar products are implemented by use of ordinary multiplication and addition and we have to beware of a lot of roundoff errors and their amplification due to cancellation. This is not necessary as has been shown by Kulisch [67, 74, 75].

The basic idea is first to multiply the floating-point vector elements exactly, that means we have to store the result in a floating-point format with double mantissa length and doubled exponent range¹. Secondly we have to accumulate these products without any roundoff errors, see Figure 3.1. One possibility to achieve this is by use

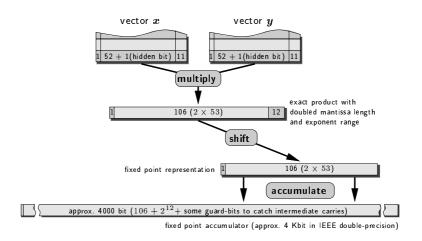


Figure 3.1: The basic idea of the exact scalar product, implemented by means of a long accumulator.

of a fixed point accumulator that covers the doubled floating-point number range plus some extra bits for intermediate overflows. At a first glance one might think that this accumulator must be very large, but in fact for the IEEE double-precision format, a little more than half a kilobyte is sufficient: 106 mantissa bits (for a zero exponent) plus 2^{11} binary digits for all possible left shifts and the same for right shifts plus one sign bit and some guard bits.



In Karlsruhe we built this operation in hardware as a numerical coprocessor called XPA-3233 (eXtended Precision Arithmetic on a 32 bit PCI bus with 33 MHz clock speed), see [10]. On the XPA-3233 we use 67×64 bit words of storage. With this 92 guard bits even a tera-flop computer would need more than a hundred million years to cause an

¹That is a sign bit, 2×53 mantissa bits and 11 + 1 bits for the exponent, i.e., a total of 119 bits for the IEEE double-precision format.

overflow². After accumulation of all products we usually have to round the result into the floating-point format which is symbolized with the \Box -operator. That is, the entire scalar product operation can be computed with just one rounding at all and therefore we have the much sharper bound

$$\left|\sum_{i=1}^{n} x_{i} y_{i} - \Box \left(\sum_{i=1}^{n} x_{i} y_{i}\right)\right| \leq \epsilon \left|\sum_{i=1}^{n} x_{i} y_{i}\right|$$

for the relative error than we usually have for scalar products if we use an ordinary floating-point arithmetic with a rounding after each multiplication and accumulation (see [61, 90])

$$\left|\sum_{i=1}^{n} x_i y_i - \sum_{i=1}^{n} x_i \boxdot y_i\right| \le \frac{n\epsilon}{1 - n\epsilon} \sum_{i=1}^{n} |x_i y_i|,$$

which can be arbitrarily bad if $\sum_{i} |x_i y_i| \gg |\sum_{i} x_i y_i|$.

3.2 Multiple Precision

There are various tools and libraries providing numbers with a higher precision than the build-in data types (usually IEEE double precision) [13, 103, 123] We can subdivide them in two fundamental types. The first implements exact numbers, i.e., numbers with infinite precision while the latter offers higher but finite precision. Because of storage and computing time requirements, we only focus on the latter multiple precision numbers. They are subdivided ac-

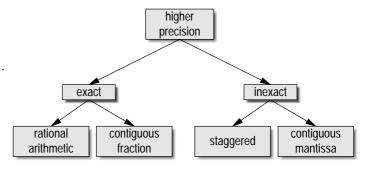


Figure 3.2: Higher precision data types.

cording to different implementation techniques in so called *staggered* numbers which are basically a sum of ordinary floating-point numbers (Section 3.2.1) and numbers with a contiguous mantissa, i.e., long floating point numbers (Section 3.2.2). The latter are mostly implemented using an integer field for the mantissa and some additional memory for the exponent and sign.

3.2.1 Staggered Precision Numbers

In this section we introduce the basic ideas of the so called staggered arithmetic. We give the definition of staggered precision numbers and show how the basic arithmetic operations can be performed by use of the exact scalar product (see Section 3.1). Instancing the square root as an example, we illustrate how elementary functions can be realized for staggered precision numbers.

Definition 3.1 Given l floating-point numbers $x^{(1)}, \ldots, x^{(l)}$ we define a staggered precision number (or short: staggered number) x with staggered length l by

$$x := \sum_{k=1}^{l} x^{(k)}.$$

²In fact, the XPA3233 only provides 90 guard bits, because 2 bits have a special meaning.

To ensure maximum precision we are interested in *non-overlapping* staggered numbers, that is, staggered numbers with $|x^{(1)}| > \ldots > |x^{(l)}|$ and the exponents of two successive summands $x^{(k)}$, $x^{(k+1)}$ differ at least by the mantissa length m of the floating-point system. In this case, the staggered number x represents a high precision number with at least $m \cdot l$ mantissa digits but with the same exponent range as the underlying floating-point system.

Let us assume, for example, we have a floating-point system with 3 decimal mantissa digits. Then with $x^{(1)} = 1.65 \cdot 10^3$, $x^{(2)} = 3.94 \cdot 10^0$, $x^{(3)} = 5.75 \cdot 10^{-5}$, and $x^{(4)} = 2.24 \cdot 10^{-8}$ the staggered number x of length 4:

$$x = 1.65 \cdot 10^3 + 3.94 \cdot 10^0 + 5.75 \cdot 10^{-5} + 2.24 \cdot 10^{-8} = 1.6539400575224 \cdot 10^3$$

represents a higher precision floating-point number with a minimum of 12 (14 in this case) decimal digits

• Basic Arithmetic Operations

Designing the basic arithmetic operations for staggered numbers, we have to decide which precision, i.e., which staggered length the result should have. Trying always to represent the exact result soon leads to very large numbers and fails already for the division where we usually get infinitely many digits. Therefore we define the staggered length of the resulting staggered number as the maximum of the staggered lengths of the operands.

Having the exact scalar product available, the algorithms for addition, subtraction and even multiplication are really simple. We just accumulate the result in a long accumulator and subsequently we round out the summands of the resulting staggered number as described in Algorithm 3.1. Further on we refer to this algorithm by the notation $z = round_to_staggered(accu)$ where z is a staggered number and accu is a long accumulator.

```
Given a long accumulator accu

l = staggered\_length\_of(z)

for k = 1, ..., l

z^{(k)} = round\_towards\_zero(accu)

accu = accu - z^{(k)}
```

Algorithm 3.1: Successively rounding out a long accumulator accu into the summands of a staggered number z.

As an example for a basic arithmetic operation we state the subtraction. Given two staggered numbers $x := \sum_{k=1}^{l_1} x^{(k)}$ and $y := \sum_{k=1}^{l_2} y^{(k)}$ we compute $z := \sum_{k=1}^{\max\{l_1, l_2\}} z^{(k)} := x - y$ as showed in Algorithm 3.2.

• Elementary Functions

Since in this thesis, we only need the square root function in addition to the basic arithmetic operations, we only describe this function. However, with the basic arithmetic operations at hand, it is also possible to compute other elementary functions. Given two staggered numbers x and y $l_1 = staggered_length_of(x); l_2 = staggered_length_of(y)$ accu = 0for $k = 1, ..., l_1$ $accu = accu + x^{(k)}$ for $k = 1, ..., l_2$ $accu = accu - y^{(k)}$ $set_length(z, max{l_1, l_2}) z = round_to_staggered(accu)$

Algorithm 3.2: Subtraction of staggered numbers x and y. The intermediate result is stored in the long accumulator *accu*.

To compute the inverse of a given elementary function it is often helpful to use a Newton iteration. In the case of square roots $y = \sqrt{x}$ we set $f(y) = y^2 - x$ and approximate the zero y^* of f, i.e., the square root of x by a sequence y_i defined by

$$y_0 = \sqrt{\Box\left(\sum_{k=1}^l x^{(k)}\right)}, \qquad y_i = y_{i-1} - \frac{f(y_{i-1})}{f'(y_{i-1})} = \frac{1}{2}\left(\frac{x}{y_{i-1}} + y_{i-1}\right), \quad i = 1, 2, \dots$$

This algorithm is also known as Heron algorithm. Since the Newton iteration converges quadratically in a neighborhood of y^* , \sqrt{l} iterations should suffice to obtain enough correct digits for y.

The staggered technique is implemented for example in the XSC languages [69] and is massively based on the availability of the exact scalar product. The special case where the number of ordinary floating-point numbers used to define a staggered number is fixed to 2 can also be coded by use of some arithmetic tricks without the exact scalar product. These tricks are mainly based on ideas by Dekker and Kahan [25]. A fast implementation of this latter technique in C++ is the doubledouble library, see [80].

• Vector Operations

Since we use the exact scalar product anyway for computing the products and additions in scalar products of staggered vectors, we can easily implement a exact scalar product for staggered vectors. Avoiding the intermediate roundings, the exact staggered scalar product is even faster than the ordinary scalar product. We describe this exact inner multiplication of staggered vectors in Algorithm 3.3.

3.2.2 Contiguous Mantissas

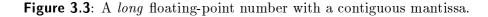
The contiguous mantissa type numbers can be addressed as long floating-point numbers. Usually, there is an integer variable containing the sign and exponent and an array of integers storing the mantissa. This is illustrated in Figure 3.3

In comparison to the staggered numbers, there is no possibility to exploit gaps, i.e., zeroes in the mantissa, but we have a more compact representation since we only store one exponent for the entire number. There exists an excellent implementation in C/assembler, the Gnu Multi-Precision library (GNU MP or GMP) [46] which

Given two staggered vectors $\boldsymbol{x} = (x_i)_{i=1}^n$ and $\boldsymbol{y} = (y_i)_{i=1}^n$ $l_1 = staggered_length_of(\boldsymbol{x})$ $l_2 = staggered_length_of(\boldsymbol{y})$ accu = 0for i = 1, ..., nfor $j = 1, ..., l_1$ $accu = accu + x_i^{(j)}y_i^{(k)}$ $z = round_to_staggered(accu)$

Algorithm 3.3: Exact scalar product of staggered vectors \boldsymbol{x} and \boldsymbol{y} . The intermediate result is stored in the long accumulator *accu*.

exponent	mantissa	
Lsign		



aims to provide the fastest multiple precision library. For this purpose there are carefully optimized assembler routines for nearly all important architectures and a generic C formulation for non-standard computers.

This results in a very high performance multiple precision arithmetic. Figure 3.4 shows the computing time for 100 000 evaluations of a typical expression in staggered precision arithmetic (C-XSC) and in a contiguous mantissa arithmetic (GMP). The speedup of GMP against staggered arithmetic levels off at approximately 25 which is mainly due to two reasons: First, the software simulation of the exact scalar product uses integer arithmetic, i.e., the floating-point summands of a staggered number have to be decomposed and composed repeatedly. This is avoided for the contiguous mantissa type numbers, because they use their own number format. Secondly this software simulation is relatively slow anyway since it is written in C instead of assembler as is the GMP arithmetic.

Since there was no object oriented interface for gmp, I implemented one with a complete set of overloaded operators. This interface is called gmp++ and is included in **vk** (see Chapter 6). It enables us to use multiple precision numbers for generic algorithms (compare Section 5.1).

3.3 Interval Arithmetic

In order to get reliable results, e.g., for error bounds of linear systems, it is not sufficient to reduce the rounding errors. We have to control this arithmetic uncertainty completely [3]. As we have seen in Section 2.1, each basic operation — even if it has floating-point operands — may have a result which is not representable exactly in the given floating point format. Thus each basic operation involves a rounding back into the floating-point screen and therefore causes an error. Clearly, there is no necessity to use *random* roundings and if we are interested in reliable bounds to the exact result, we simply have to return two floating-point numbers: One, which is guaranteed to be larger than the exact result, e.g., it's upwardly rounded value and

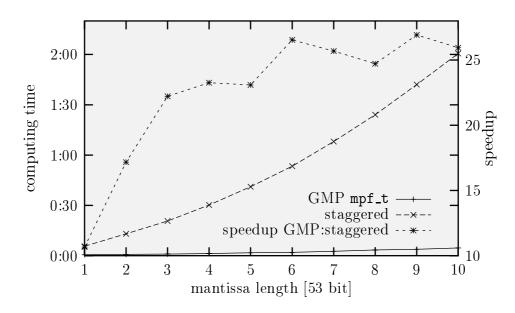


Figure 3.4: Computing time for 100 000 evaluations of a typical expression in staggered precision arithmetic (C-XSC) and in a contiguous mantissa arithmetic (GMP).

one which is smaller, e.g., it's downwardly rounded value. This idea automatically leads to an interval data type as defined in the following definition.

Definition 3.2 Let $\underline{x} \leq \overline{x}$, then we call the set $[x] := [\underline{x}, \overline{x}] := \{\xi \in \mathbb{R} \mid \underline{x} \leq \xi \leq \overline{x}\}$ an interval.

Note: even if we have such an interval on a computer, that is, \underline{x} and \overline{x} are floating-point numbers, the interval $[\underline{x}, \overline{x}]$ contains all real numbers between \underline{x} and \overline{x} and not only floating-point numbers.

Since intervals are elements of the power-set of IR, the basic arithmetic operations are defined by restriction of the power-set operations

$$[\underline{x}, \overline{x}] \circ [\underline{y}, \overline{y}] := \{ \xi \circ \eta \mid \underline{x} \le \xi \le \overline{x} \land \underline{y} \le \eta \le \overline{y} \} \text{ with } \circ \in \{+, -, \times, /\}$$

$$(and \ 0 \notin [y, \overline{y}] \text{ for } \circ = /)$$

$$(3.1)$$

Exploiting monotonicity properties, this infinitely many operations needed to compute $[x, \bar{x}] \circ [y, \bar{y}]$ in (3.1) reduce to the computation of only a few operations. For example the addition of intervals can be computed by $[x, \bar{x}] + [y, \bar{y}] = [x + y, \bar{x} + \bar{y}]$. Unfortunately, the right hand side of this equation, i.e., the bounds of the resulting interval may not be representable in our floating-point format. To make sure that we enclose the exact solution interval on the computer, we again have to replace x + y by a smaller floating-point number and $\bar{x} + \bar{y}$ by a larger one [73, 76]. Usually this lower respectively upper floating-point bounds are obtained by selecting the correct directed rounding mode (compare Section 2.1). On a computer we define the addition of intervals via

$$[\underline{x}, \overline{x}] + [\underline{y}, \overline{y}] := [\nabla(\underline{x} + \underline{y}), \triangle(\overline{x} + \overline{y})], \qquad (3.2)$$

where \bigtriangledown and \triangle denotes the downward respectively upward directed rounding. Sometimes we use the shorter notation

$$[\underline{x}, \overline{x}] + [y, \overline{y}] := \diamondsuit([\underline{x} + y, \overline{x} + \overline{y}]).$$

There are various libraries providing interval data types, including the XSC languages [57, 58, 69], Profil/BIAS [71, 72] and Intlab [111] and newly there are commercial Fortran/C/C++ compilers from Sun-Microsystems [131] providing an interval data type. Using other libraries, one must carefully inspect if correct rounding modes are used.

There is also a interval-staggered library which provides higher precision intervals in Pascal-XSC [82].

• Excursion: Enclosing Floating-Point Computations

In particular, the exact scalar product is very useful if we have to compute an enclosure of a linear expression only with floating-point arguments. For example, if we have a floating-point matrix A, a floating-point right hand side vector b, and floating-point approximate solution \tilde{x} and we want to compute an enclosure of the residual vector, we can either cast all floating-point numbers to intervals (via the \diamond -operator) and then use standard interval arithmetic (\diamond , \diamond), see Algorithm 3.4a, or we can use the exact scalar product to compute the exact residual. Finally, this exact but long number has to be rounded to a (very tight) interval with floating-point bounds and relative diameter less than or equal to ϵ , see Algorithm 3.4b.

for $i = 1,, n$ $r_i = \diamondsuit b_i$ for $j = 1,, n$ $r_i = r_i \diamondsuit a_{i,j} \oslash \tilde{x}_j$	for $i = 1,, n$ $accu = b_i$ for $j = 1,, n$ $accu = accu - a_{i,j} \cdot \tilde{x}_j$ /* exact */ $r_i = \diamondsuit accu$
a) with interval arithmetic	b) with the exact scalar product

Algorithm 3.4: Computation of an enclosure of the residual with two different techniques.

With approach a), the relative error in component i can only be bounded by

$$\frac{(n+1)\epsilon}{1-(n+1)\epsilon} \cdot \left(|r_i| + \sum_{j=1}^n |a_{i,j}\tilde{x}_j| \right),$$

which might be arbitrary bad, if $|r_i| + \sum_{j=1}^n |a_{i,j}\tilde{x}_j| \gg \left|r_i - \sum_{j=1}^n a_{i,j}\tilde{x}_j\right|$.

CHAPTER

Error Bounds for Solutions of Linear Systems

⁶⁶ No method of solving a computational problem is really available to a user until it is completely described in an algebraic computing language and made completely reliable. **?**

George E. Forsythe, 1967

This chapter gives an overview about the most important verification methods for linear systems of equations. Section 4.1 shows straight forward extensions of pointalgorithms to interval algorithms, while Section 4.2 focuses on verification algorithms based on fixed point theorems [76, 86, 108]. The latter class of algorithms is much more general than the first one, but suffers in two important ways from the underlying interval arithmetic. First, due to the software simulated interval arithmetic, they are relatively slow and secondly, one has to pay either with a big computational overhead or a significant loss of accuracy because of the so called wrapping effect (see [92]).

In Section 4.3 we present a different class of verification algorithms. Instead of delivering an enclosure of each solution component, they only compute a rigorous bound for the error norm. This might be disadvantageous if the solution components have highly different magnitudes. However, since most of the computation

can be done approximately (i.e., in ordinary floating-point arithmetic) with only a subsequent verification procedure, the resulting algorithms often are much faster than the previous ones.

4.1 Interval Extensions of Point Algorithms

The most obvious way to obtain a verifying algorithm is using a given pointalgorithm and replacing every floating-point operation by the corresponding interval operation. It has been shown that, for example, the interval version of Gaussian elimination is executable in this way for diagonally dominant matrices or M-matrices. For general matrices the intervals tend to grow in diameter rapidly and it may soon happen that a pivot column solely consists of intervals containing zero. In this case, the algorithm terminates prematurely without computing an enclosure of the solution. However, if \boldsymbol{A} has special properties, the Interval Gauß algorithm (IGA) may even be capable to produce optimal enclosures, that is, the interval vector $[\boldsymbol{x}]$ is the smallest *n*-dimensional box enclosing the solution set [87]. See Paragraph Solutions of symmetric tridiagonal systems in Section 4.3.3 for an example with particular tridiagonal matrices.

4.2 Enclosures via Fixed Point Methods

Suppose we have an approximate inverse R for A then we can define a sequence of vectors $(\boldsymbol{x}_k)_{k \in \mathbb{N}}$ by

$$\boldsymbol{x}_{k+1} = \boldsymbol{x}_k + \boldsymbol{R}(\boldsymbol{b} - \boldsymbol{A}\boldsymbol{x}_k) = \boldsymbol{R}\boldsymbol{b} + (\boldsymbol{I} - \boldsymbol{R}\boldsymbol{A})\boldsymbol{x}_k$$

This vector sequence converges for every \boldsymbol{x} if and only if the spectral radius of $\boldsymbol{I} - \boldsymbol{R}\boldsymbol{A}$ is less than one.

If \mathcal{X} is a non-empty, convex, and compact subset of \mathbb{R}^n then by Brouwer's fixed point theorem

$$\mathcal{X} \supseteq \mathcal{X} + \boldsymbol{R}(\boldsymbol{b} - \boldsymbol{A}\mathcal{X}) \quad ext{implies} \quad \exists \, \boldsymbol{x} \in \mathcal{X} : \, \boldsymbol{R}(\boldsymbol{b} - \boldsymbol{A}\boldsymbol{x}) = \boldsymbol{o}.$$

Using an interval vector $[\mathbf{x}]$ as a special non-empty, convex, and compact subset of \mathbb{R}^n , we generally have diam $([\mathbf{x}] + \mathbf{R}(\mathbf{b} - \mathbf{A}[\mathbf{x}])) > \text{diam}([\mathbf{x}])$ and thus $[\mathbf{x}]$ will never be a superset of $[\mathbf{x}] + \mathbf{R}(\mathbf{b} - \mathbf{A}[\mathbf{x}])$. Moreover, only if \mathbf{R} is nonsingular, then we guarantee that $\mathbf{A}\mathbf{x} = \mathbf{b}$ is solvable, i.e., that \mathbf{A} is nonsingular, too. These two problems are solved by the next theorem (compare [106]).

Theorem 4.1 Let A, $R \in \mathbb{R}^{n \times n}$ and $b \in \mathbb{R}^n$. Suppose for the interval vector [x] holds

$$\boldsymbol{R}\boldsymbol{b} + (\boldsymbol{I} - \boldsymbol{R}\boldsymbol{A})[\boldsymbol{x}] \subseteq [\boldsymbol{x}]$$
 (4.1)

then A and R are nonsingular and there is exactly one $x \in [x]$ satisfying Ax = b.

See [106] for a proof.

In a floating-point system (with denormalized mantissas), numbers are much narrower around zero. Therefore it is always a good idea to work with the error, i.e., the difference of the exact solution \boldsymbol{x}^* and the approximate solution $\tilde{\boldsymbol{x}}$ since this is hopefully close to zero. Applying Theorem 4.1 to $\tilde{\boldsymbol{x}} + [\boldsymbol{x}]$, condition (4.1) now reads

$$Rb + (I - RA)(\tilde{x} + [x]) \subseteq (\tilde{x} + [x])$$

$$\Rightarrow Rb + \tilde{x} - RA\tilde{x} + (I - RA)[x] \subseteq \tilde{x} + [x]$$

$$\Rightarrow R(b - A\tilde{x}) + (I - RA)[x] \subseteq [x]$$

$$(4.2)$$

If condition (4.2) does not hold one may initiate the following iteration process

$$\begin{split} & [\boldsymbol{x}]^{(0)} := [\boldsymbol{x}], \\ & \widetilde{[\boldsymbol{x}]}^{(k)} := \boldsymbol{R}(\boldsymbol{b} - \boldsymbol{A}\tilde{\boldsymbol{x}}) + (\boldsymbol{I} - \boldsymbol{R}\boldsymbol{A})[\boldsymbol{x}]^{(k-1)} \\ & [\boldsymbol{x}]^{(k)} := [1 - \epsilon, 1 + \epsilon] \cdot \widetilde{[\boldsymbol{x}]}^{(k)} & \text{for } k = 1, 2, \dots \end{split}$$

Then for some $k \in \mathbb{N}$ and $[\boldsymbol{x}]$ the inclusion $[\boldsymbol{x}]^{(k)} \subseteq [\stackrel{\circ}{\boldsymbol{x}}]^{(k-1)}$ holds. The question is for which $[\boldsymbol{x}]$ and which k we will obtain $[\boldsymbol{x}]^{(k)} \subseteq [\boldsymbol{x}]$. The answer is given by the following theorem.

Theorem 4.2 Let A, $R \in \mathbb{R}^{n \times n}$ and $b \in \mathbb{R}^n$. Then the following two statements are equivalent:

1. For each interval n-vector $[\mathbf{x}]$ with

$$\operatorname{diam}([\boldsymbol{x}]) > \frac{2}{1 - \rho(|\boldsymbol{I} - \boldsymbol{R}\boldsymbol{A}|)} \cdot |\boldsymbol{R}(\boldsymbol{b} - \boldsymbol{A}\tilde{\boldsymbol{x}})|$$

there exists a $k \in \mathbb{N}$ with $[\boldsymbol{x}]^{(k)} \subseteq [\stackrel{\circ}{\boldsymbol{x}}]^{(k-1)}$

2. $\rho(|I - RA|) < 1$

See [107] for a proof.

Obviously, we do not need \mathbf{R} explicitly and it is sufficient to have a, e.g., triangular factorization LU of \mathbf{A} to compute $\mathbf{R}(\mathbf{b} - \mathbf{A}\tilde{\mathbf{x}})$ and $\mathbf{R}\mathbf{A}[\mathbf{x}]$ via forward and backward substitution. Note that this will often produve large overestimations for $\mathbf{R}\mathbf{A}[\mathbf{x}]$ due to the wrapping effect which occurs in solving triangular systems. To avoid this problem, one can either use a coordinate transformation technique as described by R. Lohner in [81] or one can substitute intervals by zonotopes which might cover the shape of the solution more appropriate. However, these triangular factors may exploit a possible sparsity in \mathbf{A} (e.g., banded Cholesky) to make this algorithm applicable to larger matrices, compare [81].

4.3 Error Bounds via Perturbation Theory

Usually, stopping criteria for iterative solvers of linear systems are based on the norm of the residual $\mathbf{r} = \mathbf{b} - \mathbf{A}\tilde{\mathbf{x}}$. Since $\|\tilde{\mathbf{x}} - \mathbf{x}^*\|_2 \leq \|\mathbf{A}^{-1}\|_2 \cdot \|\mathbf{r}\|_2$, this gives a rough idea about the distance to the exact solution $\mathbf{x}^* = \mathbf{A}^{-1}\mathbf{b}$ if we have some information about $\|\mathbf{A}^{-1}\|_2$ or the condition of \mathbf{A} . Sometimes 'cheap' condition estimators [56] are used to estimate $\operatorname{cond}_2(A)$ but this approach gives only error estimates instead of bounds. Our first task will be to compute a verified upper bound of $\|\tilde{x} - x^*\|_2$. Then we try to improve this worst case bound by taking into account some more knowledge about the spectrum of A.

4.3.1 Basic Error Bounds

This section is essentially based on ideas of S. Rump [109, 110]. With $\|\boldsymbol{A}\|_2 \geq \sigma_{\min}(\boldsymbol{A})$ we have $\|\boldsymbol{A}^{-1}\|_2 \leq \sigma_{\min}^{-1}(\boldsymbol{A})$ and therefore

$$\| ilde{oldsymbol{x}} - oldsymbol{x}^*\|_2 \leq \sigma_{\min}^{-1}(oldsymbol{A}) \cdot \|oldsymbol{r}\|_2$$
 .

A well known method to compute the smallest singular value of a matrix \boldsymbol{A} is the inverse power method (with shift 0) (see [61]). Therefore it is necessary to have a factorization, say $(\boldsymbol{L}, \boldsymbol{U})$ of \boldsymbol{A} that enables us to compute $(\boldsymbol{L}\boldsymbol{U})^{-1}\boldsymbol{z}$ for arbitrary \boldsymbol{z} easily. Mostly, $\boldsymbol{L}\boldsymbol{U} = \boldsymbol{A}$ doesn't hold exactly but $\boldsymbol{L}\boldsymbol{U} = \tilde{\boldsymbol{A}} \approx \boldsymbol{A}$ is sufficient and it is often possible to get $\|\Delta \boldsymbol{A}\| = \|\tilde{\boldsymbol{A}} - \boldsymbol{A}\|$ fairly small. The next theorem clarifies how $\|\tilde{\boldsymbol{x}} - \boldsymbol{x}^*\|$ depends on the smallest singular value $\sigma_{\min}(\tilde{\boldsymbol{A}}), \Delta \boldsymbol{A}$ and $\|\boldsymbol{r}\|$.

Theorem 4.3 Let $\mathbf{A} \in \mathbb{R}^{n \times n}$, $\mathbf{b} \in \mathbb{R}^n$ be given as well as a nonsingular $\tilde{\mathbf{A}} \in \mathbb{R}^{n \times n}$ and $\tilde{\mathbf{x}} \in \mathbb{R}^n$. Define $\Delta \mathbf{A} := \tilde{\mathbf{A}} - \mathbf{A}$, $\mathbf{r} := \mathbf{b} - \mathbf{A}\tilde{\mathbf{x}}$ and suppose

$$\sigma_{min}(\tilde{A}) > n^{1/2} \cdot \|\Delta A\|_{\infty}$$

Then A is nonsingular and for $x^* := A^{-1}b$ holds

$$\|oldsymbol{x}^* - ilde{oldsymbol{x}}\|_\infty \leq rac{n^{1/2} \cdot \|oldsymbol{r}\|_\infty}{\sigma_{min}(ilde{oldsymbol{A}}) - n^{1/2} \cdot \|\Deltaoldsymbol{A}\|_\infty}.$$

Proof: Since $\|\tilde{\boldsymbol{A}}^{-1}\Delta \boldsymbol{A}\|_{2} \leq \sigma_{\min}(\tilde{\boldsymbol{A}})^{-1} \cdot \|\Delta \boldsymbol{A}\|_{2} \leq n^{1/2} \cdot \sigma_{\min}(\tilde{\boldsymbol{A}})^{-1} \cdot \|\Delta \boldsymbol{A}\|_{\infty} < 1$ the matrix $\boldsymbol{I} - \tilde{\boldsymbol{A}}^{-1}\Delta \boldsymbol{A} = \tilde{\boldsymbol{A}}^{-1}\boldsymbol{A}$ and hence \boldsymbol{A} itself is invertible. Now

$$(\boldsymbol{I} - \tilde{\boldsymbol{A}}^{-1} \Delta \boldsymbol{A})(\boldsymbol{x}^* - \tilde{\boldsymbol{x}}) = \tilde{\boldsymbol{A}}^{-1} \boldsymbol{A}(\boldsymbol{x}^* - \tilde{\boldsymbol{x}}) = \tilde{\boldsymbol{A}}^{-1} \cdot \boldsymbol{r}$$

and therefore

$$\|\boldsymbol{x}^* - \tilde{\boldsymbol{x}}\| = \|((\boldsymbol{I} - \tilde{\boldsymbol{A}}^{-1} \Delta \boldsymbol{A})^{-1} \cdot \tilde{\boldsymbol{A}}^{-1} \cdot \boldsymbol{r})\| \le \|\tilde{\boldsymbol{A}}^{-1} \cdot \boldsymbol{r}\| \cdot \|(\boldsymbol{I} - \tilde{\boldsymbol{A}}^{-1} \Delta \boldsymbol{A})^{-1}\|.$$
(4.3)

Using $\|(\boldsymbol{I} - \boldsymbol{B})^{-1}\|_{\infty} \leq (1 - \|\boldsymbol{B}\|_{\infty})^{-1}$ for convergent \boldsymbol{B} (i.e., $\|\boldsymbol{B}\|_{\infty} < 1$) we get

$$\|oldsymbol{x}^* - oldsymbol{ ilde{x}}\|_{\infty} \leq rac{\|oldsymbol{ ilde{A}}^{-1}\|_{\infty} \cdot \|oldsymbol{r}\|_{\infty}}{1 - \|oldsymbol{ ilde{A}}^{-1} \cdot \Delta oldsymbol{A}\|_{\infty}} \leq rac{\|oldsymbol{ ilde{A}}^{-1}\|_{\infty} \cdot \|oldsymbol{r}\|_{\infty}}{1 - \|oldsymbol{ ilde{A}}^{-1}\|_{\infty} \cdot \|\Delta oldsymbol{A}\|_{\infty}}$$

and applying

$$\|\boldsymbol{B}\|_{\infty} \le n^{1/2} \cdot \|\boldsymbol{B}\|_{2} \le n^{1/2} \cdot \sigma_{\min}(\boldsymbol{B})$$
(4.4)

yields

$$\|\boldsymbol{x}^* - \tilde{\boldsymbol{x}}\|_{\infty} \leq \frac{n^{1/2} \cdot \sigma_{\min}(\boldsymbol{A})^{-1} \cdot \|\boldsymbol{r}\|_{\infty}}{1 - n^{1/2} \cdot \sigma_{\min}(\tilde{\boldsymbol{A}})^{-1} \cdot \|\Delta \boldsymbol{A}\|_{\infty}}$$

Of course for sparse matrices this 2-norm bound (4.4) is a rough overestimation of the ∞ -norm. Suppose \boldsymbol{B} to have at most m nonzero elements per row and let $\beta := \max_{i,j} \{|\boldsymbol{B}_{i,j}|\}$ then both $\|\boldsymbol{B}\|_1$ and $\|\boldsymbol{B}\|_{\infty}$ are bounded by $m \cdot \beta$. Using $\|\boldsymbol{B}\|_2^2 \leq \|\boldsymbol{B}\|_1 \cdot \|\boldsymbol{B}\|_{\infty}$ we get

$$\|\boldsymbol{B}\|_{2} \leq (\|\boldsymbol{B}\|_{1} \cdot \|\boldsymbol{B}\|_{\infty})^{1/2} \leq m \cdot \beta.$$

Theorem 4.4 Let $\mathbf{A} \in \mathbb{R}^{n \times n}$, $\mathbf{b} \in \mathbb{R}^n$ be given as well as a nonsingular $\tilde{\mathbf{A}} \in \mathbb{R}^{n \times n}$ and $\tilde{\mathbf{x}} \in \mathbb{R}^n$. Define $\Delta \mathbf{A} := \tilde{\mathbf{A}} - \mathbf{A}$, $\mathbf{r} := \mathbf{b} - \mathbf{A}\tilde{\mathbf{x}}$ and suppose $\sigma_{\min}(\tilde{\mathbf{A}}) > (\|\Delta \mathbf{A}\|_1 \cdot \|\Delta \mathbf{A}\|_{\infty})^{1/2}$.

Then A is nonsingular and for $x^* := A^{-1}b$ holds

$$\|oldsymbol{x}^* - ilde{oldsymbol{x}}\|_\infty \leq \|oldsymbol{x}^* - ilde{oldsymbol{x}}\|_2 \leq rac{\|oldsymbol{r}\|_2}{\sigma_{min}(ilde{oldsymbol{A}}) - (\|\Deltaoldsymbol{A}\|_1 \cdot \|\Deltaoldsymbol{A}\|_\infty)^{1/2}}.$$

Proof: Starting with equation (4.3) but using the 2-norm instead of the ∞ -norm we get

$$\|oldsymbol{x}^* - oldsymbol{ ilde{x}}\|_2 \leq rac{\|oldsymbol{ ilde{A}}^{-1}\|_2 \cdot \|oldsymbol{r}\|_2}{1 - \|oldsymbol{ ilde{A}}^{-1}\|_2 \cdot \|\Deltaoldsymbol{A}\|_2} \leq rac{\|oldsymbol{r}\|_2}{\sigma_{\min}(oldsymbol{ ilde{A}}) - (\|\Deltaoldsymbol{A}\|_1 \cdot \|\Deltaoldsymbol{A}\|_\infty)^{1/2}}.$$

In practical computations it is a difficult task to get the smallest singular value of an arbitrary matrix \boldsymbol{A} or at least a reliable lower bound of $\sigma_{\min}(\boldsymbol{A})$. But if we have an approximate decomposition, say $(\boldsymbol{L}, \boldsymbol{U})$ with $\tilde{\boldsymbol{A}} = \boldsymbol{L}\boldsymbol{U}$ we can apply inverse power iteration to $\boldsymbol{L}\boldsymbol{U}$ to compute $\sigma_{\min}(\tilde{\boldsymbol{A}})$. Of course, if we can compute a lower bound of $\sigma_{\min}(\boldsymbol{A})$ directly then by setting $\tilde{\boldsymbol{A}} := \boldsymbol{A}$ we get $\Delta \boldsymbol{A} = \boldsymbol{0}$ and thus

$$\|oldsymbol{x}^* - ilde{oldsymbol{x}}\|_\infty \leq \|oldsymbol{x}^* - ilde{oldsymbol{x}}\|_2 \leq \sigma_{\min}(oldsymbol{A})^{-1} \cdot \|oldsymbol{r}\|_2.$$

4.3.2 Improved Error Bounds

The ideas of this section are mostly due to Dahlquist [24] who stated an interesting connection between Lanczos procedures and Gaussian quadrature rules. Later on, these ideas were sophisticated in [43, 45].

We start with the well known relation between the error norm and the residual norm

$$\|\boldsymbol{x}^* - \tilde{\boldsymbol{x}}\|_2 = \|\boldsymbol{A}^{-1}\boldsymbol{b} - \tilde{\boldsymbol{x}}\|_2 = \|\boldsymbol{A}^{-1}(\boldsymbol{b} - \boldsymbol{A}\tilde{\boldsymbol{x}})\|_2 = \|\boldsymbol{A}^{-1}\boldsymbol{r}\|_2.$$
 (4.5)

Suppose \boldsymbol{A} to be symmetric positive definite then we have real positive eigenvalues $\lambda_1, \ldots, \lambda_n$ (non-increasingly ordered) and an orthonormal basis of eigenvectors $\boldsymbol{q}_1, \ldots, \boldsymbol{q}_n$. Using $\boldsymbol{\Lambda} := \operatorname{diag}(\lambda_1, \ldots, \lambda_n)$ and $\boldsymbol{Q} := (\boldsymbol{q}_1 \mid \cdots \mid \boldsymbol{q}_n)$ yields

$$\|\boldsymbol{x}^{*} - \tilde{\boldsymbol{x}}\|_{2} = \|\boldsymbol{Q}^{T}(\boldsymbol{x}^{*} - \tilde{\boldsymbol{x}})\|_{2} = \|\boldsymbol{Q}^{T}\boldsymbol{A}^{-1}\boldsymbol{r}\|_{2}$$

$$= \|\boldsymbol{Q}^{T}\boldsymbol{A}^{-1}\boldsymbol{Q} \cdot \boldsymbol{Q}^{T}\boldsymbol{r}\|_{2} = \|\boldsymbol{\Lambda}^{-1} \cdot \boldsymbol{Q}^{T}\boldsymbol{r}\|_{2}$$

$$\leq \|\boldsymbol{\Lambda}^{-1}\|_{2} \cdot \|\boldsymbol{Q}^{T}\boldsymbol{r}\|_{2} = \lambda_{\min}^{-1} \cdot \|\boldsymbol{r}\|_{2}.$$
 (4.6)

This estimate is a worst case bound which does not make any use of the extent to which each of the eigenvectors \boldsymbol{q}_i is actually present in \boldsymbol{r} , i.e., of the size of the elements of $\boldsymbol{\mu} := \boldsymbol{Q}^T \boldsymbol{r}$. With (4.5) and (4.6) we see

$$egin{aligned} \|m{x}^* - ilde{m{x}}\|_2^2 &= \|m{A}^{-1}m{r}\|_2^2 = \langlem{A}^{-1}m{r} \mid m{A}^{-1}m{r}

angle \ &= \langlem{\Lambda}^{-1}m{Q}^Tm{r} \mid m{\Lambda}^{-1}m{Q}^Tm{r}
angle \ &= \langlem{\mu} \mid m{\Lambda}^{-2}m{\mu}
angle \ &= \sum_{i=1}^n \lambda_i^{-2}\cdot \mu_i^2. \end{aligned}$$

In general the relation $\langle \boldsymbol{r} | \boldsymbol{A}^{-2} \boldsymbol{r} \rangle = \sum_{i=1}^{n} \lambda_i^{-2} \mu_i^2$ holds for any analytic function f and the sum can be considered as a Riemann-Stieltjes integral, see e.g. [42]:

$$\langle \boldsymbol{r} \mid f(\boldsymbol{A})\boldsymbol{r} \rangle = \sum_{i=1}^{n} f(\lambda_i) \cdot \mu_i^2 = \int_a^b f(\lambda) d\mu$$
 (4.7)

with the piecewise constant, positive, and increasing measure μ defined via

Note that the interval [a, b] must contain the spectrum of \mathbf{A} , in particular $a \leq \lambda_{\min}$ must hold.

Unfortunately we do not know the eigenvalues and eigenvectors of A and hence we cannot evaluate (4.7) directly. However, one way to obtain bounds for Riemann-Stieltjes integrals is to use Gauß and Gauß-Radau quadrature formulas.

Evaluating (4.7) with a Gauß quadrature rule with m nodes (ξ_1, \ldots, ξ_m) and corresponding weights $(\omega_1, \ldots, \omega_m)$ we get

$$\int_{a}^{b} f(\lambda)d\mu = \underbrace{\sum_{j=1}^{m} \omega_{j}f(\xi_{j})}_{I_{\text{Gauß}}^{(m)}} + \underbrace{\frac{f^{(2m)}(\eta)}{(2m)!} \cdot \int_{a}^{b} \prod_{j=1}^{m} (\lambda - \xi_{j})^{2} d\mu}_{R_{\text{Gauß}}^{(m)}}, \quad \eta \in (a, b).$$

with an integral approximation $I_{\text{Gauß}}^{(m)}$ and the remainder $R_{\text{Gauß}}^{(m)}$. Note that for $f(\cdot) = \cdot^{-2}$ we get

$$R_{\text{Gauß}}^{(m)} = \frac{(-1)^{2m}(2m+1)!\lambda^{-(2m+2)}}{(2m)!} \cdot \int_{a}^{b} \prod_{j=1}^{m} (\lambda - \xi_j)^2 d\mu \ge 0$$

and therefore $I_{\text{Gauß}}^{(m)} \leq \int_{a}^{b} f(\lambda) d\mu$.

To obtain an upper bound, we apply a Gauß-Radau rule to (4.7) with m + 1nodes $(\xi_1, \ldots, \xi_{m+1})$ with $\xi_{m+1} := a$ and corresponding weights $(\omega_1, \ldots, \omega_{m+1})$

$$\int_{a}^{b} f(\lambda) d\mu = \underbrace{\sum_{j=1}^{m+1} \omega_{j} f(\xi_{j})}_{I_{\text{Radau}}^{(m+1)}} + \underbrace{\frac{f^{(2m+1)}(\eta)}{(2m+1)!} \cdot \int_{a}^{b} (\lambda - a) \prod_{j=1}^{m} (\lambda - \xi_{j})^{2} d\mu}_{R_{\text{Radau}}}, \quad \eta \in (a, b).$$

In this case we have

$$R_{\text{Radau}}^{(m+1)} = \frac{(-1)^{2m+1}(2m+2)!\lambda^{-(2m+3)}}{(2m+1)!} \cdot \int_{a}^{b} (\lambda - a) \prod_{j=1}^{m} (\lambda - \xi_j)^2 d\mu \leq 0$$

due to the fact that $(\lambda - a)$ is nonnegative because a is a lower bound of the spectrum of \mathbf{A} . Thus $I_{\text{Radau}}^{(m+1)} \geq \int_{a}^{b} f(\lambda) d\mu$ and together we have

$$I_{ ext{Gauß}}^{(m)} \leq \|oldsymbol{x}^* - ilde{oldsymbol{x}}\|_2 \leq I_{ ext{Radau}}^{(m+1)}.$$

Let us briefly recall how the nodes and weights are obtained in Gauß like quadrature rules. By use of the scalar product $\langle \phi | \psi \rangle_{\mu} := \int_{a}^{b} \phi \cdot \psi \, d\mu$, it is possible to define an orthonormal sequence of polynomials ϕ_1, ϕ_2, \ldots with $\deg(\phi_j) = j$. This set of orthonormal polynomials satisfies a three term recurrence relationship:

$$\beta_{j}\phi_{j}(\lambda) = (\lambda - \alpha_{j})\phi_{j-1}(\lambda) - \beta_{j-1}\phi_{j-2}(\lambda), \quad j = 1, \dots, m$$

$$\phi_{-1}(\lambda) \equiv 0, \quad \phi_{0}(\lambda) \equiv 1$$
(4.8)

if $\int_a^b d\mu = 1$ (by scaling \boldsymbol{r} to $\|\boldsymbol{r}\|_2 = 1$). In Matrix form this can be written as

$$(\boldsymbol{J}^{(m)} - \lambda \boldsymbol{I})\boldsymbol{\phi}(\lambda) = -\beta_m \phi_m(\lambda)\boldsymbol{e}_m$$
(4.9)

where $\boldsymbol{\phi}(\lambda) = (\phi_0(\lambda), \dots, \phi_{n-1}(\lambda))$ and

$$\boldsymbol{J}^{(m)} = \begin{pmatrix} \alpha_{1} & \beta_{1} & & & \\ \beta_{1} & \alpha_{2} & \beta_{2} & & \\ & \ddots & \ddots & \ddots & \\ & & \beta_{m-2} & \alpha_{m-1} & \beta_{m-1} \\ & & & & \beta_{m-1} & \alpha_{m} \end{pmatrix}$$

From (4.8), we can see that these α 's and β 's are exactly the same coefficients as computed by the Lanczos algorithm (Algorithm 1.2). Multiplying the recurrence relation with $\phi_{j-1}(\lambda)$ and using the orthogonality constraints we get $\langle \phi_{j-1}(\lambda) |$ $(\lambda - \alpha_j)\phi_{j-1}(\lambda)\rangle_{\mu} = 0$ and therefore

$$\alpha_{j} = \langle \phi_{j-1}(\lambda) \mid \lambda \phi_{j-1}(\lambda) \rangle_{\mu} = \int_{a}^{b} \lambda \phi_{j-1}^{2}(\lambda) d\mu$$
$$= \sum_{i=1}^{n} \lambda_{i} \phi_{j-1}^{2}(\lambda_{i}) \mu_{i}^{2} = \sum_{i=1}^{n} \phi_{j-1}(\lambda_{i}) \mu_{i} \cdot \lambda_{i} \phi_{j-1}(\lambda_{i}) \mu_{i}$$
$$= \langle \phi_{j-1}(\mathbf{A}) \mid \mathbf{A} \cdot \phi_{j-1}(\mathbf{A}) \rangle = \langle \mathbf{v}_{j-1} \mid \mathbf{A} \mathbf{v}_{j-1} \rangle.$$

Using $\|\phi_j(\boldsymbol{A})\|_2 = 1$, we get $\beta_j = \|(\boldsymbol{A} - \alpha_m \boldsymbol{I})\boldsymbol{v}_{j-1} - \beta_{j-1}\boldsymbol{v}_{j-2}\|_2$ in correspondence with Algorithm 1.2.

The eigenvalues of $J^{(m)}$ (which are the zeroes of ϕ_m) are the nodes of the Gauß quadrature rule. The weights are the squares of the first elements of the normalized eigenvectors of $J^{(m)}$.

In order to obtain the Gauß-Radau rule, we have to extend the matrix $J^{(m)}$ in such a way that it has one prescribed eigenvalue $\xi_{m+1} = a$, i.e., we wish to construct ϕ_{m+1} such that $\phi_{m+1}(a) = 0$. From the recurrence relation (4.8), we have

$$0 \stackrel{!}{=} \beta_{m+1}\phi_{m+1}(a) = (a - \alpha_{m+1})\phi_m(a) - \beta_m\phi_{m-1}(a).$$

This gives

$$\alpha_{m+1} = a - \beta_m \frac{\phi_{m-1}(a)}{\phi_m(a)}$$

and evaluating (4.9) at $\lambda = a$ yields

$$(\boldsymbol{J}^{(m)} - a\boldsymbol{I})\boldsymbol{\phi}(a) = -\beta_m \phi_m(a) \cdot \boldsymbol{e}_m \quad \Leftrightarrow \quad (\boldsymbol{J}^{(m)} - a\boldsymbol{I})\boldsymbol{\delta} = \beta_m^2 \boldsymbol{e}_m \tag{4.10}$$

with $\boldsymbol{\delta} = (\delta_1, \ldots, \delta_m)^T$ and

$$\delta_j = -\beta_m \frac{\phi_{j-1}(a)}{\phi_m(a)}, \quad j = 1, \dots, m$$

From these relations we can compute the tridiagonal matrix of the Gauß-Radau rule $\hat{J}^{(m+1)}$ by first solving the tridiagonal system (4.10) and then using the last element of $\boldsymbol{\delta}$ to define $\hat{J}^{(m+1)}$ via

Note that we need not compute the eigenvalues and eigenvectors of these tridiagonal matrices $\boldsymbol{J}^{(m)}$ and $\boldsymbol{\hat{J}}^{(m+1)}$.

Theorem 4.5 Let (ξ_1, \ldots, ξ_k) and $(\omega_1, \ldots, \omega_k)$ be the nodes and weights of an kpoint Gauß like quadrature rule and $\mathbf{J} = \operatorname{tridiag}(\beta_j, \alpha_j, \beta_{j+1})$ with α_j and β_j being the coefficients from the corresponding three term recurrence. Further let f be an analytic function, then

$$\sum_{j=1}^{m} \omega_j f(\xi_j) = \langle \boldsymbol{e}_1 \mid f(\boldsymbol{J}) \boldsymbol{e}_1 \rangle.$$

Proof: As shown for example in [126], the weights ω_i can be computed as

$$\omega_j = \left(\frac{y_{1,j}}{\phi_0(\xi_j)}\right)^2, \quad j = 1, \dots, m,$$

where $y_{1,j}$ is the first element of the *j*th eigenvector of \boldsymbol{J} . Since $\phi_0(\lambda) \equiv 1$, we have $\omega_j = (y_{1,j})^2 = (\boldsymbol{e}_1^T \boldsymbol{y}_j)^2$. Using $\boldsymbol{Y} := (\boldsymbol{y}_1 | \dots | \boldsymbol{y}_m)$ and $\boldsymbol{\Xi} = \text{diag}(\xi_1, \dots, \xi_m)$ we get

$$\sum_{j=1}^{m} \omega_j f(\xi_j) = \sum_{j=1}^{m} \boldsymbol{e}_1^T \boldsymbol{y}_j f(\xi_j) (\boldsymbol{y}_j)^T \boldsymbol{e}_1$$
$$= \left\langle \boldsymbol{e}_1 \middle| \sum_{j=1}^{m} \boldsymbol{y}_j f(\xi_j) (\boldsymbol{y}_j)^T \cdot \boldsymbol{e}_1 \right\rangle$$
$$= \left\langle \boldsymbol{e}_1 \middle| \boldsymbol{Y} f(\boldsymbol{\Xi}) \boldsymbol{Y}^T \cdot \boldsymbol{e}_1 \right\rangle$$
$$= \left\langle \boldsymbol{e}_1 \middle| f(\boldsymbol{J}) \boldsymbol{e}_1 \right\rangle.$$

Since we had to scale our initial residual vector to have norm 1, i.e., we solved

$$\left\langle \beta_0^{-1} \boldsymbol{r} \mid \boldsymbol{A}^{-2} \cdot \beta_0^{-1} \boldsymbol{r} \right\rangle = \beta_0^{-2} \cdot \left\langle \boldsymbol{r} \mid \boldsymbol{A}^{-2} \boldsymbol{r} \right\rangle = \beta_0^{-2} \cdot \| \boldsymbol{x}^* - \tilde{\boldsymbol{x}} \|_2^2,$$

with $\beta_0 = \|\boldsymbol{r}\|_2$. We now obtain lower and upper bounds as

$$egin{array}{lll} eta_0 \sqrt{\langle m{e}_1 \mid (m{J}^{(m)})^{-2} \, m{e}_1
angle} &\leq & \|m{x}^* - ilde{m{x}}\|_2 &\leq & eta_0 \sqrt{\langle m{e}_1 \mid (m{\hat{J}}^{(m+1)})^{-2} \, m{e}_1
angle} \end{array}$$

It should be remarked again that these bounds are only valid for symmetric positive definite matrices A. Of course, we can always transfer a linear system Ax = b into an equivalent system $\hat{A}x = \hat{b}$ with $\hat{A} = A^T A$ and $\hat{b} = A^T b$. Then \hat{A} is s.p.d. but cond $(\hat{A}) = \text{cond}(A)^2$. This limits the range of matrices we can handle to cond $(A) \leq \epsilon^{-1/2}$. However, this restriction is not as important as it seems to be because we have to use a more precise arithmetic anyway, as we will see in Section 4.3.3.

4.3.3 Verified Computation

Of course, the results of the two preceeding sections 4.3.1 and 4.3.2 assume that all computations are *exact* or at least valid bounds of the exact values. Since we cannot guarantee this by using floating-point arithmetic, we have to bring the tools from Section 3 into action.

In all subsequent algorithms, the variable accu represents a long accumulator in the sense of section 3.1. In particular, expressions of the form $accu = accu \pm x \cdot y$ denote exact accumulation of $x \cdot y$ in accu.

• Decomposition Error $\|\boldsymbol{L}\boldsymbol{U}^T - \boldsymbol{A}\|_2$

Suppose L, U to be a nonsingular lower triangular matrices. Then Algorithm 4.1 computes a rigorous upper bound for $||LU^T - A||_2$.

• Smallest Singular Value

Due to ideas of Rump [109], we compute the smallest singular value of a matrix A in two steps. First we factorize A approximately in a product of two triangular

$$\begin{split} e_1^{\text{row}} &:= e_2^{\text{row}} := \dots := e_n^{\text{row}} := 0\\ e_1^{\text{col}} &:= e_2^{\text{col}} := \dots := e_n^{\text{col}} := 0\\ \text{for } i &= 1, \dots, n\\ \text{for } j &= 1, \dots, i-1\\ accu &:= -a_{i,j}\\ \text{for } k &= 1, \dots, j\\ accu &= accu + l_{i,k} * u_{j,k}\\ e_i^{\text{row}} &= e_i^{\text{row}} \triangle \triangle (|accu|)\\ e_j^{\text{col}} &= e_j^{\text{col}} \triangle \triangle (|accu|)\\ accu &:= -a_{i,i}\\ \text{for } k &= 1, \dots, i\\ accu &= accu + l_{i,k} * u_{i,k}\\ e_i^{\text{row}} &= e_i^{\text{row}} \triangle \triangle (|accu|)\\ e_i^{\text{col}} &= e_i^{\text{col}} \triangle \triangle (|accu|)\\ e_i^{\text{col}} &= e_i^{\text{col}} \triangle \triangle (|accu|)\\ e_i^{\text{col}} &= e_i^{\text{col}} \triangle \triangle (|accu|)\\ e_{\max}^{\text{row}} &:= \max\{e_1^{\text{row}}, e_2^{\text{row}}, \dots, e_n^{\text{row}}\}\\ e_{\max}^{\text{col}} &:= \max\{e_1^{\text{col}}, e_2^{\text{col}}, \dots, e_n^{\text{col}}\}\\ \text{return } \triangle \sqrt{e_{\max}^{\text{row}} \triangle e_{\max}^{\text{col}}} \end{split}$$

Algorithm 4.1: Compute a verified upper bound for $\|\boldsymbol{L}\boldsymbol{U}^T - \boldsymbol{A}\|_2$ via the inequality $\|\boldsymbol{B}\|_2 \leq \sqrt{\|\boldsymbol{B}\|_1 \cdot \|\boldsymbol{B}\|_{\infty}}$.

matrices, say T_1 and T_2 (compare Section 1.4.2) and then we compute a lower bound of $\sigma_{\min}(T_1T_2^T)$ via¹ $\sigma_{\min}(T_1T_2^T) \geq \sigma_{\min}(T_1)\sigma_{\min}(T_2)$.

Now we have to compute the smallest singular values of these triangular matrices. The basic idea is first to compute an approximation $\tilde{\sigma} \approx \sigma_{\min}(\mathbf{T})$ and then proving that $\mathbf{T}\mathbf{T}^T - \kappa \tilde{\sigma}^2 \mathbf{I}$ is positive semidefinite, where κ is slightly less than one. In case of success, $\sqrt{\kappa}\tilde{\sigma}$ is a lower bound of $\sigma_{\min}(\mathbf{T})$. To decide whether the shifted $\mathbf{T}\mathbf{T}^T$ remains positive semidefinite, we try to compute its Cholesky factorization $\mathbf{L}\mathbf{L}^T$. Since this decomposition is usually not exact, we have to apply the following theorem from Wilkinson to guarantee that $\mathbf{L}\mathbf{L}^T$, if it exists, is not too far from $\mathbf{T}\mathbf{T}^T - \kappa \tilde{\sigma}^2 \mathbf{I}$ so that the positive definiteness of $\mathbf{L}\mathbf{L}^T$ is sufficient for the smallest eigenvalue of $\mathbf{T}\mathbf{T}^T - \kappa \tilde{\sigma}^2 \mathbf{I}$ to be nonnegative.

Theorem 4.6 Let $\boldsymbol{B}, \ \tilde{\boldsymbol{B}} \in \mathbb{R}^{n \times n}$ be symmetric and $\lambda_i(\boldsymbol{B})_{i=1}^n$, respectively $\lambda_i(\tilde{\boldsymbol{B}})_{i=1}^n$ be the eigenvalues ordered by magnitude.

Then from $\|\boldsymbol{B} - \tilde{\boldsymbol{B}}\|_{\infty} \leq d$ it follows that $|\lambda_i(\boldsymbol{B}) - \lambda_i(\tilde{\boldsymbol{B}})| \leq d$.

See [142] for a proof.

That is, if $\|\boldsymbol{L}\boldsymbol{L}^{T} - (\boldsymbol{T}\boldsymbol{T}^{T} - \kappa\tilde{\sigma}^{2}\boldsymbol{I})\|_{\infty} \leq d$ then $\sigma_{\min}(\boldsymbol{T})^{2} = \lambda_{\min}(\boldsymbol{T}\boldsymbol{T}^{T}) \geq \kappa\tilde{\sigma}^{2} - d$. Thus, if d is a verified upper bound for $\|\boldsymbol{B} - \tilde{\boldsymbol{B}}\|_{\infty}$ and $\kappa\tilde{\sigma}^{2} \geq d$ we have $\sigma_{\min}(\boldsymbol{T}) \geq \sqrt{\kappa\tilde{\sigma}^{2} - d}$.

¹In this computation of the smallest singular value hides the $O(n^3)$ effort which seems to be necessary to compute error bounds [27].

• Verifying Positive Definiteness of $TT^T - \sigma^2 I$

Suppose T to be a nonsingular lower triangular matrix. Then Algorithm 4.2 computes a rigorous lower bound for its smallest singular value.

```
\sigma = 0.9 * approx_smallest_singular_value(\mathbf{T})
start:
e_1 := e_2 := \ldots := e_n := 0
for i = 1, ..., n
   for j = 1, ..., i - 1
       accu := 0
      for k = 1, ..., j
          accu = accu + t_{i,k} * t_{j,k}
      for k = 1, ..., j - 1
          accu = accu - l_{i,k} * l_{j,k}
      l_{i,j} = \Box(accu)/l_{j,j}
      e_i = e_i \triangle \triangle (|accu - l_{i,j} * l_{j,i}|)
      e_{i} = e_{i} \triangle \triangle (|accu - l_{i,j} * l_{j,i}|)
   accu := -\sigma^2
   for k = 1, ..., i
       accu = accu + t_{i,k} * t_{i,k}
   for k = 1, ..., i
       accu = accu - l_{i,k} * l_{i,k}
   if accu < 0
      if iter < max_iter
          \sigma = 0.9 * \sigma
          iter = iter + 1
          goto start
      else
          return failed
   l_{i,i} = \sqrt{\Box(accu)}
   e_i = e_i \triangle \triangle (|accu - l_{i,i}^2|)
e_{\max} := \max\{e_1, e_2, \dots, e_n\}
if \sigma \nabla \sigma \ge e_{\max}
   return \sqrt[]{\sigma \nabla \sigma \nabla e_{\max}}
else
   return failed
```

```
Algorithm 4.2: Compute a verified lower bound \sigma \leq \sigma_{\min}(\mathbf{T}) and a lower triangular matrix \mathbf{L} with \mathbf{L}\mathbf{L}^T - (\mathbf{T}\mathbf{T}^T - \sigma^2 \mathbf{I}) \leq e_{\max}.
```

• Recursion Coefficients of the Gauß Quadrature Rule

Given an approximate solution \boldsymbol{x} we have to compute an enclosure of the residual $\tilde{\boldsymbol{v}}_0$ as described in Section 3.4. We then start a straightforward interval formulation

of the Lanczos algorithm (see Algorithm 4.3) to get enclosures of the α 's and β 's and thus for $J^{(m)}$.

In the next section we will see that it is also possible to get an enclosure for the solution of equation (4.10) on page 62, i.e., an enclosure for δ_m and therefore we are also able to compute $[\hat{J}]^{(m+1)}$.

Given
$$[\tilde{\boldsymbol{v}}]_0$$
, e.g. $[\tilde{\boldsymbol{v}}]_0 = \diamondsuit (\boldsymbol{b} - \boldsymbol{A}\boldsymbol{x}_0)$
 $[\beta]_0 = \|[\tilde{\boldsymbol{v}}]_0\|_2$
 $[\boldsymbol{v}]_0 = [\beta]_0^{-1} \cdot [\tilde{\boldsymbol{v}}]_0$
for $m = 1, 2, \dots$
 $[\alpha]_m = \langle [\boldsymbol{v}]_{m-1} \mid \boldsymbol{A}[\boldsymbol{v}]_{m-1} \rangle$
if $m = 1$
 $[\tilde{\boldsymbol{v}}]_m = (\boldsymbol{A} - [\alpha]_m \boldsymbol{I})[\boldsymbol{v}]_{m-1}$
else
 $[\tilde{\boldsymbol{v}}]_m = (\boldsymbol{A} - [\alpha]_m \boldsymbol{I})[\boldsymbol{v}]_{m-1} - [\beta]_{m-1}[\boldsymbol{v}]_{m-2}$
 $[\beta]_m = \|[\tilde{\boldsymbol{v}}]_m\|_2$
 $[\boldsymbol{v}]_m = [\beta]_m^{-1} \cdot [\tilde{\boldsymbol{v}}]_m$

Algorithm 4.3: Interval Lanczos-algorithm.

• Solutions of Symmetric Tridiagonal Systems

A well known technique for computing solutions for interval linear systems is the interval Gauß algorithm (IGA). The shape of the solution set of a interval linear system can be fairly complicated, but since we use interval arithmetic we are only able to compute a multidimensional box that contains the true solution set. For general matrices it cannot be guaranteed to get a solution box that is near to the smallest box containing the true solution but it can be shown (see [38]) that the IGA produces optimal results, i.e. smallest in diameter, for tridiagonal interval systems with system matrices $[\boldsymbol{J}]^{(m)}, [\boldsymbol{J}]^{(m)} - a\boldsymbol{I}$, or $[\hat{\boldsymbol{J}}]^{(m+1)}$ respectively. However, we only sketch the algorithm here (Algorithm 4.4).

$$\begin{split} & [c]_1 = [\alpha]_1 \\ & [e]_1 = [b]_1 \\ & \mathbf{for} \ i = 2, \dots, m \\ & [c]_i = [\alpha]_i - [\beta]_{i-1}[\beta]_{i-1}/[c]_{i-1} \\ & [e]_i = [b]_i - [\beta]_{i-1}[e]_{i-1}/[c]_{i-1} \\ & [x]_m = [e]_m/[c]_m \\ & \mathbf{for} \ i = m - 1, \dots, 1 \\ & [x]_i = ([e]_i - [\beta]_i[x]_{i+1})/[c]_i \end{split}$$

Algorithm 4.4: Interval Gauß algorithm for tridiag($[\beta]_i, [\alpha]_i, [\beta]_{i+1}$) $\cdot [x] = [b]$.

CHAPTER

5

High Performance Object Oriented Numerical Linear Algebra

⁶⁶ I always knew C++ templates were the work of the Devil, and now I'm sure.

Cliff Click, 1994

⁶⁶ My desire to inflict pain on the compilers is large. They've been tormenting me for the last 8 years. Now is my chance to strike back!

Scott Haney, 1996

An often used prejudice against modern object oriented programming techniques, is that object orientation is almost equivalent to low performance. At a first glance, if one compares the speed of a simple routine once written in, e.g., Fortran and once naively written in C++, this proposition seems to be true. Object oriented programming is massively based on abstraction, encapsulation of data, access restriction, and polymorphism. Most of these features imply a big organizational overhead because many decisions have to be done at runtime like dereferencing of polymorphic types or allocation and deallocation of temporary variables. Besides these runtime penalties, also the compiler is unable to optimize around virtual function calls which prevents instruction scheduling, data flow analysis, loop unrolling, etc.

Thus, the situation — at least at the beginning of object oriented programming — was that though we had really nice, well readable, and excellent maintainable code, we had to pay with relatively low performance. Therefore and because most of the existent programs where coded in old Fortran versions, the scientific computing community decided to stay in the *stone age* of software technique.

Meanwhile, started in the middle of the 1990's, there have been made several proposals to improve the performance of object oriented programs. Using these techniques, it is possible to write highly abstract, object oriented programs that are comparable in speed with Fortran or C and are sometimes even faster [79, 139]. These improvements can be roughly splitted into two categories.

- The first kind tries to use the object orientation itself to remove redundancies, reduce code size and separate conceptually non-coupled program units. This isolation of performance critical code sections actually enables writing portable high performance codes. The key to this kind of structured programming is called *genericity*. We describe some of the most important concepts in Section 5.1.
- The second category aims to reduce the organizational overhead by relocating performance critical parts from run-time execution to compile-time execution. This technique can be viewed as a code generation system that removes, e.g., virtual function calls which are essentially required by polymorphic types. This compile-time polymorphism is called *static polymorphism* and has much more favorable optimization properties. The key technique is called *compile-time programming* and we describe some aspects in Section 5.2.

5.1 Genericity

The traditional approach writing basic linear algebra routines is a combinatorial affair. There are typically four precision types that need to be handled (single and double precision real, single and double precision complex), several dense storage types, a multitude of sparse storage types (the Sparse BLAS Standard Proposal includes 13 different sparse storage types [125]), as well as row and column orientations for each matrix type. On top of that, if one wants to parallelize these codes, there are several data distributions to be supported. To provide a full implementation one might need to code literally hundreds of versions of the same routine! It is no wonder the NIST implementation of the Sparse BLAS contains over 10 000 routines and an automatic code generation system [113].

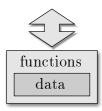
This combinatorial explosion arises because with most programming languages, algorithms and data structures are more tightly coupled than is conceptually necessary. That is, one cannot express an algorithm as a subroutine independently from the type of data that is being operated on. Thus, although abstractly one might have only a single algorithm to be expressed, it must be realized separately for every data type that is to be supported. As a result, providing a comprehensive linear algebra library — much less one that also offers high-performance — would seem to be an impossible task.

Fortunately, certain modern programming languages, such as Ada and C++, provide support for generic programming, a technique whereby an algorithm can be expressed independently of the data structure to which it is being applied. One of the most celebrated examples of generic programming is the C++ Standard Template Library (STL). Especially for numerical linear algebra, there is a library called the Matrix Template Library (MTL) which extends this generic programming approach to cover the needs of scientific computing [118].

The principal idea behind genericity is that many algorithms can be abstracted away from the particular data structures on which they operate. Algorithms typically need the abstract functionality of traversing through a data structure and accessing its elements. If data structures provide a standard interface for traversal and access, generic algorithms can be freely mixed and matched with data structures [117].

5.1.1 Data Structures: Containers

In object oriented numerics (OON), data structures like **records** in Pascal, **TYPEs** in Fortran or **structs** in C together with a set of functions operating on this data are called containers. In C++, the equivalent to a container is called **class**. Containers basically consist of two parts: an internal representation which is only directly accessible by a set of authorized functions and a public interface which provides functions and methods to access the encapsulated data.



Let us consider the following example to demonstrate the basic ideas of containers. Suppose we want to design the concept of matrices. First we have to think about the storage types we wish to support. Assume we need

- banded matrices, i.e., we only store the diagonals between lower_bandw and upper_bandw and implicitly define the remaining elements to be zero,
- general sparse matrices, i.e., all nonzero elements are stored in a list with elements of type (row, col, value), and
- dense matrices.

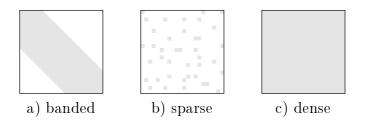


Figure 5.1: The storage types for our matrices

In this example, the internal representation might be

- a set of vectors storing the used diagonals (banded matrices),
- a list of triples (row, col, value) storing all nonzero elements (general sparse matrices), or
- a num_rows times num_cols sized memory block (dense matrices).

However, in the public interface we only need functions like get_num_rows() or get_lower_bandwith() to get informations about the shape of the matrices and most important we need access to the matrix entries. Traditionally only simple access functions like get_entry_at(row, col) were provided.

With this technique we have two possibilities to write algorithms which need access to the matrix entries: One is to traverse through all num_rows times num_cols elements (although most of them are zero) which is very slow. The other is to provide special algorithms with exact knowledge of the sparsity pattern for each matrix type which results in an enormous amount of code and therefore in an enormous amount of errors. Additionally, if we add a new storage type or modify an existing one, we have to add/modify a complete set of algorithms (compare Figure 5.2, left).

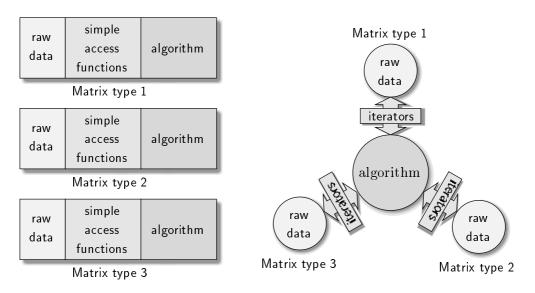


Figure 5.2: Separating the raw data structures from algorithms by the use of generic access and traversal functions — so called iterators.

Fortunately we can do much better by providing generic access via so called Iterators as will be described in the following section.

5.1.2 Traversing and Accessing Elements: Iterators

Instead of accessing the matrix data element-wise we conceptually design our matrices to be containers of containers (although they are typically not actual implemented in this way). For example we interprete a matrix to be a column vector with row vectors as element type (or vice versa). These vector containers, which haven't allocated any storage but only refer to the memory of their parent matrix are called iterators. Designing these Iterators to have a common public interface, enables us to use only one routine for each algorithm (see Figure 5.2, right).

Thus, iterators can be used to traverse through a matrix in the following sense. Each matrix container provides a datatype called, e.g., row_2Dlterator (line 3) and two functions returning special row_2Dlterator's, one is called begin() (line 4) and one is called end() (line 5) returning the first respectively last row vector of the matrix.

```
1 class Matrix {
2 public:
3 typedef row_2DIterator my_row_iterator;
4 row_2DIterator begin();
5 row_2DIterator end();
6 ...
7 };
```

The row_2DIterator itself — which in this case is implemented as a my_row_iterator — provides the datatype 1DIterator (line 10) and again the two functions begin() (line 11) and end() (line 12) here returning the first respectively last element of the according row. Additionally, it provides the function next() (line 13) which returns the successor of the row from which it is called.

```
8 class my_row_iterator {
9 public:
10 typedef 1DIterator my_element_iterator;
11 1DIterator begin();
12 1DIterator end();
13 my_row_iterator next();
14 int get_index();
15 ...
16 };
```

With this concept, we can ask the Matrix for the begin() row_2DIterator to get its first row-vector. With the next() function we can traverse through all rows until we end up with end().

Beside these functions for traversing through the data structure we usually have functions for accessing the stored information. In this case we have get_index() which returns the row number (line 14).

Finally we have the 1DIterator which doesn't refer to any subsequent iterator. Here we only have one traversal function next() and two access functions get_index() and get_value().

```
17 class my_element_iterator {
18 public:
19 my_vector_iterator next();
20 int get_index();
21 double get_value();
22 ...
23 };
```

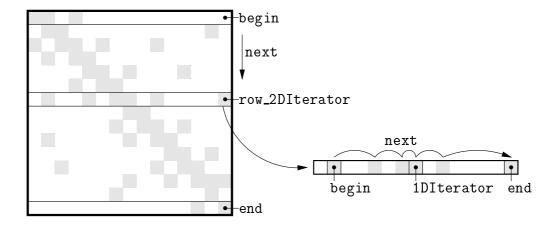


Figure 5.3: Traversing and accessing elements via iterators

```
1 void print( Matrix A )
 2 {
 3
     A::row_2DIterator
                               current_row;
 4
     current_row::1DIterator current_entry;
 5
     current_row = A.begin();
 6
 7
     do {
 8
       current_entry = current_row.begin()
 9
       do {
10
         cout << "[" << current_row.get_index()</pre>
                                                     << ","
                      << current_entry.get_index() << "] = "
11
                      << current_entry.get_value() << endl;
12
         current_entry = current_entry.next();
13
14
       } while( current_entry!=current_row.end() );
       current_row = current_row.next();
15
16
     } while( current_row!=A.end() );
17 }
```

Algorithm 5.1: A generic routine for printing arbitrary matrices.

In the same way we iterate through the rows of the matrix. With the 1DIterator it is now possible to traverse through the elements of each row_2DIterator to access the row and column number and value of each nonzero entry, compare Figure 5.3.

In Algorithm 5.1 we illustrate this technique with the simple example for printing an arbitrary matrix. In line 3 we declare the variable current_row to be of type A::row_2DIterator, that is, A's row_2DIterator type and in line 4 we declare current_entry to be a 1DIterator of current_row. In line 6 we set current_row to be the first row of A and then iterate in the outer do-while-loop until current_row reaches the last row of A (line 16). For each current_row we set current_entry to be the first entry and iterate until current_entry reaches the last entry of current_row. Meanwhile we print current_row.get_index() (line 10) which is the row index, current_entry.get_index() (line 11): the column index and finally current_entry.get_value() (line 12): the value of the current entry. This print() routine is capable to print all matrix types which provides these coupled 2D/1D iterator set. That is, each new matrix type only has to provide iterators with the same interface as row_2DIterator and 1DIterator to be automatically printable via this function. For example, if we provide a matrix container with a row_2DIterator that actually represents the columns, we immediately can print the transposed of this matrix.

Similar to this print function, we can provide algorithms to multiply a matrix with a vector, to multiply two matrices, to add or copy matrices, and so on, only by writing one function for each algorithm.

5.1.3 A Point of View

Generalizing the above idea of printing a transposed matrix leads to so called *views* or *adaptors*. A view is a special container that actually has no own storage allocated but only refers to the memory of a *legal* container. The difference is, that a view provides a modified set of iterators. For example we could define a transposed view of a matrix via

```
1 class transposedMatrix : public Matrix {
2 public:
3 typedef row_2DIterator Matrix::col_2DIterator;
4 ...
5 };
```

From line 1 we see that a transposedMatrix is derived from a Matrix but exports its row_2DIterator as Matrix's col_2DIterator. Consequently, if we access the rows of a transposedMatrix we actually get the columns of the original Matrix.

With this technique we can, e.g., also implement sub-matrix views by modifying the **begin()** and **end()** functions or diagonal views by modifying the **next()** function to return the next element on a given diagonal. All these operations are O(1), i.e., they need constant time and (nearly) no storage.

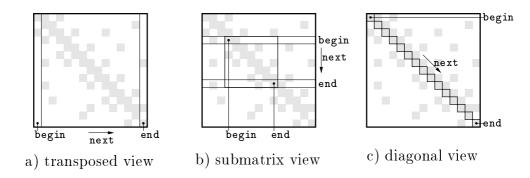


Figure 5.4: Some views of a matrix

5.2 Two-Stage Programming

In this section we describe some advanced techniques to improve the performance of object oriented code [138]. Although they are hard to write and maintain and increase compile times, they are great for library designers. Many of theses problems can be hidden from the end user. All ideas presented here, are based on a two-stage programming process. That is, beside the usual run-time execution, some parts of the program are evaluated at compile-time and thereby, e.g., generate specialized problem oriented code without polymorphism and thus without virtual function calls, or expressions without temporary variables, or even explicitly unrolled loops, etc. In C++ this technique was enabled by introducing so called *templates*, which makes it possible to write programs in a subset of C++ which are interpreted at compile-time. In this section we first introduce some basic concepts of compile time programming and then present two very powerful techniques to improve the performance of object oriented programs: namely expression templates and automatic self optimization.

5.2.1 Compile Time Programming

To demonstrate the power of template meta programming, i.e., routines that are evaluated completely by the compiler, we start with an example taken from [136].

```
template<int N_factorial>
class Value {};

template<int N>
class Factorial {
  public:
    enum { value = N * Factorial<N-1>::value };
  };

class Factorial<1> {
  public:
    enum { value = 1 };
  };

void foo()
 {
    Value<Factorial<6>::value> dummy = Factorial<6>();
 }
```

Using this Factorial-class, the value N! (factorial of N) is available at compiletime as Factorial<N>::value. How does it work? When Factorial<N> is instantiated, the compiler needs Factorial<N-1> in order to assign the enum value. So it instantiates Factorial<N-1>, which in turn requires Factorial<N-2>, requiring Factorial<N-3>, and so on until Factorial<1> is reached, where template specialization is used to end the recursion. The compiler effectively performs a for loop to evaluate N! at compile-time.

Thus compiling the above C++ program with gcc factorial.cc one gets the compiler error:

 Although this technique might seem like just a cute C++ trick, it becomes powerful when combined with normal C++ code. In this hybrid approach, source code contains two programs: the normal C++ run-time program, and a template metaprogram which runs at compile-time. Template metaprograms can generate useful code when interpreted by the compiler, for example massively in-lined algorithm — such as an implementation of an algorithm which works for a specific input size, and has its loops unrolled. This results in large speed increases for many applications.

There are template-meta-program equivalents for most C++ flow control structures like 'if/else if/else', 'for', 'do/while', 'switch', or subroutine calls. See Tables 5.1 and 5.2 for some examples [136]. Of course these compile-time versions aren't very handy but we don't intend to write entire programs in this way. However, this kind of programming is worth the work in highly critical parts of an algorithm, e.g., in inner loops. Here we can partially unroll loops or reverse the order of computations to exploit memory caches or pipeline facilities (see Section 5.2.2).

C++ version	Template metaprogram version
<pre>if (condition) { statement1; } else { statement2; }</pre>	<pre>// Class declarations: template<bool c=""> class _if {}; class _if<true> { public: static inline void _then() { statement1; } };</true></bool></pre>
	<pre>class _if<false> { public: static inline void _then() { statement2; } }; // Replacement for 'if/else' statement: _if<condition>::_then();</condition></false></pre>

Table 5.1: A C++ if/else structure and its template metaprogram equivalent.

Another valuable field for using templates is to avoid run-time polymorphism. Let us first briefly describe this concept by using again our matrix example from the beginning of Section 5.1.1. Suppose we want to implement the simple access function get_entry_at(row, col).

C++ version	Template metaprogram version		
<pre>int i = N;</pre>	<pre>// Class declarations: template<int i=""></int></pre>		
<pre>do { statement(i);</pre>	class _do_from { public:		
i; } while (i>0);	<pre>static inline void _downto_1() { statement(I);</pre>		
	_do_from <i-1>::_downto_1(); }</i-1>		
	};		
	<pre>class _do_from<0> { public:</pre>		
	<pre>static inline void _downto_1() {} };</pre>		
	<pre>// Replacement for 'do/while' statement: _do_from<n>::_downto_1();</n></pre>		

Table 5.2: A C++ do/while structure and its template metaprogram equivalent.

```
class Matrix {
public:
 virtual double get_entry_at( int row, int col ) = 0;
};
class BandedMatrix : public Matrix {
public:
 virtual double get_entry_at( int row, int col );
};
class SparseMatrix : public Matrix {
public:
 virtual double get_entry_at( int row, int col );
};
class DenseMatrix : public Matrix {
public:
 virtual double get_entry_at( int row, int col );
};
```

Here we defined a polymorphic type Matrix because a Matrix can represent several specialized matrix types. Writing, e.g., a maximum norm function

```
double max_norm( const Matrix* A ) {
  double maxnorm = 0.0,
      rowsum;
  for( int row=0; row<A->num_rows(); ++row ) {
    rowsum = 0.0;
    for( int col=0; col<A->num_cols(); ++col )
      rowsum += abs( A->get_entry_at( row, col ) );
    maxnorm = max( maxnorm, rowsum );
  }
  return maxnorm;
}
Matrix* B = new SparseMatrix;
max_norm( B );
```

causes the run-time system to decide which particular get_entry_at(row, col) is to be called every time when a Matrix::get_entry_at(row, col) is requested. That is num_rows×num_cols times it has to be figured out to which particular matrix type A points to. This will ruin the performance of any matrix algorithm!

One way to replace this run-time polymorphism by a static polymorphism is to use structure parameters which encapsulate particular storage information:

```
class BandedMatrix {
   // Storage information for banded matrices
};
....
template<class T_structure>
class Matrix {
   private:
    T_structure _data;
};
template<class T_structure>
double max_norm( Matrix<T_structure>& A ) { ... }
Matrix<BandedMatrix> B;
max_norm( B );
```

Here, the Matrix type is quasi polymorphic but only up to compilation. After interpreting the template metaprogram part, Matrix<BandedMatrix> is a static type and every call to one of its functions (e.g., max_norm) is non-virtual. The disadvantages of this solution are

- Matrix has to constantly delegate operations to the structure objects.
- The interface between the Matrix and the T_structure object must be identical for all structures.
- Interfaces must expand to accommodate every supported matrix structure. For example if we need banded matrices, every matrix type must provide a get_lower_bandwith() function.

Another approach to avoid virtual function calls are so called curiously recursive template patterns. Here we have a base class (Matrix) with a derived class (e.g., SparseMatrix) as template parameter. Inside the base class we have a function that explicitly converts the base class itself to be of the derived class type (lines 4-6). Each function simply delegates its execution to the corresponding leaf class by changing its type to T_leafType and calling an appropriate function of this leaf class (compare lines 7-9). Since, e.g., the code for a function Matrix<T_leafType>::get_lower_bandw() is only generated at compile-time if it is actually needed in the program, there is no need to write meaningless functions such as DenseMatrix::get_lower_bandwith() which would be called at runtime by Matrix<DenseMatrix>::get_lower_bandwith().

```
1 template<class T_leafType>
 2 class Matrix {
 3 public:
 4
     T_leafType& asLeaf() {
 5
       return static_cast<T_leafType&>( *this );
 6
     }
 7
     double get_entry_at( int row, int col ) {
       return asLeaf().get_entry_at( row, col );
 8
     }
 9
10 };
11
12 class SparseMatrix : public Matrix<SparseMatrix> {
     double get_entry_at( int row, int col );
13
14 };
15
16 ...
17
18 template<class T_leafType>
19 double max_norm( Matrix<T_leafType>& A ) { ... }
20
21 Matrix<SparseMatrix> B;
22 max_norm( B );
```

5.2.2 Self Optimization

The bane of portable high performance numerical linear algebra is the need to tailor key routines to specific execution environments. For example, to obtain high performance on a modern microprocessor, an algorithm must properly exploit the associated memory hierarchy and pipeline architecture (typically through careful loop blocking and structuring). Ideally, one would like to be able to express high performance algorithms in a portable fashion, but there is not enough expressiveness in languages such as C or Fortran to do so. Recent efforts [29] have resorted to going outside the language, i.e., to code generation systems in order to gain this kind of flexibility. Another approach is the Basic Linear Algebra Instruction Set (BLAIS) [116], a library specification that takes advantage of certain features of the C++ language to express high-performance loop structures that can be easily reconfigured for a particular architecture. To demonstrate the ability of C++ in generating environment dependent code, we give an example of a scalar product routine with partially unrolled loops. These blocks of unrolled loops could for example be distributed to different processors in a multiprocessor environment. In order not to complicate our example more than necessary, we assume that BlockSize divides Size (the size of the vectors to be multiplied). Furthermore we do not do anything special with the unrolled blocks than simply adding the results.

```
1 template<int N>
 2 class cnt {};
 3
 4 template<int Size, int BlockSize, class InIter1, class InIter2,
            class Out>
 5
 6 inline Out scalp( InIter1 x, InIter2 y, Out,
                     cnt<Size>, cnt<BlockSize> ) {
 7
     return scalp( x, y, Out(), cnt<BlockSize>(), cnt<1>() )
 8
 9
          + scalp( x+BlockSize, y+BlockSize, Out(),
                   cnt<Size-BlockSize>, cnt<BlockSize> );
10
11 }
12
13 template<class InIter1, class InIter2, class Out>
14 inline Out scalp( InIter1 x, InIter2 y, Out, cnt<1>, cnt<1>) {
15
     return *x * *y;
16 }
17
18 template<int BlockSize, class InIter1, class InIter2, class Out>
19 inline Out scalp( InIter1 x, InIter2 y, Out,
20
                     cnt<0>, cnt<BlockSize> ) {
21
     return 0.0;
22 }
```

The counter class cnt (line 1,2) is used to encapsulate integers. In line 4 the template parameters Size, BlockSize, InIter1/2 (used to iterate through the input vectors), and Out (type of the result) are introduced. Then we define the function scalp with result type Out. This function recursively computes the scalar product of the first block and that of the remaining blocks (lines 8 and 9). Two template specializations are used to terminate the recursion. The first actually computes scalar products with BlockSize=1 (lines 13-16) and the second returns 0.0 if we are outside the vector range (lines 18-22). All functions are in-lined, i.e., the entire scalar product finally appears as one block of code without function calls.

Figure 5.5 illustrates what happens at compile-time. Each step represents a recursion depth when all possible recursion calls are executed. In step 6 we see the final code, which is generated by the template metaprogram. Note: *(z+i) denotes the dereferenciation of the iterator z at position i, i.e., the *ith* entry of the vector z.

5.2.3 Expression Templates

Expression templates are a C++ technique for evaluating vector and matrix expressions in a single pass without temporaries. This technique can also be used

<pre>double x[6], y[6]; scalp(x, y, double(), cnt<6>(), cnt<2>());</pre>		step 1
<pre>scalp(x, y, double(), cnt<2>(), cnt<1>()) + scalp(x+2, y+2, double(), cnt<4>(), cnt<2>());</pre>]•	step 2
<pre>scalp(x, y, double(), cnt<1>(), cnt<1>()) + scalp(x+1, y+1, double(), cnt<1>(), cnt<1>()) + scalp(x+2, y+2, double(), cnt<2>(), cnt<1>()) + scalp(x+4, y+4, double(), cnt<2>(), cnt<2>());</pre>		step 3
<pre>*x * *y + *(x+1) * *(y+1) + scalp(x+2, y+2, double(), cnt<1>(), cnt<1>()) + scalp(x+3, y+3, double(), cnt<1>(), cnt<1>()) + scalp(x+4, y+4, double(), cnt<2>(), cnt<1>()) + scalp(x+6, y+6, double(), cnt<0>(), cnt<2>());</pre>		step 4
<pre>*x * *y + *(x+1) * *(y+1) + *(x+2) * *(y+2) + *(x+3) * *(y+3) + scalp(x+4, y+4, double(), cnt<1>(), cnt<1>()) + scalp(x+5, y+5, double(), cnt<1>(), cnt<1>()) + 0.0;</pre>		step 5
<pre>*x * *y + *(x+1) * *(y+1) + *(x+2) * *(y+2) + *(x+3) * *(y+3) + *(x+4) * *(y+4) + *(x+5) * *(y+5);</pre>		step 6

Figure 5.5: In-lining a recursively defined blocked scalar product.

for passing expressions as function arguments.. The expression can be inlined into the function body, which results in faster and more convenient code than C-style callback functions. In benchmark results, one compiler evaluates vector expressions at 95-99% efficiency of hand-coded C using this technique (for long vectors). The speed is 2-15 times that of a conventional C++ vector class, see [137].

Expression templates solve the pairwise evaluation problem associated with operator-overloaded array expressions in C++. A naive implementation of

```
Vector<double> a, b, c, d;
a = b + c + d;
```

results in:

```
double* _t1 = new double[N];
for (int i=0; i<N; ++i )
   _t1[i] = b[i] + c[i];
double* _t2 = new double[N];
for (int i=0; i<N; ++i )</pre>
```

```
_t2[i] = _t1[i] + d[i];
for (int i=0; i<N; ++i)
    a[i] = _t2[i];
delete [] _t2;
delete [] _t1;
```

For small arrays, the overhead of **new** and **delete** results in very poor performance: about 1/10th that of C. For medium (in-cache) arrays, the overhead of extra loops and cache memory accesses hurts (by about 30-50% for small expressions). The extra data required by temporaries cause the problem to go out-of-cache sooner.

For large arrays, the cost is in the temporaries: all that extra data has to be shipped between main memory and cache. Typically scientific codes are limited by memory bandwidth (rather than flops), so this really hurts. For N distinct array operands and M operators, the performance is about

$$\frac{N+1}{3M}$$

that of C/Fortran. This is particularly bad for stencils, which have N = 1 (or otherwise very small) and M very large. It is not unusual to get 1/9 (5-point stencil — first order discretization on a 2D mesh), or even 1/24 (second order discretization on a 2D mesh) the performance of C/Fortran for big stencils.

To avoid these temporary variables we have do delay the evaluation up to the point where the entire expression is parsed. That is, we have to parse the expression ourself and afterwards evaluate this parse tree in one pass. Fortunately this can be done inside C++. Let us shortly introduce the needed tools.

A class can take itself as a template parameter. This makes it possible to build linear lists or trees in the following sense:

```
template<class T1, class T2>
class X {};
```

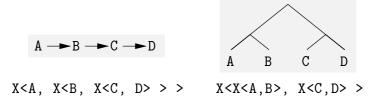


Figure 5.6: Representing lists and trees with recursive template patterns.

The basic idea behind expression templates is to use operator overloading to build parse trees. For example:

Array A, B, C, D; D = A + B + C; The expression A+B+C could be represented by a type such as

Op<plus, Op<plus, Array, Array>, Array>

Building such types is not hard:

```
template<class T>
Op<plus, T, Array> operator+(T, Array)
{
   return Op<plus, T, Array>();
}
```

Then:

```
D = A + B + C;
= Op<plus, Array, Array>() + C;
= Op<plus, Op<plus, Array, Array>, Array>();
```

Of course to be useful we need to store data in the parse tree (e.g. pointers to the arrays). Here is a minimal expression templates implementation for 1-D arrays. First, the **plus** function object:

```
class plus {
public:
   static double apply( double a, double b ) {
    return a+b;
   }
};
```

This class only provides the function apply which will be called to evaluate an object of type Op<plus, Array, Array>().

The parse tree node:

```
template<class T_op, class T1, class T2 >
class Op {
public:
    Op( T1 a, T2 b ) {
        leftNode_ = a;
        rightNode_ = b;
    }
    double operator[]( int i ) {
        return T_op::apply( leftNode_[i], rightNode_[i] );
    }
private:
    T1 leftNode_;
    T2 rightNode_;
};
```

Here we have a constructor which simply stores its arguments in the private variables leftNode_ and rightNode_. Additionally, the class Op provides an index operator which applies the operator T_op to the node-data and returns its result.

Now a simple array class:



```
class Array {
public:
  Array( double* data, int size ) {
    data_ = data;
    size_ = size;
  }
  template<class T_op, class T1, class T2 >
  operator=( Op<T_op, T1, T2> expr ) {
    for( int i=0; i<size_; ++i )</pre>
      data_{[i]} = expr[i];
  }
 double operator[]( int i ) {
    return data_[i];
  }
private:
 double* data_;
  int size_;
};
```

The Array constructor stores a pointer to the data and the number of elements. The operator= assigns the values of an expression (represented by its parse tree) to data_. Note that calling the index operator of expr actually causes the evaluation of the entire parse tree in one pass. Additionally Array itself provides an index operator which behaves traditionally, i.e., it simply returns the array entries.

And finally the **operator+** which actually does not add anything but constructs the parse tree:

```
template<class T>
  Op<plus, T, Array> operator+( T a, Array b ) {
    return Op<plus, T, Array>(a,b);
  }
Now see it in action:
  int main() {
    double a.data[] = { 2, 3, 5, 9 },
            b.data[] = \{ 1, 0, 0, 1 \},\
            c.data[] = \{ 3, 0, 2, 5 \},
            d.data[4];
    Array A(a.data,4),
           B(b.data,4),
           C(c.data,4),
           D(d.data,4);
    D = A + B + C;
    for (int i=0; i < 4; ++i)
      cout << D[i] << " ";</pre>
    cout << endl;</pre>
```

```
return 0;
}
Output: 6 3 7 15
See how operator+ builds up the parse tree step by step:
D = A + B + C;
= Op<plus, Array, Array>(A, B) + C;
= Op<plus, Op<plus, Array, Array>, Array>
( Op<plus, Array, Array>(A, B), C);
Then it matches to template Array::operator=:
```

See how expr[i] is evaluated successively by Op<T_Op, Array, Array>::operator[] calling T_op::apply():

... more or less. It's all clear now, right?

• Excursion: Exact Scalar Product

The exact scalar product, as introduced in Section 3.1, is often useful in critical computations. Computing residual vectors, for example, highly suffers from cancellation errors because many large numbers are added to a (hopefully) small sum. This error source can be completely avoided using exact scalar products. However, simulated in software, this routine is relatively slow compared to the execution time of an ordinary scalar product. Thus it would be advantageous if we could easily switch between the slower but exact computation and the fast approximative one (and maybe some precisions in between).

Usually in C++ libraries that provide operator overloading for matrix/vector expressions, we either have all scalar products evaluated exactly or none. Using expression templates enables us to introduce a Pascal-XSC [68] like notation for switching between exact and naive scalar products.

Pascal-XSC version:

r #*(b - A*x);

C++ version (compare [78]):

```
ExprMode::beginAccurate( RoundToNearest );
r = b - A*x;
ExprMode::endAccurate( RoundToNearest );
```

CHAPTER

6

vk — A Variable Precision Krylov Solver

⁶⁶ The first American Venus probe was lost due to a program fault caused by the inadvertent substitution of a statement of the form
 DD 3 I = 1.3 for one of the form DD 3 I = 1,3⁹⁹

Jim Horning, 1979

Almost all ideas and concepts presented in this thesis are implemented in the **v**ariable precision **k**rylov solver **vk**. The program is written in the C++ programming language and makes extensive use of generic programming paradigms and compile-time programming. The main emphasis on writing **vk** was to produce an easily maintainable and extendable code (at least outside the kernel) while simultaneously providing an acceptable performance.

Several ideas on algorithms and data structures are taken from [28, 36, 39, 84]

6.1 Functional Description of vk

Figure 6.1 gives a quick overview over the main units of \mathbf{vk} . The central object of \mathbf{vk} is a project. A project stores all information about the problem to solve

and manages all necessary steps from reading the data, creating the preconditioner, solving the linear system, logging some interesting information, verifying the iterated solution, up to finally writing the result. Additionally it provides the data types for each arithmetic class (see Section 6.1.1) and hosts the multiplicators and accumulators for computing various scalar products in.

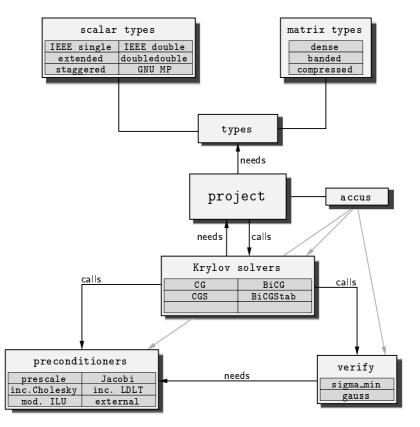


Figure 6.1: Functional structure of vk.

6.1.1 Variable Precision

In \mathbf{vk} , nearly all computations can be performed in almost arbitrary precision ranging from IEEE single up to thousands of mantissa digits. For this purpose all computations were subdivided into several *arithmetic type-classes*. All variables of such classes are of equal data type and all scalar products (or matrix-vector products) inside one class are performed in the same precision (component products and accumulation). Table 6.1 lists the possible data types (compare Section 3.2), while Table 6.2 shows the arithmetic type-classes. To realize, e.g., an exact scalar product for IEEE double vectors in *internal* computations, one would have to set

```
INTERNAL_PREC = -1, INTERNAL_PROD = 2, INTERNAL_ACCU = 67.
```

In fact, INTERNAL_PROD = -3 would suffice, but is significantly slower due to additional data conversions.

Туре	Name	type_id	Mantissa	Exp.
IEEE single precision	single	0	23 bit	8 bit
IEEE double precision	double	-1	$53 \mathrm{bit}$	$11 \mathrm{\ bit}$
Intel extended precision ^{a}	extended	-2	$64 \mathrm{bit}$	$15 \mathrm{bit}$
staggered with length 2	doubledouble	-3	106 bit	$11 \mathrm{\ bit}$
staggered ^b with length $l = -id - id$	staggered < l >	$-4, -5, \ldots$	$l \times 53$ bit	11 bit
3				
GNU MP floating-point number	multiple <l></l>	$1, 2, \ldots$	$l \times 64$ bit	31 bit
with length $l = id$				

Table 6.1:Available data types.

 a if supported by the used platform

 $^{b}\mathrm{not}$ supported by the standard version of \mathbf{vk}

Symbol	Description		
PRECOND_FACT_PREC	Datatype to store the preconditioner in (e.g. Cholesky factors).		
PRECOND_FACT_PROD	Precision to compute products of type		
	PRECOND_FACT_PREC × PRECOND_FACT_PREC with.		
PRECOND_FACT_ACCU	Datatype to accumulate numbers of type PRECOND_FACT_PREC		
	or PRECOND_FACT_PROD in.		
PRECOND_APPL_PREC	Datatype to store the preconditioned search directions in.		
PRECOND_APPL_PROD	Precision to compute products of type		
	PRECOND_APPL_PREC × PRECOND_APPL_PREC or		
	PRECOND_APPL_PREC×INTERNAL_PREC with.		
PRECOND_APPL_ACCU	Datatype to accumulate numbers of type $\tt PRECOND_APPL_PREC$		
	or PRECOND_APPL_PROD in.		
INTERNAL_PREC	Datatype to store internal used quantities in (α 's, β 's,		
	residuals, auxiliary vectors,).		
INTERNAL_PROD	Precision to compute products of type		
	INTERNAL_PREC×INTERNAL_PREC with.		
INTERNAL_ACCU	Datatype to accumulate numbers of type $INTERNAL_PREC$ or		
	INTERNAL_PROD in.		
SOLUTION_PREC	Datatype to store the iterated solution in.		
SOLUTION_CALC	Precision to compute the saxpy-operations with, used to		
	update the solution.		
VERIFICATION_PREC	Precision to compute the floating-point part of the verification		
	step in.		

Table 6.2: Adjustable arithmetic type-classes. The _PROD and _ACCU types can
be used to realize the exact scalar product (see Section 3.1).

6.1.2 Matrix Types

Currently, **vk** supports three matrix types: dense, banded, and compressed.

- **dense** A dense matrix is stored in a dim by dim memory block. This matrix type is mainly implemented for testing purposes since **vk** aims to solve sparse systems of equations.
- **banded** A banded matrix actually is a mixture between dense and sparse matrices. The elements are stored in a dense dim by bandwidth memory block data and there is a simple mapping from matrix coordinates to memory coordinates and vice versa (inside the band):

$$oldsymbol{A}(i,j) \mapsto ext{data}[i, lower_bw + j - i]$$

 $oldsymbol{data}[i,j] \mapsto oldsymbol{A}(i, i + j - lower_bw)$

compressed In vk, compressed matrices are stored in a data structure which mainly consists of a doubly linked list and two vectors to find the first element in each row, respectively column. The elements of the list are of type compressed_element (compare Figure 6.2). With this matrix type, we can also perform transpose matrix-vector products. Without this feature we could save up to fifty percent of storage and the internal routines of this matrix container would be much simpler.

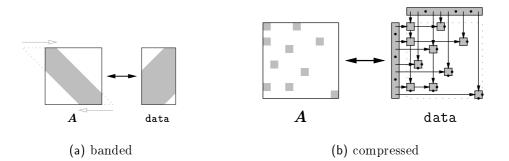


Figure 6.2: Storage schemes for banded and compressed matrices.

```
template<class T>
class compressed_element {
    int col;
    int row;
    T value;
    T* next_in_row;
    T* next_in_col;
};
```

The actually used matrix type is automatically selected by construction to get maximum performance and minimum storage overhead.

There are 3 built-in linear systems for testing purposes and the possibility to read matrices stored in the MatrixMarket [85] format. The build in types are

Gregory/Karney 4.16 This matrix is out of a collection of matrices, edited by Gregory and Karney [52]. It is defined¹ via

$$\mathsf{GK4.16} = \begin{pmatrix} 5 & -4 & 1 & & & \\ -4 & 6 & -4 & 1 & & \\ 1 & -4 & 6 & -4 & 1 & & \\ & \ddots & \ddots & \ddots & \ddots & \ddots & \\ & & 1 & -4 & 6 & -4 & 1 \\ & & & 1 & -4 & 6 & -4 \\ & & & & 1 & -4 & 5 \end{pmatrix}$$
(6.1)

This matrix is s.p.d. and its condition number is approximately n^4 .

- **Hilbert** Hilbert matrices are good examples for extremely high condition numbers at small dimensions. Usually these matrices are defined as $A_{i,j} := 1/(i+j-1)$, with i, j = 1, ..., n. Scaling with the least common multiple of the numbers 1, ..., 2n - 1 yields an integer matrix which can be stored exactly in IEEE double precision up to n = 21. Hilbert matrices are also s.p.d.
- Gregory/Karney 4.20 This matrix again is out of Gregory/Karney's matrix collection. There, we have a parameter a which we set to 1. The matrix is symmetric but indefinite and is defined as

$$\mathsf{GK4.20} = \begin{pmatrix} -1 & 2 & 1 & & & \\ 2 & 0 & 2 & 1 & & & \\ 1 & 2 & 0 & 2 & 1 & & \\ & \ddots & \ddots & \ddots & \ddots & \ddots & \\ & & 1 & 2 & 0 & 2 & 1 \\ & & & 1 & 2 & 0 & 2 \\ & & & & 1 & 2 & -1 \end{pmatrix}.$$
(6.2)

6.1.3 Preconditioners

In \mathbf{vk} we have several built-in preconditioners. Additionally, there is the possibility to import externally computed preconditioners from a file. To gain maximal performance, the preconditioner type must be selected at compile-time in order to enable the compiler to produce highly specialized code for preconditioned Krylov solvers. In particular we have the following types:

- **prescale** is a Jacobi (see below) preconditioner but applied in advance to the linear system instead of applying it in each iteration.
- **Jacobi** This preconditioner is realized by doing one step of the Jacobi iteration (compare Section 1.4.1).
- **Cholesky** decomposition. This preconditioner is only applicable to s.p.d. systems (for this and the next two compare Section 1.4.2).

¹The matrix results from a centered difference discretization of a fourth order differential equation, describing the bending line of weighted beam fixed at both ends [19].

- $\boldsymbol{L}\boldsymbol{D}\boldsymbol{L}^{T}$ A generalized Cholesky preconditioner that works for arbitrary symmetric matrices.
- LDM^{T} A modified (in the sense of Section 1.4.2) LU preconditioner.
- **external** This kind of 'preconditioner' allows us to read a Cholesky factor (L) or lower and upper triangular matrices (L and U) and a permutation matrix (P). These matrices are supposed to fulfill $LL^T \approx A$ or $LU^T \approx PA$, respectively. That is any kind of approximate triangular factorization (with pivoting) can be used.

Actually, only the prescale, Jacobi and Cholesky preconditioners can be computed in variable precision, i.e., with PRECOND_CALC_* \neq double.

6.1.4 Krylov Solvers

The implemented Krylov solvers are CG, BiCG, CGS, and BiCGStab. All solvers support preconditioning, variable precision, and verification of their iterated solution. Due to generic programming, the latter two features are completely separated from the solver. Variable precision is a feature of the underlying matrix-vector arithmetic and the verification step can be considered as a separate post processing of the solution. Thus it is easy to transform any given Krylov solver to a verified Krylov solver with variable precision.

6.1.5 Verification

As mentioned in the section above, verification of iterated solutions with \mathbf{vk} is completely separated from the solver itself. Both implemented verification algorithms only need the system matrix \mathbf{A} , the right hand side vector \mathbf{b} , an approximate triangular decomposition and obviously the approximate solution to be verified.

The supported verification algorithms are

sigma_min This is a verification method via *basic error bounds*, see Section 4.3.1.

gauss This verification method uses *improved error bounds* described in Section 4.3.2.

6.1.6 Output

Usually, there are two ASCII output files written by \mathbf{vk} . One contains the iterated solution vector and the other is used for logging all important information about the system itself and the solving and verification process. The log-files are designed to be directly usable as gnuplot [40] input data files.

```
# vk-log-file
                            : gk416_0000128_cg_Cholesky-none_[...].vk
# filename
# date
                            : Wed May 17 10:55:15 2000
# precond calc precision
                            =
                                 53 Bit (double)
# precond calc accu length =
                                 53 Bit (double)
# precond calc prod length =
                                 53 Bit (double)
                                 53 Bit (double)
# precond apply precision
                           =
# precond apply accu length = 4288 Bit (multiple<67>)
# precond apply prod length = 128 Bit (multiple< 2>)
                                53 Bit (double)
# internal precision
                           =
```

```
= 4288 Bit (multiple<67>)
# internal accu length
# internal prod length
                      = 128 Bit (multiple< 2>)
# solution precision
                      =
                         53 Bit (double)
                      = 128 Bit (multiple< 2>)
# solution calc prec
# maxCount
                      = 128
# max. residual norm
                      = 2.46204e-12
# max. error norm
                      = 1e-05
# algorithm
                      : cg
# preconditioning
                      : Cholesky (none) (0 0:00:00.00 sec)
# matrix
                      : gk416_0000128
# exact solution
                      : none
                      = 128
# dimension
                      = 634 (3.87%)
# nnz
# lower bandwidth
                      = 2
# upper bandwidth
                      = 2
# min. singular value
                      = 2.46204e-07 (0 0:00:00.00 sec)
                      = ----
# decomposition error
# ||A*x-b|| (x=exact)
                      = ---
# iter
          time res.norm
                                 abs.error ub of res.nrm.
0 1.1456e-08 ---
  1
                                             1.5628e-08
           0.01
                   2.2175e-19
                                   ____
  2
                                              1 6908e-09
# verification failed ( upper error bound \in [0.0068674,0.0068674])
 3 0.02 5.6131e-31 --- 3.3959e-09
\mbox{\tt \#} verification failed ( upper error bound \in [0.013793,0.013793])
# STAGNATION after 3 steps.
# needed 3 iterations to reach ||r||: 5.61e-31 (updated)
                              : 0.0069
# verified upper error bound
# EOF (0 0:00:00.02 sec)
```

6.2 Using vk

Since **vk** makes extensive use of the two stage programming paradigm (see Section 5.2), several quantities have already to be known at compile-time. This enables the compiler to produce highly specialized code, e.g., for the particular data types we want to use. Therefore, using **vk** consists of two steps. First we have to compile an appropriate executable (see Section 6.2.1) and secondly we need to run the executable with proper command line options (Section 6.2.2).

6.2.1 Compiling vk

Before compiling \mathbf{vk} , one have to choose the precision for each arithmetic class (compare Section 6.1.1) as well as the preconditioner. A preconditioner is selected by setting PRECOND_TYPE to an appropriate *prec_id*, i.e., by adding the definition -D PRECOND_TYPE=*prec_id* to the compile command line. Table 6.3 shows all supported preconditioners with according *prec_ids*. Similarly, the data type (Table 6.1) has to

Name	$prec_id$	Name	$prec_id$
none	0	LDL^{T}	4
prescale	1	LDM^{T}	5
Jacobi	2	external	6
Cholesky	3		

Table 6.3: All preconditioners supported by vk.

be selected for each arithmetic type-class, by adding -D class_name = type_id to the compiler options. The default precision for any class is IEEE double (type_id=-1).

If one arithmetic class is adjusted to have the data type doubledouble you need to link the doubledouble library [16] and add the options:

-L path/to/doubledouble/library/ -l doubledouble -I path/to/doubledouble/includes/ -D DD_INLINE -D x86 -m 486 -D VK_USE_DOUBLEDOUBLE

You always need to link the Profil and Bias libraries² [70]:

-L path/to/profil_bias/library/ -l Profil -l Bias -I path/to/profil_bias/includes/

Finally you need

-L path/to/gmp/library/ -l gmp -I path/to/gmp/includes/ -I path/to/gmp++/source/

to tell the compiler where to find the gmp++ library³ which provides the multiple<N> data type.

All together the command line for compiling \mathbf{vk} should look as follows:

g++ -o my_vk vk.cc -I ... options for doubledouble options for profil/bias options for gmp/gmp++ define for the preconditioner defines for the data types

Assume we need a Krylov solver with Cholesky preconditioner. The Cholesky factorization shall be computed in a higher precision, say in doubledouble, but stored in the IEEE double format. Additionally we need all internal scalar products to be exact ones. Then we may compile \mathbf{vk} via

```
g++ -o my_vk vk.cc -I ..
-L ~/lib/doubledouble -l doubledouble
-I ~/include/doubledouble
-D DD_INLINE -D x86 -m 486 -D VK_USE_DOUBLEDOUBLE
-L ~/lib/profil_bias -l Profil -l Bias
-I ~/include/profil_bias
-L /usr/lib -l gmp
-I /usr/include/gmp
-I ~/source/gmp++
```

 $^{^2\}mathrm{Note}$ that the original versions produces oodles of warnings, but should compile anyhow.

³This library is obtainable from the author of this thesis.

-D	PRECOND_TYPE	=	3
-D	PRECOND_FACT_PROD	=	-3
-D	PRECOND_FACT_ACCU	=	-3
-D	INTERNAL_PROD	=	2
-D	INTERNAL_ACCU	=	67

6.2.2 Command Line Options

Table 6.4 shows all currently supported command line options $% \left(\frac{1}{2} \right) = 0$

Switch	Arg. type	Description
-h,help	none	prints a short help message (similar to
		this table).
-V,version	none	prints a description of the executable.
-m,matrix	string	filename of the matrix or $gk416_n$,
		$gk420_n$, or $hilbert_n$.
-r,rhs	string	filename of the right hand side vector
		or 'set' to set $\boldsymbol{b} = \boldsymbol{A}\boldsymbol{x}$, where \boldsymbol{x} is the
		comparative solution.
-c,compsol	string	file-name of the comparative solution.
-a,algorithm	string	'CG', 'BiCG', 'CGS', or 'BiCGStab'.
-p,preconditioner	string	option for the preconditioner (e.g., the
		file-name for 'external').
-n,maxcount	int	maximum number of iterations.
-e,eps	float	maximum residual norm (or maximum
		error norm, if verification is enabled).
-v,verify	none	enable verification
-w,write	string	name of the file to store the iterated
		solution in.
-l,logfile	string	name of the file to store the logging
		messages in (may be 'auto' for auto-
		matically creating a file name).
-q,query	string	Used for communication between ${\bf vk}$
		and xvk

Table 6.4:	Command	line	options	for	vk
------------	---------	------	---------	-----	----

6.3 xvk — A Graphical User Interface

Because there are a lot of compile-time and run-time options, include paths, libraries, and so on, it might be a little difficult to get \mathbf{vk} running. To assist you with that, there is a graphical user interface (GUI) for \mathbf{vk} called \mathbf{xvk} . It is written in C and uses the X-toolkit GTK [53].

The GUI of **xvk** is organized as a notebook with 4 pages: 'Compile-Time Constants', 'Compiler Options', 'Run-Time Parameters', and 'Browse Logfiles' (the latter actually is in progress).

The page 'Compile-Time Constants' (see Figure 6.3) allows you to set all compiletime constants to appropriate values. The arithmetic type-classes (compare Table 6.2) are subdivided into 5 categories. For each type-class, there is a little menu that lets you choose a numerical data type (compare Table 6.1). If you select multiple<N>, the 'Precision' entry becomes sensitive. You may adjust N here.

Additionally, you can select the preconditioner type, by simply clicking on its radio button.

statiyp es:			Preconditionens
Preconditioning			⇔ none
Factorization			⇔ prescale
Datatype:	double	Precision: 1	🗢 Jacobi
Products:	double	Precision: 2 >	🗢 inc. Cholesky
Accumulation:	double	Precision: 87 3	↓ mod. inc. LDL^T
Application			⇒ erdernal
Datatype:	double	교 Precision: 1 순	
Products:	double	Precision: 2	
Accumulation:	double	Precision: 67	
Internal Computation			
Datatype:	double	Precision: 1	
Products:	multiple <n></n>	Precision: 2	
Accumulation:	multiple <n></n>	→ Precision: 67 →	
Solution			
SAXPY	extended	교 Precision: 1 순	
Verification			
Datatype:	double	Precision: 1	Load Parameters from

Figure 6.3: xvk: Compile-time constants.

On the page 'Compiler Options' (see Figure 6.4) all remaining options of the compiler command line can be adjusted. Since it shouldn't be necessary to change

these settings after installation, it is possible to save them on disk in order to get them preloaded every time you start xvk.

Additionally, at the bottom of this page, there is a window displaying the output of all compiler calls initiated by xvk. This output is also written into a logfile.

🛚 xvk - a variab	e precision krylov solver Axel Facius 2000	CAVN, University of Ka	rtsrutie 💶 🗆 🗵
Compile-Time Co	nstants Compiler Options Run-Time Parameters Browse	Logfiles	
Compilers			
⇒ KCC ∧ e	jcswith debug		
C++ optimizatio	an: 🔽 🕀		
C optimization:			
Compiler option	ns: -====================================		_
bin directory:	/home/axel/bin/vk		Browse
Includes			
vit:	/home/axel/source/vk	Browse	oad Settings
gmp++:	/home/axel/source/gmp++	Browse	Save Settings
doubledouble:	/home/avai/include/doubledouble	Browse	
profil/bias:	/homa/axel/include/profi_bias	Browse	
Libraries			
libgrap.a:	/usn/lb/lbgmp.a	Browse	
lib double doubl	e.a /home/avai/lb/lb/bubledouble.a	Browse	
libprofil.a:	/home/avai/lib/libprofil.a	Browse	
libbias.a:	/home/avail/b/libbias.a	Browse	Out
Logfie			
Filename: vk.	og Browse	Res	can Ciear
			<u> </u>
			R I

Figure 6.4: xvk: Compiler options.

The notebook page 'Run-Time Parameters' (see Figure 6.5) allows you to adjust all options understood by \mathbf{vk} and beyond it you can choose some additional options only provided by \mathbf{xvk} . The latter are computed by calling MATLAB preliminary to \mathbf{vk} .

In particular you may

- select the system matrix either as one of the built-in types or by specifying a file in the MatrixMarket format (see [85]) or
- select the right-hand-side vector either by setting all its components to one, by computing it according to the comparative solution (must be given in this case), or by specifying a file in the MatrixMarket format and
- select a comparative solution,

- pass some options to the compiled-in preconditioner,
- select a reordering algorithm,
- choose a Krylov algorithm,
- specify the stopping criteria(s),
- set the verification mode, and finally
- adjust some project properties

xvk - a variable precis	ian krylov solver	Axel Factus 2000	CAVN, University of	if Karlsruhe 🔲 🗖 🖻
Compile-Time Constants	Compiler Options	Run-Time Parameters Brow	vse Logfiles	
System				
System Matrix				
© Gregory/Kamey 4.18	8 512 ÷	Neumaier:	512 >	
© Gregory/Kamey 4.21	512 ->	 Hilbert 	12 >	
 MatrixMarket from fil 	e: /home/avai/mat	ionarket/astroph/mcca.mtx		Browse
Right-hand-side vecto	e			
→ set to (1,1)	compute as A	w (x is the comparative solu	(inoit)	
Read from file:				Browse
Comparative solution:				Browse
External Preconditioner Incomplete Factorizatio exploit symmetry Algorithms			fil-in 	Browse
÷06 ÷9	RES	log(epsilor):	-5 👌 👳 ni	109
	(8	Iteration:	20 2 45	gma_min
⇔ BIOG → TF	QMR.	count single iterations	↓ G	auss
- BICGStab				
Logging				
Project dir:	log/astroph/ncc	(a)		Browse
Log-File:	auto			Browse
Iterated solution:	/dev/null			Browse
Start Computation Lo	ad Parameters from	1		Gut

Figure 6.5: xvk: Run-time parameters.

CHAPTER

Computational Results

⁶⁶ Numerical subroutines should deliver results that satisfy simple, useful mathematical laws whenever possible. **??**

Donald E. Knuth, 1981

7.1 Level of Orthogonality

As a first example of the effectiveness of higher precision arithmetic, we solved again the system GK4.16(1023) (compare Section 2.3.2, Figure 2.4). The residual norms and error norms, achieved by using a staggered precision arithmetic, are plotted in Figure 7.1. The letter *l* denotes the *staggered length*, i.e., the number of floating-point numbers defining a staggered number (see Section 3.2.1). The case l = 1 corresponds with Figure 2.4. Since the staggered arithmetic is simulated in software, it cannot compete with the built-in double arithmetic (l = 1) in computing time. However, despite getting more accurate solutions, (which might be unnecessary for practical problems) we observe a significant saving in the number of iterations.

In Figure 7.2 we show the level of orthogonality of the new residual-vector \mathbf{r}_{m+1} to the previous ones: $\max_{k=1}^{m} \{ \langle \mathbf{r}_k \mid \mathbf{r}_{m+1} \rangle / (||\mathbf{r}_k||_2 ||\mathbf{r}_{m+1}||_2) \}$. Beside the expectedly better orthogonality at the beginning, the loss of orthogonality is not delayed very much. However, this little improvement is sufficient to give a significantly better convergence.

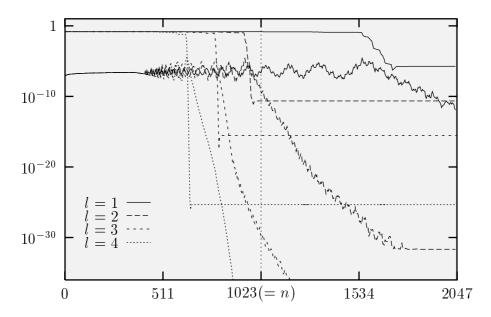


Figure 7.1: The Euclidean norms of the residuals (oscillating) and errors (more or less piecewise constant) during solving the GK(1023) system with staggered length l from 1 to 4.

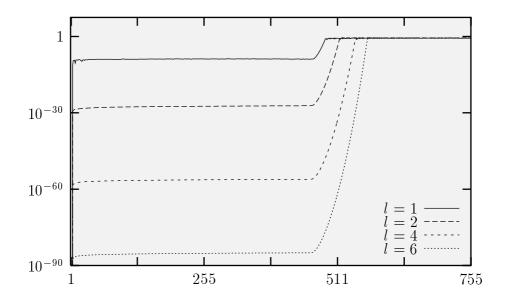


Figure 7.2: The level of orthogonality during solving the GK(1023) system with staggered length $l \in \{1, 2, 4, 6\}$.

7.2 High Precision and Exact Scalar Products

In this example, we demonstrate the effect of high precision and exact scalar products. The used linear system is called fidap009 and is described in the appendix as Matrix A.5. We solved this linear system with a preconditioned Conjugate Gradient solver. For the preconditioner we used an incomplete Cholesky factorization with drop-tolerance 10^{-10} . For several larger droptolerances, i.e. more sparse preconditioners, we got no convergence, neither with double, nor with extended arithmetic. The used arithmetics were:

double	IEEE double precision,
doubleX	IEEE double precision with exact scalar products,
extended	Intel's extended precision format, and
extendedX	Intel's extended precision format with exact scalar products ^{a}

^{*a*}This accumulator needs approximately 32 kbyte. Since \mathbf{vk} provides a central accumanagement, only one accumulator of this size is allocated.

Figure 7.3 shows the relative error norms vs. number of iterations. As we can see, the higher precision used for accumulation, results in faster convergence and increased accuracy.

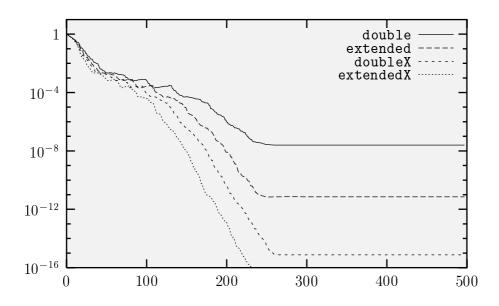


Figure 7.3: Solving a fidap009 system (Matrix A.5) with an incomplete Cholesky preconditioned CG solver. The curves represent the relative error norms vs. number of iterations achieved by computing with differently precise arithmetics (IEEE double resp. Intel's extended with standard scalar products (double/extended) and with exact scalar products (doubleX/extendedX.)

Taking into consideration, that exact scalar products, sufficiently supported in hardware, need not to be slower than ordinary scalar products, there it is simply no reason not to utilize this technique. However, presently, this operation is simulated in software only and therefore is relatively slow compared to the built-in arithmetic. Figure 7.4 shows the same experiments as Figure 7.3 but now plotted against computing time (on a Pentium II, 400 Mhz).

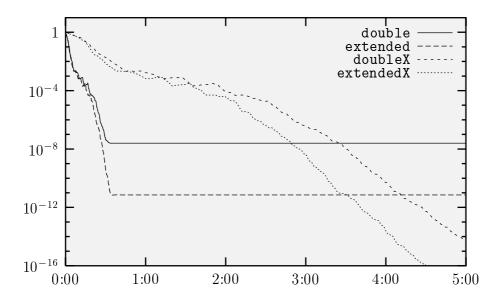


Figure 7.4: This figure shows the same experiments as Figure 7.3 but now plotted against computing time (in sec) instead of iteration counts.

Of course, for practical problems, the exact scalar product often provides much more precision than is actually needed. To get an idea on how many precision would suffice, we solved the fidap009 system with basic data type double and different accumulator precisions.

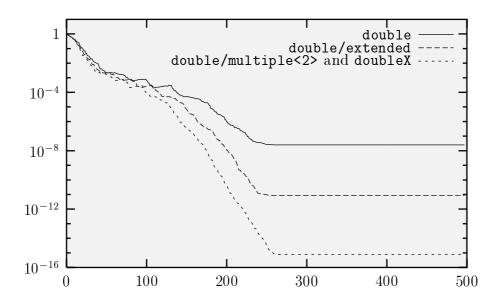


Figure 7.5: This figure shows the same experiments as Figure 7.3 but now we used different scalar products while leaving the basic data type fixed at double.

Namely, we used accumulation in double, extended, multiple<2> (128 bit mantissa length, see Section 3.2.2), and exact accumulation. The results are shown in Figure 7.5. The curves that corresponds with exact accumulation and multiple<2>accumulation differ so little that they appear as a single line. That is, in this particular case, 128 bit mantissa length was sufficient for almost error free accumulation.

7.3 Beyond Ordinary Floating-Point Arithmetic

While in the latter section we only saved iterations by using a more precise arithmetic, we now show that there are examples that are not solvable in standard floating-point arithmetics at all or at least speed up by using higher precision.

For this purpose, we solved a linear system with the Hilbert matrix of dimension 13. To get a simple stopping criterion, we first computed a verified solution that is guaranteed to have at least 16 correct decimal digits and then stopped each of the following iterations, when the approximated solutions coincide with the verified solution within the first five digits.

Each experiment was carried out twice, once with a (complete) Cholesky preconditioner and once without preconditioning. The results for various arithmetics are displayed in Table 7.1. If there are two entries in the 'Arithmetic' column (separated by a slash), then the first denotes the data-type and the second is the accumulation precision. If there is only one arithmetic data type given, then all computations are performed with this type.

Arithmetic	No Precond.		Cho	olesky
Arithmetic	iter	time	iter	time
double	>130	()	>130	()
double/extended	>130	()	>130	()
extended	>130	()	>130	()
double/multiple<2>	89	(0.13)	3	(<0.01)
double/exact	89	(0.13)	3	(<0.01)
extended/exact	37	(0.04)	3	(<0.01)
multiple<2>	23	(0.04)	3	(0.01)
multiple<4>	16	(0.02)	3	(0.01)
multiple<5>	13	(0.02)	3	(0.01)

Table 7.1: This table shows the number of iterations and the computing time, needed to obtain at least 5 correct decimal digits in the iterated solution. We stopped the process at a maximum of 130 steps (displayed in gray letters).

As we can see, the *smallest* arithmetic enabling convergence, is IEEE double with 128 bit accumulation (double/mutiple<2>). Using extended precision (with mutiple<2> accumulation), significantly speeds up the computation in the non-preconditioned case and further increasing the precision saves up to 85% computing time. With 320 bit mantissa length mutiple<5>, we match the *exact precision* property of convergence after at most n steps.

Though we used a (complete) Cholesky preconditioner in the right column, i.e. a direct solver in each iteration, we also got convergence only with at least 128 bit accumulation. Further increasing the precision does not save more iterations and consequently does not save computing time. However, even with this direct solver in each step, we need 3 iterations to get 5 correct digits.

7.4 Does Higher Precision Increase the Computational Effort?

Inspecting, for example, the MATLAB implementations of Krylov subspace solvers, we can see that the residual norm which is used for evaluating the stopping criterion, is always computed as norm r = norm(b - A * x); instead of using the norm of the updated residual. This effort is often necessary because in finite precision these two theoretically equal values tend to differ significantly after sometimes only a few iterations (see Figure 7.6).

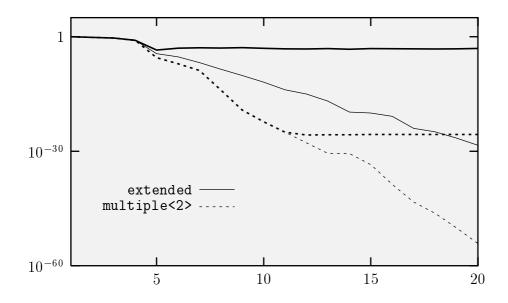


Figure 7.6: This figure compares the norm of the iteratively updated residual (thin lines) and the exact residual $\|\boldsymbol{b} - \boldsymbol{A}\boldsymbol{x}\|_2$ (thick lines) computed with differently precise arithmetics (extended: solid, multiple<2>: dashed).

However, there is one extra matrix-vector multiplication at each step and this information is solely used to decide whether the iteration should be stopped or not. Fortunately, we can do much better. Assume we have the internal precision adjusted to doubledouble (staggered with length 2). For generating the Krylov spaces, we need

$$Ar = A(r^{(1)} + r^{(2)}) = Ar^{(1)} + Ar^{(2)}$$

That means, *increasing the precision by one* is comparable with doing one extra matrix-vector product. However, with this approach, we do not only improve the stopping criterion, but also significantly improve the accuracy of the iterated so-

lution, see Figure 7.6. This effect is demonstrated with the mcca matrix which is described in Matrix A.7.

7.5 Solving III-Conditioned Test-Matrices

Here we solved some GK4.16(n) (see (6.1) on page 89) systems with a preconditioned CG algorithm with Cholesky preconditioner (see Table 7.2).

Matrix		double	doubleX	extended	multiple<2>
n = 100	error	$1.3 \cdot 10^{-9}$	$1.6 \cdot 10^{-15}$	$2.0 \cdot 10^{-12}$	$1.4 \cdot 10^{-23}$
$\sigma_{\rm min} = 8.4 \cdot 10^{-7}$	bound	$7.5 \cdot 10^{-3}$	$2.0 \cdot 10^{-3}$	$4.2 \cdot 10^{-6}$	$4.9 \cdot 10^{-15}$
vtime < 0.01 sec	iter	3*	3*	2	2
	time	< 0.01	< 0.01	< 0.01	0.01
n = 1000	error	$9.6 \cdot 10^{-5}$	$5.3 \cdot 10^{-15}$	$9.0 \cdot 10^{-8}$	$6.9 \cdot 10^{-25}$
$\sigma_{\rm min} = 9.7 \cdot 10^{-11}$	bound	> 1	> 1	> 1	$1.0 \cdot 10^{-11}$
vtime = 0.04sec	iter	3*	3*	3*	3
	time	< 0.01	0.13^{a}	0.02	0.15
n = 10000	error	> 1	$3.2 \cdot 10^{-14}$	$2.7 \cdot 10^{-3}$	$4.8 \cdot 10^{-25}$
$\sigma_{\rm min} = 9.7 \cdot 10^{-15}$	bound	> 1	> 1	> 1	$2.5 \cdot 10^{-6}$
vtime = 0.47 sec	iter	3*	5^{*}	3*	5
	time	0.08	2.33^{a}	0.10	3.06
n = 50000	error	> 1	$2.5 \cdot 10^{-13}$	> 1	$1.9 \cdot 10^{-14}$
$\sigma_{\rm min} = 2.2 \cdot 10^{-17}$	bound	> 1	> 1	> 1	$3.3 \cdot 10^{-7}$
vtime = 3.01sec	iter	3*	12^{*}	13^{*}	9
	time	0.45	28.66^{a}	2.83	28.73
n = 100000	error	> 1	$2.8 \cdot 10^{-13}$	> 1	$4.7 \cdot 10^{-14}$
$\sigma_{\rm min} = 4.1 \cdot 10^{-18}$	bound	> 1	> 1	> 1	$7.0 \cdot 10^{-7}$
vtime = 4.83 sec	iter	3*	11*	7^{*}	12
	time	0.93	52.55 ^a	2.96	77.31

Table 7.2: Solving some $\mathsf{GK4.16}(n)$ systems with an incomplete Cholesky preconditioned CG solver. The iteration was stopped after 5 correct digits were guaranteed (by the verification procedure) or after stagnation (gray). The cases where we have 5 digits accuracy, compared to the previously computed highly precise verified solution (but not verified), are displayed in dark gray.

^aNote that this loss of convergence speed is caused by the slow software simulation of the exact scalar product. Sufficiently supported in hardware, doubleX should need the same time as double.

In this and all following examples, we stopped the iteration as soon as five correct digits of the solution could be guaranteed or after stagnation of the residual norm (marked by an *). With *error* we denote the actual relative error of the approximate solution. Usually, we have no exact solution available and therefore we cannot compute the *error*. However, for these examples we use a very tight enclosure of the exact solution, computed with a high precision arithmetic. With *bound* we denote the computed upper bound of the error and *iter* and *time* are the

number of iterations and the time needed for these iterations (in sec, measured on a PentiumII/400). The quantity *vtime* denotes the time needed to compute a rigorous bound for the smallest singular value of A.

As we can see, verification with the techniques described in Section 4, is only possible in conjunction with higher precision arithmetic. Even if we are only interested in a non-verified solution, we cannot trust in standard floating-point (double) arithmetic. However, simply replacing all floating-point scalar products by exact scalar products suffices to deliver always enough correct digits in the approximate solution, although not verified. Using a 128 bit arithmetic we always achieved fast convergence and highly accurate verified solutions.

In the next example we solve some Hilbert systems, again with a Cholesky preconditioned CG solver, see Table 7.3. With a standard double or extended arithmetic for the solver, we can only handle very small dimensions, while a mantissa length of 128 bit always is sufficient to get fast convergence and good approximations to the solution. At dimension 14, the Cholesky decomposition (computed in double) fails.

		double		extended		multiple<2>		Chol.
\dim	$\sigma_{ m min}$	iter	time	iter	time	iter	time	prec.
8	$3.60 \cdot 10^{-05}$	3	< 0.01	1	< 0.01	1	< 0.01	t
10	$2.29 \cdot 10^{-05}$	> 10		2	< 0.01	2	< 0.01	doudle
12	$5.11 \cdot 10^{-07}$	> 12		> 12	_	3	< 0.01	00
13	$3.05 \cdot 10^{-08}$	> 13		> 13	_	4	< 0.01	
15	$4.92 \cdot 10^{-09}$	> 15		> 15		1	0.01	aultiples
17	$2.83 \cdot 10^{-10}$	> 17		> 17		1	0.02	Ze,
19	$9.62 \cdot 10^{-12}$	> 19		> 19		1	0.02	lt i
21	$3.60 \cdot 10^{-13}$	> 21		> 21	_	1	0.02	TIL

Table 7.3: Here we aimed to verify 5 correct digits in the solution of various Hilbert systems. For dimensions up to 13, the Cholesky preconditioner was computed in double while the higher dimensional Hilbert matrices were factorized with a multiple<2> arithmetic.

Increasing the precision used to compute the Cholesky decomposition enables us to handle larger Hilbert matrices. In Table 7.3, rows 15-21 we used the data type multiple<2> for computing the preconditioner. Now the Cholesky decomposition is sufficient to solve the linear system in one step.

We stop at dimension 21, because it is not possible to store higher dimensional Hilbert matrices in IEEE double exactly (compare Section 6.1.2).

This high precision preconditioning also allows us to handle larger dimensions for the GK4.16 matrices. Table 7.4 shows the results of the GK4.16(2000000) system.

Just to see how far we can go with Hilbert matrices, I extended \mathbf{vk} to allow multiple precision system matrices (in the standard version, system matrices are always stored in double). With this extension we can even solve Hilbert matrices of dimension 42 [1] and higher¹.

¹In fact, **vk** solved the Hilbert 42 system in less than 8 seconds with 113 guaranteed decimals (using a data type with 2560 bits mantissa length, i.e. approximately 770 decimal digits).

Matrix		doubleX	extended	multiple<2>	multiple<3>
n = 2000000	error				
$\sigma_{\rm min} = 4.2 \cdot 10^{-24}$	bound	> 1	> 1	$1.8 \cdot 10^{-2}$	$6.7 \cdot 10^{-7}$
vtime = 123.27sec	iter	8*	2^{*}	2*	2
	time	903.34^{a}	18.15	261.42	306.54

Table 7.4: This table shows the same experiment as Table 7.2 but now with
dimension $2\,000\,000$ (this was the largest possible dimension solvable
on my PC due to storage limitations). See Table 7.2 for explanation
of footnote^a.

Our final 'test' example is matrix $\mathsf{GK4.20}$ (see (6.2) on page 89). This matrix is symmetric and indefinite but it nevertheless turns out that CG works fine. Here we used a modified \boldsymbol{LDL}^T preconditioner, Table 7.5.

Matrix		double	doubleX	extended	multiple<2>
n = 100000	error	$2.7 \cdot 10^{-7}$	$1.5 \cdot 10^{-8}$	$1.3 \cdot 10^{-10}$	$3.5 \cdot 10^{-14}$
$\sigma_{\min} = 9.8 \cdot 10^{-11}$	bound	$5.6 \cdot 10^{-4}$	$3.1 \cdot 10^{-4}$	$2.9 \cdot 10^{-7}$	$1.7 \cdot 10^{-13}$
vtime = 15.44sec	iter	3*	3*	2	2
	time	2.43	13.87^{a}	0.74	10.61
n = 500000	error	$1.5 \cdot 10^{-6}$	$1.4 \cdot 10^{-7}$	$7.3 \cdot 10^{-10}$	$8.6 \cdot 10^{-14}$
$\sigma_{\min} = 2.9 \cdot 10^{-12}$	bound	$4.6 \cdot 10^{-2}$	$2.2 \cdot 10^{-2}$	$2.9 \cdot 10^{-5}$	$2.2 \cdot 10^{-9}$
vtime = 78.11sec	iter	3*	3*	3*	2
	time	12.08	68.7^a	16.78	53.67
n = 1000000	error	$7.3 \cdot 10^{-6}$	$1.6 \cdot 10^{-6}$	$3.6 \cdot 10^{-9}$	$1.1 \cdot 10^{-13}$
$\sigma_{\rm min} = 8.1 \cdot 10^{-13}$	bound	0.39	> 1	$2.4 \cdot 10^{-4}$	$8.2 \cdot 10^{-9}$
$vtime = 157.03 \operatorname{sec}$	iter	3*	3*	3*	2
	time	24.58	141.4^{a}	33.55	1:48.3

Table 7.5: This table shows the results of our experiments with the GK4.20 matrices. See Table 7.2 for explanation of the used notations. See Table 7.2 for explanation of footnote^a.

7.6 Verified Solutions for 'Real-Life' Problems

In this section we investigate some example systems taken from various application areas such as fluid dynamics, structural engineering, computer component design, and chemical engineering (see Appendix A).

Symmetric positive definite systems we always solved with a Cholesky preconditioned Conjugate Gradient solver. In the nonsymmetric case we show only the results of the fastest ILU preconditioned Krylov solver.

Particularly we utilized various solvers (BiCG, CGS, and BiCGStab) and incomplete preconditioners. In this context of high precision arithmetics, the GMRES algorithm couldn't compete with the *short recurrence solvers*. Since GMRES needs most arithmetic operations and storage anyway, this lack is even reinforced by the increased requirements in memory and computing time for the high precision arithmetic operations.

In Table 7.6 we compare the results achieved by using several arithmetics (see Table 7.2 for explanation). Table 7.7 gives a quick overview over some systems we solved with \mathbf{vk} .

Matrix		double	doubleX	extended	multiple<2>
fidap009	error	$1.9 \cdot 10^{-4}$	$1.1 \cdot 10^{-13}$	$1.6 \cdot 10^{-6}$	$6.2 \cdot 10^{-11}$
n = 4683	bound	0.33	$6.6 \cdot 10^{-2}$	$2.0 \cdot 10^{-4}$	$1.1 \cdot 10^{-8}$
$\sigma_{\rm min} = 2.9 \cdot 10^{-4}$	iter	3*	4*	3*	2
vtime = 22.55 sec	time	0.53	6.72^{a}	0.68	2.82
s3rmt3m1	error	$3.4 \cdot 10^{-5}$	$8.6 \cdot 10^{-14}$	$2.5 \cdot 10^{-8}$	$4.3 \cdot 10^{-11}$
n = 5489	bound	> 1	> 1	$9.7 \cdot 10^{-2}$	$5.2 \cdot 10^{-7}$
$\sigma_{\rm min} = 3.5 \cdot 10^{-7}$	iter	3*	3*	3*	2
$vtime = 507.6 \operatorname{sec}$	time	1.14	18.40^{a}	1.37	12.42
e30r5000	error	$1.1 \cdot 10^{-11}$	$1.5 \cdot 10^{-14}$	$5.4 \cdot 10^{-12}$	$1.2 \cdot 10^{-38}$
n = 9661	bound	> 1	> 1	0.7	$1.4 \cdot 10^{-32}$
$\sigma_{\min} = 3.7 \cdot 10^{-12}$	iter	3*	3*	3*	2
vtime = 2h09.47min	time	3.86	69.4^{a}	5.45	70.41
e40r5000	error	$2.8 \cdot 10^{-11}$	$1.9 \cdot 10^{-14}$	$2.3 \cdot 10^{-14}$	$1.6 \cdot 10^{-32}$
n = 17281	bound	> 1	> 1	> 1	$1.3 \cdot 10^{-29}$
$\sigma_{\rm min} = 1.5 \cdot 10^{-13}$	iter	3*	3*	3*	2
vtime = 6h48.09min	time	9.04	232.3^{a}	13.00	848.0

Table 7.6: Solving some 'real-life' problems with different arithmetics. See Ta-
ble 7.2 for explanation of footnote^a and the used notations.

Again, we often have enough correct digits in the approximate solution but usually we are not aware of this fact. Particularly, we are only able to prove this by using a higher precision arithmetic.

7.7 Verification via Normal Equations

In the nonsymmetric case, we have to meet the assumption

$$\sigma_{\min}(\boldsymbol{L}\boldsymbol{U}) > \|\boldsymbol{L}\boldsymbol{U} - \boldsymbol{A}\|_2 \tag{7.1}$$

which is sometimes a problem if either $\sigma_{\min}(\boldsymbol{L}\boldsymbol{U})$ is very small or \boldsymbol{A} is ill-conditioned and its elements are large. In such cases it is often advantageous to switch to the normal equations. We stress that we actually do not have to compute $\boldsymbol{A}^T \boldsymbol{A}$ and $\boldsymbol{A}^T \boldsymbol{b}$ [98]. Although we have squared the smallest singular value (which possibly makes it more difficult to find a verified lower bound if $\sigma_{\min}(\boldsymbol{L}\boldsymbol{U}) < 1$), we now do not have to fulfill (7.1) anymore.

Using this technique, we solved, e.g., the mcca system (see Matrix A.7). For this matrix, vk computes a lower bound for the smallest singular value as $2.4 \cdot 10^{-2}$ (this

Name	dim	nnz	cond	bound	time
fs 680 1	680	2646	$2.1 \cdot 10^4$	$3.56 \cdot 10^{-38}$	$21.75 \sec$
west2021	2021	7353	$7.5 \cdot 10^{12}$	$9.43 \cdot 10^{-25}$	$705.93 \sec$
mvmtls4000	4000	8784	$2.7 \cdot 10^{7}$	$3.62 \cdot 10^{-30}$	$1\mathrm{h}03.55\mathrm{min}$
pores2	1224	9613	$3.31 \cdot 10^{8}$	$4.87 \cdot 10^{-17}$	$292.68 \mathrm{sec}$
bcsstk08	1074	12960	$4.7 \cdot 10^{7}$	$9.49 \cdot 10^{-28}$	$82.72 \sec$
pde2961	2961	14585	$9.49 \cdot 10^2$	$1.77 \cdot 10^{-14}$	7.22 sec
add32	4960	23884	$2.14 \cdot 10^2$	$6.89 \cdot 10^{-15}$	$1\mathrm{h}08.28\mathrm{min}$
fidap009	4683	95053	$1.04 \cdot 10^{7}$	$1.1 \cdot 10^{-8}$	$25.37 \sec$
s3rmt3m1	5489	112505	$1.33 {\cdot} 10^{10}$	$5.2 \cdot 10^{-7}$	$520.02 \sec$
e30r5000	9661	306356	$1.27 \cdot 10^{11}$	$1.2 \cdot 10^{-38}$	$2\mathrm{h}10.57\mathrm{min}$
e40r5000	17281	553956	$1.4 \cdot 10^{16}$	$1.3 \cdot 10^{-29}$	$7h02.09 \min$

Table 7.7: A quick overview over some systems we solved with vk. The objective was to get 5 correct digits in the iterated solution. Since convergence sometimes was very fast, we overshot at times. We only display the results of the smalles arithmetic that delivers these 5 digits (almost always multiple<2>).Note that 'time' denotes the overall time for solving and verification.

seems to be roughly underestimated due to the very high condition number) and an upper bound for $\|\boldsymbol{L}\boldsymbol{U} - \boldsymbol{A}\|_2$ as $2.31 \cdot 10^9$. That is we cannot apply Theorem 4.4 directly. Switching to the normal equations, **vk** finds $7.67 \cdot 10^3$ as a lower bound for $\sigma_{\min}(\boldsymbol{A}^T \boldsymbol{A})$. Applying a Cholesky preconditioned CG algorithm, we are able to verify five decimal digits in less than three seconds.

7.8 Performance Tuning

The verification time depends strongly on the number of nonzero elements and the bandwidth of L and U. Therefore, it is advantageous to reduce these quantities. We discuss two possibilities to achieve this reduction: column/row reordering algorithms and incomplete factorizations (see Section 1.4.2).

Drop-Tol.	nnz	$\sigma_{\min}(oldsymbol{L}oldsymbol{U})$	$\ oldsymbol{L}oldsymbol{U}-oldsymbol{A}\ _2$	total time	bound
complete	54058~(6.67%)	$4.2284 \cdot 10^{-2}$	$4.50 \cdot 10^{-15}$	$9.95 \sec$	$1.23 \cdot 10^{-26}$
$1 \cdot 10^{-4}$	32083~(3.96%)	$4.2456 \cdot 10^{-2}$	$2.43 \cdot 10^{-3}$	$7.05 \sec$	$1.60 \cdot 10^{-8}$
$1 \cdot 10^{-3}$	26843~(3.31%)	$4.4330 \cdot 10^{-2}$	$7.68 \cdot 10^{-2}$	failed	

Table 7.8: Here we demonstrate the possible speed up by using incomplete LUfactorizations at the example of Matrix pde900 (A.8).

The problem with incomplete factorizations is that we have to bear inequality (7.1) in mind. Particularly for ill-conditioned systems $\|LU - A\|_2$ grows heavily with increasing sparsity in L and U. However, if the smallest singular value of A is not too small, we can achieve a significant improvement, as shown in Table 7.8 at

the example of Matrix pde900 (A.8). Switching from a complete LU preconditioner to incomplete LU with drop-tolerance 10^{-4} saves nearly 30% computing time

Applying reordering algorithms can sometimes dramatically speed up the verification process. We illustrate the effect of reordering with the bcsstk08 and tols4000 matrix (see Matrices A.2 and A.12). Because bcsstk08 is symmetric, we applied the symmetric minimum degree reordering algorithm. This algorithm computes a permutation matrix P to a given matrix A, such that the Cholesky factors of PAP^T have less nonzero elements, see Figure 7.7.

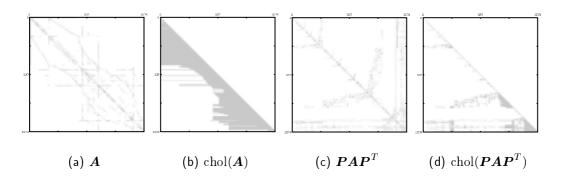


Figure 7.7: spy

Matrix tols4000 is nonsymmetric and therefore we applied a reverse Cuthill-McKee algorithm, especially designed to deliver smaller bandwidths in the LU factors.

Table 7.9 shows some experiments with and without reordering.

Reordering	Precond.	Pree	$\operatorname{conditioner}$	total	
Algorithm	(droptol)	nnz	lo/up bandw.	time	bound
none	$Chol(10^{-5})$	115645	591/1	$158.62 \sec$	$9.01 \cdot 10^{-7}$
SymmMinDeg	$Chol(10^{-5})$	23688	1054/1	$59.35 \sec$	$5.35 \cdot 10^{-8}$
none	$ILU(10^{-6})$	13584	2401/3218	$63.55\mathrm{min}$	$3.62 \cdot 10^{-30}$
Cuthill-McKee	$ILU(10^{-6})$	14820	89/90	$34.84 \sec$	$6.10 \cdot 10^{-28}$

Table 7.9: Reordering algorithms can significantly speed up convergence. Thistable shows a symmetric and a nonsymmetric example. The first aimsto reduce the number of nonzero elements while the second tries toreduce the bandwidth.

Conclusion

⁶⁶ Calvin: I think we've got enough information now, don't you? Hobbes: All we have is one "fact" you made up. Calvin: That's plenty. By the time we add an introduction, a few illustrations, and a conclusion, it will look like a graduate thesis. ??

Calvin and Hobbes (by Bill Watterson), 1991

⁶⁶ So eine Arbeit wird eigentlich nie fertig, man muß sie für fertig erklären, wenn man nach Zeit und Umständen das Möglichste getan hat.² **99**

Johann Wolfgang von Goethe, Italienische Reise II, 16.3.1787

As the main result of our investigations, we conclude that traditionally used arithmetics (mostly IEEE double precision) are often *not the best choice* for solving linear systems of equations.

Both, theoretically and by examples, we showed that iterative solvers — particularly Krylov subspace methods — heavily suffer from rounding errors. Usually, computationally expensive reorthogonalization strategies (or even full orthogonalizing methods) are utilized to work against arithmetic insufficiencies.

In this thesis, we showed that using *improved* arithmetics can lead to much better results, compared to those obtained from *ordinary* floating-point arithmetic. Particularly, exchanging the classically used floating-point scalar product by the exact scalar product often suffices to obtain significantly more accuracy in the computed solutions, at least for not too ill-conditioned matrices. For symmetric systems we mostly obtained nearly maximum accuracy (13 to 15 correct decimal digits), although not verified.

Sufficiently supported in hardware, the exact scalar product can be computed as fast as an ordinary scalar product. Thus, we urgently postulate this technique to be implemented in future processors in hardware.

However, if we need to guarantee the computed solutions, this arithmetical improvement does often not benefit to obtain small error bounds. For this purpose,

 $^{^2}$ "A work of this kind actually never finishes. You have to declare it finished when you did all in your power, dependent on time and circumstances."

or when the condition number is too large, we have to switch to higher precision numbers. This enables us to solve practically arbitrary ill-conditioned systems with almost any accuracy and guaranteed error bounds (if needed).

Considering the results of this work, it should be discussed whether hardware manufacturers should be asked to develop hardware support for multiple precision arithmetics, maybe solely based on integer arithmetic. The number type might consist of a few bytes for the sign, exponent, and some status information, and a, propably variable, number of bytes for the mantissa. Alternatively, we could utilize staggered precision numbers, which would greatly suffice from a hardware supported exact scalar product. With this approach, we only had to extend modern computer architectures by one operation to get a high performance multiple precision arithmetic.

Combining these techniques, it should easily be possible, not only to save iterations (as we always did in our tests) but also to save real computing time, while simultaneously getting more accurate solutions. Additionally, the verification process will speed up significantly due to its extensive usage of exact scalar products.



Used Matrices

Here we list several test matrices used throughout this thesis. The matrices are taken from the Matrix-Market [85] and are alphabetically ordered.

Matrix A.1: add32							
Computer component design, 32-bit adder							
S. Hamm, Motorola Inc. Semicond. Systems Design Technology							
Size	Туре	Properties					
dim = 4960	real,	$\ \cdot \ _F = 1.6$					
nnz = 23884	unsymmetric	$cond = 2.14 \cdot 10^2$					
bandw = 4030/4030		$\sigma_{\min} = 2.99 \cdot 10^{-4}$					

Matrix A.2: bcsst	k08						
Structural engineering							
John Lewis, Boeing Computer Services							
Size	Туре	Properties					
dim = 1074	real,	$\ \cdot\ _F = 1.0 \cdot 10^{11}$					
nnz = 7017	symmetric	$cond = 4.7 \cdot 10^7$					
bandw = 591/591		$\sigma_{\min} = 2.1 \cdot 10^3$					

1	Matrix	Δ 3.	e30r5000	
	IVIALIIA	A.J.	60010000	

Driven	cavity,	30x30	elements,	Re = 5000
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Andrew Chapman, University of Minnesota				
Size	Туре	Properties		
dim = 9661	real,	$\ \cdot\ _F = 2.20 \cdot 10^3$		
nnz = 306356	unsymmetric	$cond = 1.27 \cdot 10^{11}$		
bandw = 342/342		$\sigma_{\rm min} = 3.7 \cdot 10^{-12}$		

(Matrix A.4: e40r5000

Driven cavity , 40x40 elements, Re=5000

Andrew Chapman, University of Minnesota				
Size	Туре	Properties		
dim = 17281	real,	$\ \cdot\ _F = 2.10 \cdot 10^3$		
nnz = 553956	unsymmetric	$cond = 7.68 \cdot 10^{10}$		
bandw = 452/452		$\sigma_{\rm min} = 1.5 \cdot 10^{-13}$		

Matrix A.5: fidap009

Finite element modeling of fluid dynamics			
Isaac Hasbani, Fluid Dynamics International			
Size	Туре	Properties	
dim = 3363	real,	$\ \cdot\ _F = 3.00 \cdot 10^{10}$	
nnz = 99397	symmetric	$cond = 4.05 \cdot 10^{13}$	
bandw = 86/86		$\sigma_{\rm min} = 2.27 \cdot 10^{-4}$	

Matrix A.6: fs680	1			
Chemical kinetics pre-	Chemical kinetics problems			
Alan Curtis, Comput	ter Science and System	ns Division		
Size	Туре	Properties		
dim = 680	real,	$\ \cdot\ _F = 1.2 \cdot 10^{14}$		
nnz = 2646	unsymmetric	$cond = 2.1 \cdot 10^4$		
bandw = 561/281		$\sigma_{\min} = 7.44 \cdot 10^8$		

Matrix A.7: mcca				
Nonlinear radiative t	ransfer and statistical	equilibrum in astro-		
physics				
Mats Carlson, Institute of Theoretical Astrophysics				
Size	Туре	Properties		
dim = 180	real,	$\ \cdot\ _F = 2.3 \cdot 10^{19}$		
nnz = 2659	unsymmetric	$cond = 3.6 \cdot 10^{17}$		
bandw = 43/66		$\sigma_{\min} =$		

Matrix A.8:	pde900
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Elliptic partial differential equation

H. Elman, University of Maryland		
Size	Туре	Properties
dim = 900	real,	$\ \cdot\ _F = 2.2 \cdot 10^2$
nnz = 4380	unsymmetric	$cond = 8.73 \cdot 10^2$
bandw = 31/31		$\sigma_{\min} = 4.43 \cdot 10^{-2}$

Matrix A.9: pde2961

Elliptic partial differential equation

H. Elman, University of Maryland		
Size	Туре	Properties
dim = 2961	real,	$\ \cdot\ _F = 2.2 \cdot 10^2$
nnz = 14585	unsymmetric	$cond = 9.49 \cdot 10^2$
bandw = 48/48		$\sigma_{\min} = 4.23 \cdot 10^{-2}$

Matrix A.10:pores2Reservoir modeling

John Appleyard, Harwell Laboratory			
Size	Туре	Properties	
dim = 1224	real,	$\ \cdot \ _F = 1.5 \cdot 10^8$	
nnz = 9613	unsymmetric	$cond = 3.31 \cdot 10^8$	
bandw = 472/471		$\sigma_{\min} = 2.63 \cdot 10^{-2}$	

Matrix A.11: s3rmt3m1					
Finite element analysis of cylindrical shells					
Reijo Kouhia, Helsinki University of Technology					
Size	Туре	Properties			
dim = 5489	real,	$\ \cdot\ _F = 1.7 \cdot 10^5$			
nnz = 112505	symmetric	$cond = 1.33 \cdot 10^{10}$			
bandw = 192/192		$\sigma_{\min} = 3.50 \cdot 10^{-7}$			

Matrix A.12: tols4000				
Aeroelasticity, stability analysis of an airplane in flight				
S. Godet-Thobie, CERFACS and C. Bès, Aerospatiale				
Size	Туре	Properties		
dim = 4000	real,	$\ \cdot\ _F = 3 \cdot 10^8$		
nnz = 8784	unsymmetric	$cond = 2.7 \cdot 10^7$		
bandw = 2401/2418		$\sigma_{\min} = 3.03 \cdot 10^{-12}$		

Matrix A.13: west	2021			
Chemical engineering plant models				
Art Westerberg, University of Pittsburgh				
Size	Туре	Properties		
dim = 2021	real,	$\ \cdot\ _F = 1.8 \cdot 10^6$		
nnz = 7353	unsymmetric	$cond = 7.50 \cdot 10^{12}$		
bandw = 1888/1309		$\sigma_{\min} = 2.90 \cdot 10^{-8}$		
		·		

Appendix

Free and Open Source Software

⁶⁶ When we speak of free software, we are referring to freedom, not price. Our General Public Licenses are designed to make sure that you have the freedom to distribute copies of free software (and charge for this service if you wish), that you receive source code or can get it if you want it, that you can change the software or use pieces of it in new free programs; and that you know you can do these things.

GNU General Public License (Version 2), 1991

At this place, I wish to thank the hundreds of programmers that spend their time, energy and knowledge in producing free and open source software. This thesis would basically be impossible in this form without these programs. In the following, I enumerate the most important of them, used for writing this thesis and coding the programs.

First of all, I want to mention the operating system itself: LINUX. All text editing was done with XEmacs combined with auctex-mode. For formatting the thesis I used $T_{EX}/I_{A}T_{EX}$ together with a couple of packages and my own document style 'dissbook'. The graphics where created using XFig, The Gimp and gnuplot. Besides the dozens of really helpful little (and large) utilities, my programming environment consisted of the GNU C compiler, egcs, xxgdb and again XEmacs with cc-mode. Additionally I used several libraries, like gmp (GNU multiple precision), doubledouble, profil, BIAS and gtk (GNU toolkit).

Appendix

Curriculum Vitae

Name		Axel Facius
Address		Jenaer Strasse 8, 76139 Karlsruhe
Date of Birth/ Birthplace		June 21, 1969 in Schwäbisch Gmünd
Nationality		German
School Education	1976 - 1980 1980 - 1989	Grundschule (primary school) Gymnasium (high school)
University	1991 - 1997	study of Mathematics, Computer Science and Electrical Engineering at the University of Karlsruhe
	1995	project on "Simulation of Dynamical Oscillators in Neural Networks on Parallel Computers"
	1996-1997	Diploma-Thesis at the Institute of Logic, Complexity, and Deduction Systems, Prof. Menzel, subject: "Reconstruction and Analysis of Independent Components with Neuronal Networks"
Diploma	11/1997	graduation to 'Diplom Technomathematiker'
Thesis	since 1998	work on Ph.D. at the Institute of Applied Mathematics, advisors Prof. Kulisch and Dr. Lohner

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