



COMPUTATIONAL STRUCTURES FOR BLIND LONG-CODE WCDMA RECEIVERS

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Long-code wideband CDMA receivers are for computational reasons usually based on simple matched-filter techniques, and hence suffer from multiaccess interference. To mitigate this problem, we propose computationally efficient decorrelating RAKE and MMSE receivers for the uplink of long-code CDMA systems. The decorrelating matched filter eliminates multiaccess interference by code matrix inversion. Channel parameters and data symbols are then estimated jointly via rank-one decompositions. The feasibility of the proposed schemes hinges upon their efficient implementation in terms of time-varying state space realizations, with a complexity comparable to that of the conventional RAKE receiver.

1. INTRODUCTION

Current receivers for long-code (or aperiodic spreading code) wideband CDMA are typically based on RAKE receivers, i.e. banks of matched filters which correlate the received data with the desired user's code, followed by a combining of the outputs (RAKE fingers). Since multiuser interference is not completely cancelled, the performance degrades, especially when the network is heavily loaded and power control imperfect.

In this paper, we consider the uplink and assume that the base station knows all codes. We model multiuser interference explicitly and propose a blind decorrelating RAKE and MMSE receiver to estimate the channel and user symbols, based on all samples in a frame. The decorrelating RAKE was presented earlier by us in [1, 2] with an emphasis on identifiability and performance; the MMSE receiver is similar. Here, we also take the noise covariance into account and focus in particular on the efficient implementation of these receivers.

The decorrelating matched filter asks for the inversion of a code matrix whose long dimension is equal to the number of chips over the complete frame. This is a formidable task, but fortunately, the sparse structure of this matrix admits computationally efficient techniques. As an application of the work in [3] on the inversion of infinite-size matrices, we derive efficient time-varying state-space implementations of the various steps in the algorithm.

Blind channel estimation and multiuser detection for long code CDMA has been considered by a number of other authors. In particular, second order moment techniques [4–8] rely on the convergence of time averages, which often requires hundreds to thousands of symbols. Weiss and Friedlander [9] focus on the down link, where users can be considered synchronous.

2. DATA MODEL

We consider the uplink of a slotted system with I asynchronous users. In a frame, the i -th user transmits a vector \mathbf{s}_i consisting of K_i symbols s_{ik} . Each symbol is spread by an aperiodic code \mathbf{c}_{ik} of length G_i . After multipath propagation over a channel with length

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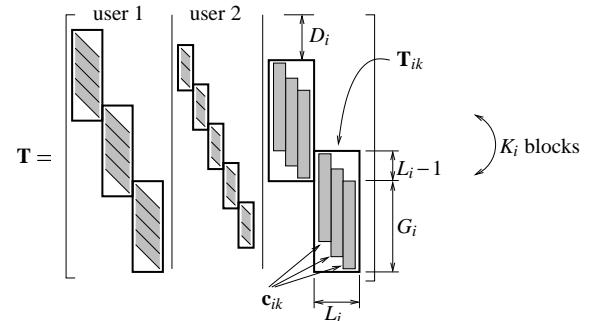


Figure 1. Structure of the code matrix \mathbf{T} .

L_i chips and relative delay D_i , pulse shape matched filtering and sampling at the chiprate¹, the receiver stacks the received samples in a frame in a vector \mathbf{y} . The contribution of s_{ik} is a linear combination of the transmitted signal $\mathbf{c}_{ik} s_{ik}$, plus delays of it, properly scaled by the L_i channel coefficients collected in a vector \mathbf{h}_i , or

$$\mathbf{y}_{ik} = \mathbf{T}_{ik} \mathbf{h}_i \mathbf{s}_{ik}, \quad k = 1, \dots, K_i.$$

\mathbf{T}_{ik} is a Toeplitz matrix whose L_i columns consist of shifts of the code \mathbf{c}_{ik} . Including all users and the noise, we have

$$\begin{aligned} \mathbf{y} &= \mathbf{THs} + \mathbf{w} \\ \mathbf{T} &:= [\mathbf{T}_{11} \dots \mathbf{T}_{1,G_1}, \dots, \mathbf{T}_{I1} \dots \mathbf{T}_{I,G_I}] \\ \mathbf{H} &:= \text{diag}(\mathbf{I}_{K_1} \otimes \mathbf{h}_1, \dots, \mathbf{I}_{K_I} \otimes \mathbf{h}_I), \end{aligned} \quad (1)$$

where matrix \mathbf{H} is block diagonal with $\mathbf{I} \otimes \mathbf{h}_i$ as the i th block, vector \mathbf{s} is a stacking of all symbol vectors, and \mathbf{w} is a vector representing the additive Gaussian noise. The structure of the code matrix \mathbf{T} is illustrated in figure 1. Note that different spreading gains G_i are part of the model. Multiple antennas are a simple extension.

We will assume that the code matrix \mathbf{T} is known, “tall” and has full column rank. This implies that the receiver knows the codes, the delay offsets D_i , and the number of paths L_i of all users.

3. BLIND RECEIVER ALGORITHMS

3.1. Conventional RAKE

The conventional RAKE receiver consists of a bank of matched filters and projects the received signal into the code domains of the individual users, by correlating with several shifts of the code vectors, or $\mathbf{r} = \mathbf{T}^H \mathbf{y}$. Since the codes are not exactly orthogonal (let alone shift-orthogonal), $\mathbf{T}^H \mathbf{T} \neq \mathbf{I}$, and contributions of each user enter into the projections of any other user. This makes the performance interference-limited.

3.2. Decorrelating RAKE

The proposed decorrelating RAKE uses a decorrelating matched filter, or $\mathbf{T}^\dagger = (\mathbf{T}^H \mathbf{T})^{-1} \mathbf{T}^H$. This removes all multi-user interference. The output of the decorrelating matched filter is given by

$$\mathbf{u} = \mathbf{T}^\dagger \mathbf{y} = \text{diag}(\mathbf{I} \otimes \mathbf{h}_1, \dots, \mathbf{I} \otimes \mathbf{h}_I) \mathbf{s} + \mathbf{n}, \quad (2)$$

¹Oversampling is equivalent to a system with multiple receive antennas.

where $\mathbf{n} = \mathbf{T}^\dagger \mathbf{w}$ is now a colored noise vector.

After computing \mathbf{u} , we estimate the channel and the data symbols, blindly and independently for each user. Partition \mathbf{u} into segments \mathbf{u}_{ik} of length L_i . The structure of \mathbf{u} in (2) implies that \mathbf{u}_{ik} corresponds to symbol k of user i ,

$$\mathbf{u}_{ik} = \mathbf{h}_i s_{ik} + \mathbf{n}_{ik}, \quad k = 1, \dots, K_i, \quad (3)$$

and is free from multiuser interference. Collecting all data for user i gives $\mathbf{U}_i = [\mathbf{u}_{i1}, \dots, \mathbf{u}_{iK_i}] = \mathbf{h}_i \mathbf{s}_i^T + \mathbf{N}_i$. This is a rank-1 data model, and estimates of \mathbf{h}_i and \mathbf{s}_i (with an unknown scaling factor) are found from a rank-one factorization of \mathbf{U}_i . In other words, denoting

$$\boldsymbol{\Psi}_i := \frac{1}{K_i} \sum_{k=1}^{K_i} \mathbf{u}_{ik} \mathbf{u}_{ik}^H, \quad (4)$$

we obtain the least squares estimates

$$\hat{\mathbf{h}}_i = \arg \max_{\|\mathbf{g}\|=1} \mathbf{g}^H \boldsymbol{\Psi}_i \mathbf{g}, \quad \hat{s}_{ik} = \hat{\mathbf{h}}_i^H \mathbf{u}_{ik}. \quad (5)$$

The solution $\hat{\mathbf{h}}_i$ is given as the dominant eigenvector of $\boldsymbol{\Psi}_i$. The scaling ambiguity is resolved by a single pilot symbol. See [1, 2] for further results and performance simulations.

3.3. Whitened Estimator

The channel and symbol estimator given in (5) did not take into account that the noise process \mathbf{n}_{ik} is colored, both in k and in its components. If we ignore the coloring in k , then a simple whitening approach can be applied. Specifically, since $\mathbf{n} = \mathbf{T}^\dagger \mathbf{w}$, we have that $\mathbf{n}_{ik} \sim \mathcal{N}(\mathbf{0}, \sigma^2 \boldsymbol{\Sigma}_{ik})$ where $\boldsymbol{\Sigma}_{ik}$ is an $L_i \times L_i$ submatrix on the diagonal of $\mathbf{T}^\dagger (\mathbf{T}^\dagger)^H$. We have

$$\mathbb{E}(\boldsymbol{\Psi}_i) = \frac{\|\mathbf{s}_i\|^2}{K_i} \mathbf{h}_i \mathbf{h}_i^H + \sigma^2 \boldsymbol{\Delta}_i, \quad \boldsymbol{\Delta}_i := \frac{1}{K_i} \sum_{k=1}^{K_i} \boldsymbol{\Sigma}_{ik}$$

where $\boldsymbol{\Delta}_i$ is a known matrix. The channel can then be estimated from the following modification which whitens the noise on $\boldsymbol{\Psi}_i$:

$$\mathbf{g}^* = \arg \max_{\|\mathbf{g}\|=1} \mathbf{g}^H (\boldsymbol{\Delta}_i^{-1/2} \boldsymbol{\Psi}_i \boldsymbol{\Delta}_i^{-H/2}) \mathbf{g}, \quad \hat{\mathbf{h}}_i = \boldsymbol{\Delta}_i^{1/2} \mathbf{g}^*.$$

The symbol estimator given in (5) is replaced by $\hat{s}_{ik} = \hat{\mathbf{h}}_i^H \boldsymbol{\Sigma}_{ik}^{-1} \mathbf{u}_{ik}$.

3.4. MMSE Receiver

Based on the data model (1), the estimated data sequence by a linear minimum mean square error (MMSE) receiver is known to be

$$\hat{\mathbf{s}} = (\mathbf{H}^H \mathbf{T}^H \mathbf{T} \mathbf{H} + \sigma^2 \mathbf{I})^{-1} \mathbf{H}^H \mathbf{T}^H \mathbf{y}. \quad (6)$$

This receiver can be implemented using the previously estimated channel matrix \mathbf{H} , and assuming that the noise power σ^2 is known. Compared to the decorrelating RAKE, the MMSE can have a significantly improved performance.

4. EFFICIENT IMPLEMENTATIONS

The code matrix \mathbf{T} can be very large. Without an efficient technique to compute and apply the left inverse $\mathbf{T}^\dagger = (\mathbf{T}^H \mathbf{T})^{-1} \mathbf{T}^H$, the proposed receiver structures would not be feasible. Fortunately, \mathbf{T} is sparse. Using the Matlab sparse toolbox, $\mathbf{u} = \mathbf{T}^\dagger \mathbf{y}$ can be computed efficiently via a sparse QR factorization $\mathbf{T} = \mathbf{QR}$, and $\mathbf{u} = \mathbf{R}^{-1} \mathbf{Q}^H \mathbf{y}$, or, avoiding the storage of \mathbf{Q} , as

$$\begin{bmatrix} \mathbf{R} \mathbf{v} \\ \mathbf{0} \varepsilon \end{bmatrix} := \text{qr}([\text{sparse}(\mathbf{T}) \mathbf{y}])$$

$$\mathbf{u} := \mathbf{R} \backslash \mathbf{v}$$

$\mathbf{v} = \mathbf{Q}^H \mathbf{y}$, and $\mathbf{R} \backslash \mathbf{v}$ denotes $\mathbf{R}^{-1} \mathbf{v}$, implemented efficiently via back-substitution. This does not reveal how the sparse computations can actually be implemented in a practical system. It is also unclear how the noise whitening (computation of $\boldsymbol{\Sigma}_{ik}$) can be implemented

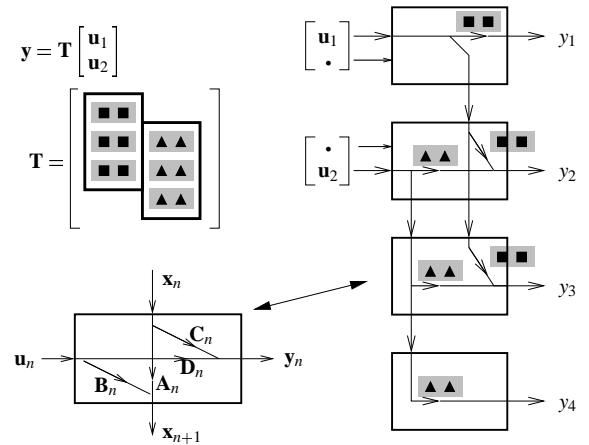


Figure 2. Computational network for $\mathbf{T} = [\mathbf{T}^{(1)} \mathbf{T}^{(2)}]$.

efficiently. Explicit computation of $\boldsymbol{\Sigma} = \mathbf{R}^{-1} \mathbf{R}^{-H}$ is to be avoided because \mathbf{R}^{-1} is not sparse even if \mathbf{R} is. In this section, we show how time-varying state space representations can be used for this purpose. The theory behind it is available in [3].

4.1. State Space Representation of a Matrix

Consider an input signal \mathbf{u} and output signal \mathbf{y} , with arbitrary block-partitioning $\mathbf{u} = [\mathbf{u}_1^T, \dots, \mathbf{u}_N^T]^T$, $\mathbf{y} = [\mathbf{y}_1^T, \dots, \mathbf{y}_N^T]^T$. The partitioning introduces the notion of “time”, or a stage in a computational procedure. The blocks do not need to be of equal size, and some dimensions can even be zero (such a block is denoted by “ \cdot ”).

A time-varying state space realization has the form

$$\begin{cases} \mathbf{x}_{n+1} = \mathbf{A}_n \mathbf{x}_n + \mathbf{B}_n \mathbf{u}_n \\ \mathbf{y}_n = \mathbf{C}_n \mathbf{x}_n + \mathbf{D}_n \mathbf{u}_n \end{cases} \Leftrightarrow \begin{bmatrix} \mathbf{x}_{n+1} \\ \mathbf{y}_n \end{bmatrix} = \mathbf{T}_n \begin{bmatrix} \mathbf{x}_n \\ \mathbf{u}_n \end{bmatrix}, \quad \mathbf{T}_n = \begin{bmatrix} \mathbf{A}_n & \mathbf{B}_n \\ \mathbf{C}_n & \mathbf{D}_n \end{bmatrix}$$

The realization starts at time 1 with $\mathbf{x}_1 = \cdot$ (or: no state), and ends with $\mathbf{x}_{N+1} = \cdot$. Hence $\mathbf{A}_1 = \cdot$, $\mathbf{A}_N = \cdot$, $\mathbf{C}_1 = \cdot$, $\mathbf{B}_N = \cdot$.

A time-varying state-space realization specifies a linear mapping of \mathbf{u} to \mathbf{y} , hence a matrix \mathbf{T} such that $\mathbf{y} = \mathbf{T} \mathbf{u}$. In particular, it defines a factorization of \mathbf{T} into factors \mathbf{T}_n . The inherent causality translates to \mathbf{T} being block-lower triangular. However, by playing with dimensions, *any* matrix can fit this model, as the next examples illustrate. Consider first an arbitrary $N \times L$ matrix \mathbf{T} , with rows \mathbf{t}_n^H . A (trivial) realization that models $\mathbf{y} = \mathbf{T} \mathbf{u}$ is obtained by setting $\mathbf{u}_1 = \mathbf{u}$, $\mathbf{u}_2 = \dots = \mathbf{u}_N = \cdot$ (i.e., the complete input vector is entered at time 1), and

$$\begin{bmatrix} \mathbf{A}_1 & \mathbf{B}_1 \\ \mathbf{C}_1 & \mathbf{D}_1 \end{bmatrix} = \begin{bmatrix} \cdot & \mathbf{I} \\ \cdot & \mathbf{t}_1^H \end{bmatrix}, \quad \begin{bmatrix} \mathbf{A}_n & \mathbf{B}_n \\ \mathbf{C}_n & \mathbf{D}_n \end{bmatrix} = \begin{bmatrix} \mathbf{I} & \cdot \\ \mathbf{t}_n^H & \cdot \end{bmatrix}, \quad \begin{bmatrix} \mathbf{A}_N & \mathbf{B}_N \\ \mathbf{C}_N & \mathbf{D}_N \end{bmatrix} = \begin{bmatrix} \cdot & \cdot \\ \mathbf{t}_N^H & \cdot \end{bmatrix}, \quad n = 2, \dots, N-1$$

As a second example, let $\mathbf{T} = [\mathbf{T}^{(1)} \mathbf{T}^{(2)}]$ be an arbitrary block-partitioned matrix, where $\mathbf{T}^{(1)}$ has realization $\{\mathbf{A}_n^{(1)}, \mathbf{B}_n^{(1)}, \mathbf{C}_n^{(1)}, \mathbf{D}_n^{(1)}\}$ and $\mathbf{T}^{(2)}$ has realization $\{\mathbf{A}_n^{(2)}, \mathbf{B}_n^{(2)}, \mathbf{C}_n^{(2)}, \mathbf{D}_n^{(2)}\}$. Then

$$\mathbf{T}_n = \begin{bmatrix} \mathbf{A}_n^{(1)} & 0 & \mathbf{B}_n^{(1)} & 0 \\ 0 & \mathbf{A}_n^{(2)} & 0 & \mathbf{B}_n^{(2)} \\ \hline \mathbf{C}_n^{(1)} & \mathbf{C}_n^{(2)} & \mathbf{D}_n^{(1)} & \mathbf{D}_n^{(2)} \end{bmatrix}$$

is a realization of \mathbf{T} . Its structure is shown in Fig. 2.

The code matrix \mathbf{T} in our case has a block structure as shown in Fig. 1. By combining the two examples, we can represent any code matrix \mathbf{T} . The number of state space time points is equal to the number of rows of \mathbf{T} . The input vector is partitioned in blocks of L_i entries which enter the system at appropriate time points, determined by the starting points of the individual code blocks. The

state dimension at each time point is the number of nonzero entries in the corresponding row of \mathbf{T} .

4.2. QR Factorization and Inversion in State Space

To compute the left inverse \mathbf{T}^\dagger , our aim is to first compute a QR factorization $\mathbf{T} = \mathbf{QR}$ where $\mathbf{Q}^H \mathbf{Q} = \mathbf{I}$ and \mathbf{R} is square and lower triangular, and then to invert each of the factors: $\mathbf{T}^\dagger = \mathbf{R}^{-1} \mathbf{Q}^H$. The computation of the QR factorization can be done in state space, as is demonstrated by the following theorem.

Theorem 1. (cf. [3], p.156) *Let \mathbf{T} be full column rank and have realization $\{\mathbf{A}_n, \mathbf{B}_n, \mathbf{C}_n, \mathbf{D}_n\}_{n=1,\dots,N}$. Let $\mathbf{Y}_{N+1} = \cdot$ and consider the sequence of (economy-size) QR factorizations*

$$\begin{bmatrix} \mathbf{Y}_{n+1} \mathbf{A}_n & \mathbf{Y}_{n+1} \mathbf{B}_n \\ \mathbf{C}_n & \mathbf{D}_n \end{bmatrix} =: \underbrace{\begin{bmatrix} \mathbf{A}_n^Q & \mathbf{B}_n^Q \\ \mathbf{C}_n^Q & \mathbf{D}_n^Q \end{bmatrix}}_{\mathbf{Q}_n} \begin{bmatrix} \mathbf{Y}_n & 0 \\ \mathbf{C}_n^R & \mathbf{D}_n^R \end{bmatrix}, \quad n = N, N-1, \dots, 1 \quad (7)$$

where \mathbf{Q}_n is isometric ($\mathbf{Q}_n^H \mathbf{Q}_n = \mathbf{I}$), and the right factor is lower triangular (possibly staircase) and partitioned such that \mathbf{Y}_n has the same number of columns as \mathbf{A}_n , \mathbf{D}_n^R has the same number of columns as \mathbf{D}_n , and both \mathbf{Y}_n and \mathbf{D}_n^R are full row rank. Then all \mathbf{D}_n^R are square, lower triangular and invertible. Further define the realizations

$$\mathbf{Q}_n = \begin{bmatrix} \mathbf{A}_n^Q & \mathbf{B}_n^Q \\ \mathbf{C}_n^Q & \mathbf{D}_n^Q \end{bmatrix}, \quad \mathbf{R}_n = \begin{bmatrix} \mathbf{A}_n & \mathbf{B}_n \\ \mathbf{C}_n^R & \mathbf{D}_n^R \end{bmatrix}.$$

Then $\mathbf{T} = \mathbf{QR}$, where \mathbf{Q} is specified by \mathbf{Q}_n and is isometric ($\mathbf{Q}^H \mathbf{Q} = \mathbf{I}$), and \mathbf{R} is specified by \mathbf{R}_n and is lower triangular and invertible. The structure of the factorization is shown in Fig. 3(a). Note that in our application, \mathbf{A}_n and \mathbf{B}_n are trivial: embeddings of identity matrices of appropriate sizes. Hence the multiplication by \mathbf{Y}_{n+1} is trivial and the only actual work in (7) is the QR factorization.

Theorem 2. *Suppose that \mathbf{R} is a square invertible lower triangular matrix. Then its inverse is lower triangular too. If \mathbf{R} has state space realization*

$$\mathbf{R}_n = \begin{bmatrix} \mathbf{A}_n^R & \mathbf{B}_n^R \\ \mathbf{C}_n^R & \mathbf{D}_n^R \end{bmatrix}, \quad n = 1, \dots, N$$

then $\mathbf{S} := \mathbf{R}^{-1}$ has state space realization

$$\mathbf{S}_n = \begin{bmatrix} \mathbf{A}_n^R - \mathbf{B}_n^R \mathbf{D}_n^{R-1} \mathbf{C}_n^R & \mathbf{B}_n^R \mathbf{D}_n^{R-1} \\ -\mathbf{D}_n^{R-1} \mathbf{C}_n^R & \mathbf{D}_n^{R-1} \end{bmatrix}, \quad n = 1, \dots, N$$

PROOF Note that $\mathbf{R}\mathbf{u} = \mathbf{y} \Leftrightarrow \mathbf{S}\mathbf{y} = \mathbf{u}$, hence \mathbf{S} maps \mathbf{y} to \mathbf{u} . Since \mathbf{S} is lower triangular (causal),

$$\mathbf{y}_n = \mathbf{C}_n^R \mathbf{x}_n + \mathbf{D}_n^R \mathbf{u}_n \Leftrightarrow \mathbf{u}_n = -\mathbf{D}_n^{R-1} \mathbf{C}_n^R \mathbf{x}_n + \mathbf{D}_n^{R-1} \mathbf{y}_n$$

Backsubstitution in $\mathbf{x}_{n+1} = \mathbf{A}_n^R \mathbf{x}_n + \mathbf{B}_n^R \mathbf{u}_n$ gives the result. \square

The left-inverse of the isometric factor \mathbf{Q} is \mathbf{Q}^H , with *anti-causal* state space realization (backward recursion)

$$\begin{cases} \mathbf{x}_n = \mathbf{A}_n^{QH} \mathbf{x}_{n+1} + \mathbf{C}_n^{QH} \mathbf{u}_n \\ \mathbf{y}_n = \mathbf{B}_n^{QH} \mathbf{x}_{n+1} + \mathbf{D}_n^{QH} \mathbf{u}_n \end{cases} \quad n = N, N-1, \dots, 1.$$

The preceding theorems can be used to invert more general matrices, in particular the code matrix \mathbf{T} . We obtain an implementation of $\mathbf{T}^\dagger = \mathbf{S}\mathbf{Q}^H$ in factored form, where \mathbf{T}^\dagger , \mathbf{R} and \mathbf{Q} are never explicitly evaluated. The structure of the computational network is shown in Fig. 3(b). As is seen from this structure, the “complexity” of \mathbf{T} and \mathbf{T}^\dagger is the same, even if \mathbf{T}^\dagger is a full matrix without visible sparse structure.

Assume for simplicity that all I users have the same code length G , number of RAKE fingers L , and number of symbols in the frame K . The complexity of computing the state-space representation of \mathbf{T}^\dagger is in the order of $GK(IL)^2$ operations, which is linear in K and comparable to the complexity of a decorrelating receiver in the short code case. The storage requirement of \mathbf{T}^\dagger in state-space factored form is about 2 times the number of non-zero entries in \mathbf{T} , or order $GKIL$. The complexity of applying \mathbf{T}^\dagger to the observation vector is also order $GKIL$. This is the same as the complexity of applying the matched filter \mathbf{T}^H . In contrast, note that \mathbf{T}^\dagger is a full matrix, with GK^2IL entries. Computing \mathbf{T}^\dagger directly requires order $GK^3I^2L^2$ operations, and applying it to a vector requires GK^2IL operations. The benefit in complexity of using state space representations is thus in the order of K^2 and K , respectively.

4.3. Computation of Σ_{ik}

In the computation of the noise covariance, expressions for Σ_{ik} are needed. We can apply the following theorem:

Theorem 3. *Let \mathbf{T} have state space realization $\{\mathbf{A}_n, \mathbf{B}_n, \mathbf{C}_n, \mathbf{D}_n\}$. A realization for the lower triangular part of $\mathbf{N} := \mathbf{TT}^H$ is given by*

$$\mathbf{N}_n = \begin{bmatrix} \mathbf{A}_n & \mathbf{A}_n \mathbf{A}_n^H + \mathbf{B}_n \mathbf{B}_n^H \\ \mathbf{C}_n & \mathbf{C}_n \mathbf{A}_n \mathbf{C}_n^H + \mathbf{D}_n \mathbf{D}_n^H \end{bmatrix}$$

where \mathbf{A}_n is specified by the forward recursion

$$\mathbf{A}_1 = \cdot; \quad \mathbf{A}_{n+1} = \mathbf{A}_n \mathbf{A}_n^H + \mathbf{B}_n \mathbf{B}_n^H, \quad n = 1, 2, \dots, N,$$

PROOF By inspection of Fig. 3(c) and following the mapping of $\mathbf{x}_n, \mathbf{u}_n$ to $\mathbf{x}_{n+1}, \mathbf{y}_n$. The causal part of the state is \mathbf{x}_n , the non-causal part is \mathbf{x}'_n , and \mathbf{A}_n represents the transfer of \mathbf{x}'_n to \mathbf{x}_n . (A formal proof appears in [3, p.366].) \square

The preceding recursions are useful in the computation of the noise covariance after the decorrelating matched filter. If \mathbf{w} is a white noise vector with power normalized to $\sigma^2 = 1$, and $\mathbf{n} = \mathbf{T}^\dagger \mathbf{w} = (\mathbf{T}^H \mathbf{T})^{-1} \mathbf{T}^H \mathbf{w}$, then the covariance of \mathbf{n} is given by

$$\Sigma := \mathbf{E}(\mathbf{n} \mathbf{n}^H) = (\mathbf{T}^H \mathbf{T})^{-1} = \mathbf{S} \mathbf{S}^H$$

where $\mathbf{T} = \mathbf{QR}$ and $\mathbf{S} = \mathbf{R}^{-1}$. A state space realization of \mathbf{S} was derived before. Thus, theorem 3 (applied to \mathbf{S}) gives a recursion to compute a realization for the lower part of $\mathbf{S} \mathbf{S}^H$. The upper part is simply the transpose.

In the identification algorithm in section 3.3, we are only interested in the main (block)-diagonal of $\mathbf{E}(\mathbf{n} \mathbf{n}^H)$ (the auto-covariances of size $L_i \times L_i$). In this case, it suffices to compute

$$\mathbf{E}(\mathbf{n}_i \mathbf{n}_i^H) = \mathbf{C}_n^S \mathbf{A}_n \mathbf{C}_n^{SH} + \mathbf{D}_n^S \mathbf{D}_n^{SH}, \quad \mathbf{A}_{n+1} = \mathbf{A}_n^S \mathbf{A}_n^{SH} + \mathbf{B}_n^S \mathbf{B}_n^{SH}$$

4.4. Computation of the MMSE Receiver in State Space

Recall the MMSE receiver (6). It is known that equations of this form can be efficiently computed via a QR factorization. Indeed, note that

$$\begin{aligned} \hat{\mathbf{s}} &= (\mathbf{H}^H \mathbf{T}^H \mathbf{T} \mathbf{H} + \sigma^2 \mathbf{I})^{-1} \mathbf{H}^H \mathbf{T}^H \mathbf{y} \\ &= (\mathbf{H}^H \mathbf{T}^H \mathbf{T} \mathbf{H} + \sigma^2 \mathbf{I})^{-1} \begin{bmatrix} \mathbf{H}^H \mathbf{T}^H & \sigma \mathbf{I} \end{bmatrix} \begin{bmatrix} \mathbf{y} \\ \mathbf{0} \end{bmatrix} = \underbrace{\begin{bmatrix} \mathbf{T} \mathbf{H} \\ \sigma \mathbf{I} \end{bmatrix}}_{\mathbf{M}}^\dagger \begin{bmatrix} \mathbf{y} \\ \mathbf{0} \end{bmatrix} \end{aligned} \quad (8)$$

Thus, if $\mathbf{M} =: \mathbf{Q}^M \mathbf{R}^M$ is an economy-size QR factorization for \mathbf{M} (where \mathbf{R}^M is square triangular, and \mathbf{Q}^M is tall and isometric), then

$$\hat{\mathbf{s}} = (\mathbf{R}^M)^{-1} (\mathbf{Q}^M)^H \begin{bmatrix} \mathbf{y} \\ \mathbf{0} \end{bmatrix}.$$

The QR factorization and factor inversion can be done in state space as before. Thus, $\hat{\mathbf{s}}$ is the output of a computational structure similar to the one in Fig. 3(b). The only new aspect is the derivation of a realization for \mathbf{M} .

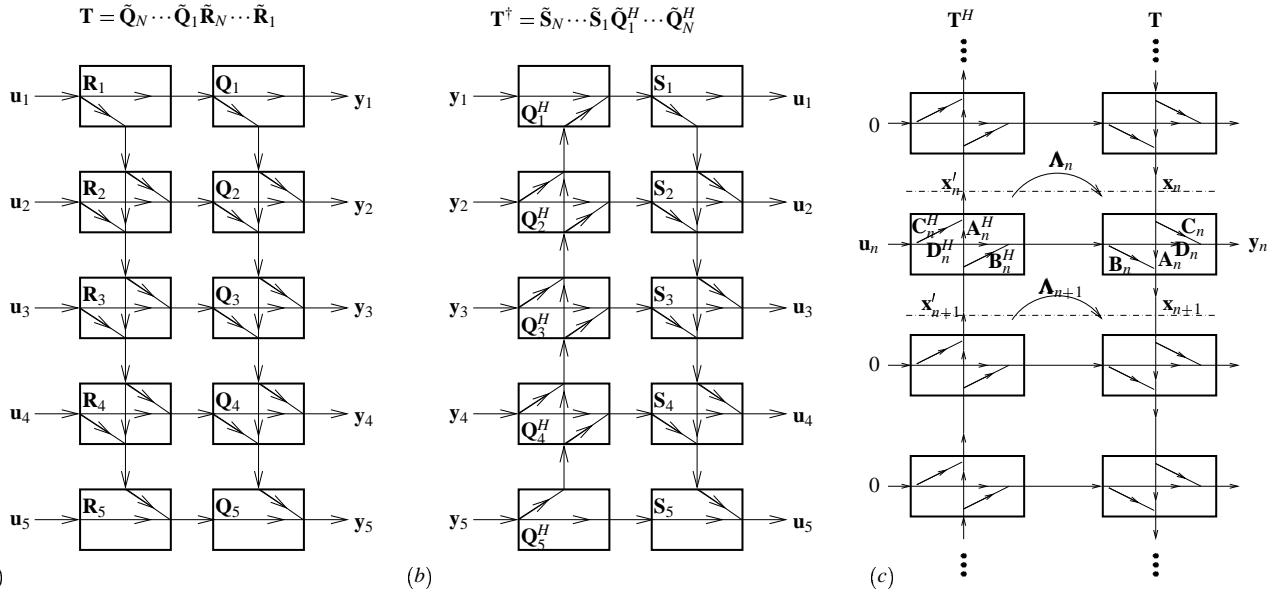


Figure 3. (a) Structure of the QR factorization, (b) structure of the inverse. Note that the inverse is not causal. (c) Structure of $\mathbf{T}\mathbf{T}^H$.

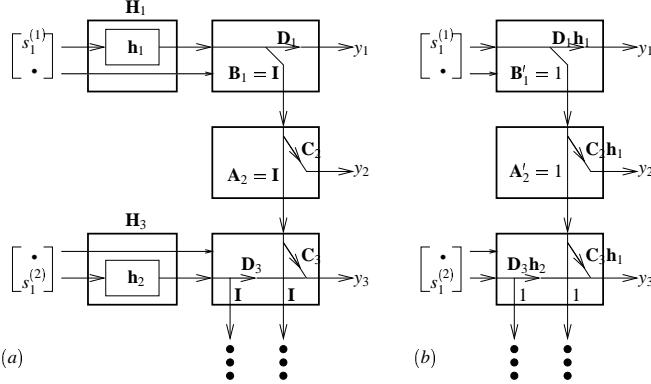


Figure 4. Realization of \mathbf{TH} : (a) direct realization, (b) reduced state dimensions.

A realization $\{\mathbf{A}_n, \mathbf{B}_n, \mathbf{C}_n, \mathbf{D}_n\}$ for \mathbf{T} is already known. \mathbf{H} is block-diagonal, with blocks \mathbf{h}_i matching the inputs of \mathbf{T} . Define

$$\mathbf{H}_n := \begin{bmatrix} \mathbf{h}_{1,n} \\ \vdots \\ \mathbf{h}_{I,n} \end{bmatrix}, \quad \mathbf{b}_{i,n} := \begin{cases} \mathbf{h}_i, & \mathbf{T} \text{ has an input for user } i \text{ at } n \\ \cdot, & \text{otherwise.} \end{cases}$$

A realization for \mathbf{TH} is then given by

$$(\mathbf{TH})_n = \begin{bmatrix} \mathbf{A}_n & \mathbf{B}_n \mathbf{H}_n \\ \mathbf{C}_n & \mathbf{D}_n \mathbf{H}_n \end{bmatrix}, \quad n = 1, \dots, N.$$

This is illustrated in Fig. 4(a). Finally, a realization for \mathbf{M} is simply obtained by extending the D -matrix by $\sigma \mathbf{I}$:

$$\mathbf{M}_n = \begin{bmatrix} \mathbf{A}_n | \mathbf{B}_n \mathbf{H}_n \\ \mathbf{C}_n | \mathbf{D}_n \mathbf{H}_n \\ \mathbf{0} | \sigma \mathbf{I} \end{bmatrix}, \quad n = 1, \dots, N.$$

A few remarks are in order. Since we have already performed a QR-factorization $\mathbf{T} = \mathbf{QR}$, with \mathbf{R} having smaller dimensions than \mathbf{T} , we can exploit this. Recall (8). Since $\mathbf{T}^H \mathbf{T} = \mathbf{R}^H \mathbf{R}$, we can write

$$\hat{\mathbf{s}} = (\mathbf{H}^H \mathbf{R}^H \mathbf{R} \mathbf{H} + \sigma^2 \mathbf{I})^{-1} \mathbf{H}^H \mathbf{R}^H \mathbf{Q}^H \mathbf{y} = \begin{bmatrix} \mathbf{R} \mathbf{H} \\ \sigma \mathbf{I} \end{bmatrix}^\dagger \begin{bmatrix} \mathbf{Q}^H \mathbf{y} \\ \mathbf{0} \end{bmatrix}.$$

Thus define $\mathbf{v} = \mathbf{Q}^H \mathbf{y}$ (it was already computed for the channel estimation step), and use the realization for \mathbf{R} in place of that of \mathbf{T} , and \mathbf{v} in place of \mathbf{y} .

Secondly, the shown realization for \mathbf{TH} (or \mathbf{RH}) is not minimal. We can reduce the state dimensions from L_i per user-input to 1. This is illustrated in figure 4(b).

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