ON THE SAMPLE COMPLEXITY OF GRAPHICAL MODEL SELECTION FROM NON-STATIONARY SAMPLES

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ABSTRACT
We characterize the sample size required for accurate graphical model selection for a system which is observed via samples (measurements) forming a non-stationary vector-valued time series. In particular, the observed data is modelled as a vector-valued zero-mean Gaussian random process whose samples are uncorrelated but have different covariance matrices. This model contains as special cases the standard setting of i.i.d. samples as well as the case of samples forming a stationary time series. More generally, our approach applies to time series data for which efficient decorrelation transforms, such as the Fourier transform for stationary time series, are available. By analyzing a particular model selection method, we derive a sufficient condition on the required sample size for accurate graphical model selection based on non-stationary data.

1. INTRODUCTION

A powerful approach to managing massive datasets (big data) is based on network or graph representations of the datasets [1]–[4]. Examples of networked data are found in signal processing where signal samples can be arranged as a chain, in image processing with pixels arranged on a grid, in wireless sensor networks where measurements conform to sensor proximity [1]. Organising data using networks is also used in knowledge bases (graphs) whose items are linked by relations [5], [6].

Using network models is beneficial from a computational and statistical perspective. Indeed, network models for data lend naturally to highly scalable learning algorithms in the form of message passing on the data network [7]. Moreover, the network structure allows to borrow statistical strength across different localized high-dimensional statistical models which are associated with individual data points (nodes) [2], [8]. Finally, network models provide a high level of flexibility in order to cope with heterogeneous datasets composed of different data types (e.g., mixtures of audio, video and text data).

In some applications, the network structure underlying the data is not known explicitly but has to be learned in a data-driven fashion. This task can be accomplished in a principled way by using probabilistic graphical models (PGM) [3], [9]. Within a PGM, we interpret data points as realizations of random variables. A particular type of PGM is based on representing the conditional independence relations between individual data points using a network structure (graph) [9], [10]. The problem of estimating the network structure of a PGM from observed data is known as graphical model selection (GMS).

Many efficient methods have been proposed for GMS for data which is modelled as sequences of i.i.d. realizations of some underlying random vector [11]–[13]. The extension of GMS from the i.i.d. setting to cope with correlations between vector samples using stationary process models has been studied in [14]–[18]. A robust GMS method which is able to cope with outliers is proposed in [19]. In this paper, we consider the extension of GMS to non-stationary time series data. As we will detail below, our approach includes GMS for stationary time series as a special case.

It is of practical relevance for the usage of GMS methods to understand the fundamental requirements on the available data such that accurate GMS is possible. For data which can be modelled as i.i.d. realizations of a Gaussian random vector (Gaussian Markov random field), the required sample size is well understood. A lower bound on the sample size has been obtained by [20], which does not place any computational constraints on the GMS method. Remarkably, this lower bound can be achieved by computationally tractable convex optimization methods [21] proving them as optimal in terms of sample size requirement. By adapting the information-theoretic approach of [20], a lower bound on the sample size required for accurate GMS from data conforming to a stationary random process model is presented in [16].

Contribution. Our focus is on the required sample (data) size which allows for accurate GMS. In contrast to most existing work, we study GMS for data which cannot be well modelled as a stationary random process. To this end, we propose a simple but useful probabilistic model for non-stationary data whose statistical properties vary over time or space (see Section 2). This model requires that samples can be grouped into blocks (of known size) within which the samples can be considered as i.i.d. Our model includes, as important special cases, the case of i.i.d. data as well as data forming a stationary time series. Moreover, the model also applies to data which can be represented as either cyclostationary [22], locally stationary [23] or underspread random processes [24]. Thus, in contrast to existing GMS
methods [14], [15], [18], [25], [26] which require stationary time series, we consider GMS from non-stationary time series data.

In general, the process model used in this paper is applicable whenever an efficient decorrelation transformation, which allows one to turn the raw data into blocks of i.i.d. random vectors, is available. An important example is the Fourier transform of a stationary time series which can be well approximated as block-wise i.i.d. samples (see Section 2). Our process model has also been used in [27], [28] in the context of a bioinformatics application. However, while [27], [28] aims at learning different graphical models for each block, we are interested in learning a single global graphical model for all blocks.

The main focus of [27], [29] is the design of computationally feasible GMS methods (e.g., based on convex optimization). Instead, our aim is not the design of a computationally tractable (“polynomial time”) GMS method but rather a characterization of the required amount of data (sample size) for reliable GMS. To this end we provide a careful analysis of a computationally intractable neighborhood regression method which amounts to an exhaustive search for conditional dependencies between two particular data points (represented by two nodes in the PGM), when conditioning on all remaining data points.

Our conceptual approach to GMS extends the sparse neighborhood regression approach put forward in [30] for GMS from i.i.d. samples to the non-stationary setting. However, while [30] proposes a computationally attractive convex relaxation of sparse neighborhood regression using a Lasso-based estimator, we are mainly interested in the fundamental limits on the required sample size without constraining the computational complexity of the GMS method.

The main contribution of this work is a precise characterization of the sample size required for accurate GMS from non-stationary data. In particular, we show that the required sample size depends crucially on the minimum average connection strength between the individual process components. If this quantity is sufficiently large, accurate GMS is possible even in the high-dimensional regime, where the length of the vector samples might (drastically) exceed the number of available training samples (data points).

**Outline.** After formalizing the problem setup in Section 2 we analyze a simple GMS method, which we term sparse neighborhood regression, in Section 3. In particular, for a given sample size and sparsity level of the network structure, we derive an upper bound on the probability that sparse neighborhood regression fails in recovering the correct network structure of the PGM. This upper bound on the error probability implies an upper bound on the required sample size such that GMS is feasible. We verify our theoretical findings by means of numerical experiments in Section 4.

**Notation:** For a vector $\mathbf{x} = (x_1, \ldots, x_d)^T \in \mathbb{R}^d$, the Euclidean and $\infty$-norm are $|\mathbf{x}|_2 := \sqrt{\mathbf{x}^T \mathbf{x}}$ and $|\mathbf{x}|_\infty := \max_i |x_i|$, respectively. The $m$-th largest eigenvalue of a positive semidefinite (psd) matrix $\mathbf{C}$ is $\lambda_m(\mathbf{C})$. Given a matrix $\mathbf{Q}$, we denote its transpose, trace, rank, spectral norm and Frobenius norm by $\mathbf{Q}^T$, $\text{tr}\{\mathbf{Q}\}$, $\text{rank}\{\mathbf{Q}\}$, $||\mathbf{Q}||_2$ and $||\mathbf{Q}||_F$, respectively. For a finite sequence of matrices $\mathbf{Q}_l \in \mathbb{R}^{d \times d}$, with $l = 1, \ldots, B$, we denote by $\text{blkdiag}\{\mathbf{Q}_l\}$ the block diagonal matrix of size $Bd \times Bd$ with the $l$th diagonal block given by $\mathbf{Q}_l$. The identity matrix of size $d \times d$ is $\mathbf{I}_d$. The minimum (maximum) of two numbers $a$ and $b$ is denoted $a \wedge b$ ($a \vee b$). The set of non-negative real (integer) numbers is denoted $\mathbb{R}_+$ ($\mathbb{Z}_+$). The probability of an event $\mathcal{E}$ is $P(\mathcal{E})$. The complement of an event $\mathcal{A}$ is denoted $\mathcal{A}^c$. The expectation of a random variable $y$ is $E\{y\}$.

## 2. PROBLEM FORMULATION

We consider a system which is constituted by $p$ components $\mathbf{x}_i$, for $i = 1, \ldots, p$. In a bioinformatics application, such a system might be a gene regulatory network with the components $\mathbf{x}_i$ representing concentrations of particular genes [31]. The system is observed by acquiring $N$ vector-valued samples $\{\mathbf{x}[n]\}_{n=1}^N$, each sample

$$\mathbf{x}[n] = (x_1[n], \ldots, x_p[n])^T \in \mathbb{R}^p$$

constituted by $p$ scalar “measurements” $x_i[n]$ for $i = 1, \ldots, p$.

The samples $\mathbf{x}[n]$ are modelled as realizations of zero-mean Gaussian random vectors, which are uncorrelated such that

$$E\{\mathbf{x}[n]\mathbf{x}[n']^T\} = 0 \quad \text{for} \quad n \neq n'.$$

The probability distribution of the samples $\mathbf{x}[n]$ is fully specified by the covariance matrices

$$\mathbf{C}[n] := E\{\mathbf{x}[n]\mathbf{x}[n]^T\}. \quad (1)$$

In general, the covariance matrix $\mathbf{C}[n]$ varies with sample index $n$, i.e., $\mathbf{C}[n] \neq \mathbf{C}[n']$ for $n \neq n'$ in general. However, we do not allow for arbitrary variation of the covariance matrix but require it to be constant over blocks of $L$ consecutive samples $\mathbf{x}[n], \ldots, \mathbf{x}[n+L-1]$. We model the observed samples as blocks of i.i.d. Gaussian random vectors,

$$\mathbf{x}[1], \ldots, \mathbf{x}[L], \ldots, \ldots, \mathbf{x}[N-L+1], \ldots, \mathbf{x}[N]. \quad (2)$$

Our goal is to estimate the conditional dependencies between the components $\mathbf{x}_i$ which are represented by the sequences $\mathbf{x}_i[1], \ldots, \mathbf{x}_i[N]$ in [2]. Such a global dependence structure between entire sequences has also been considered in [14]. However, [14] considered stationary time series data, we consider global dependence structure between quantities that are modelled as a non-stationary process of the form (2) (which contains the Fourier transform of stationary time series as a special case).
The vector samples $x[n]$ in (2) are uncorrelated (independent) zero-mean Gaussian vectors with covariance matrix

$$C[n] = C^{(b)}$$

for $n \in \{(b-1)L + 1, \ldots, bL\}$. (3)

For ease of exposition and without essential loss of generality, we henceforth assume the sample size $N$ to be an integer multiple of the block length $L$ (which is assumed fixed and known), such that $N = BL$, with the number $B$ of data blocks. Moreover, we tacitly assume the covariance matrices $C[n]$ to be non-singular (invertible) with inverse \((C[n])^{-1}\) (see Assumption 3 below).

The model (2) reduces to the i.i.d. setting for $B = 1$ and block length $L = N$. In this paper, we study the fundamental limits of accurate GMS based on non-stationary data which conforms to the model (2) with $B > 1$ (the non-stationary setting).

At first glance, the process model (2) might seem overly restrictive as it still requires blocks of consecutive samples to be i.i.d. However, as we will now discuss, the model (2) can be used as an approximation at least for some important classes of random processes. For each of these process classes we are able to identify useful choices for the block length $L$ in (2).

**Stationary Processes.** The model (2) covers the case where the observed samples form a stationary process \([14], [15], [18], [25]\). Indeed, consider a zero-mean Gaussian stationary process $x[n]$ with auto-covariance function

$$R_x[m] := E\{x[n](x[n-m])^T\}$$

and spectral density matrix (SDM) \([25]\)

$$S_x(\theta) := \sum_{m=-\infty}^{\infty} R_x[m] \exp(-j2\pi\theta m).$$

Let

$$\check{x}[k] := (1/\sqrt{N}) \sum_{n=1}^{N} x[n] \exp(-j2\pi(n-1)(k-1)/N)$$

denote the discrete Fourier transform (DFT) of the stationary process $x[n]$. Then, by well-known properties of the DFT (see, e.g., \([22]\)), the vectors $\check{x}[k]$, for $k = 1, \ldots, N$, are approximately uncorrelated Gaussian random vectors with zero mean and covariance matrix $C[k] = S_x(k/N)$. For a stationary process $x[n]$ with (effective) correlation width $W$, the SDM is approximately constant (flat) over a frequency interval of length $1/W$. Thus, the DFT vectors $\check{x}[k]$ approximately conform to the process model (2) with block length $L = N/W$ (since the DFT vectors correspond to SDM samples at evenly spaced frequency points separated by $1/N$).

**Cyclostationary Processes.** As detailed in \([22]\), (discrete-time) cyclostationary processes can be transformed to vector-valued (or multivariate) stationary processes which can then, in turn, be transformed to a process of the form (2) via a DFT.

**Locally Stationary Processes.** The process model (2) applies to locally stationary processes \([23], [33], [34]\). The i.i.d. blocks of $L$ consecutive vector samples in (2) correspond to the homogeneity intervals defined in \([33]\). Particular approaches for optimally choosing the block length $L$ for the model (2) are studied in \([33], [35]\). One important example of locally stationary processes are time-varying autoregressive processes which extend traditional autoregressive process models by allowing time-varying regression coefficients \([32], [36]\).

**Underspread Processes.** The process model (2) can be used for underspread non-stationary processes \([24], [37]\). A continuous-time random process $x(t)$ is underspread if its expected ambiguity function (EAF)

$$\tilde{A}_x(\tau, \nu) := \int_{-\infty}^{\infty} E\{x(t+\tau/2)x^T(t-\tau/2)\} \exp(-j2\pi\nu\tau)dt$$

is well-concentrated around the origin in the $(\tau, \nu)$ plane. In particular, if the EAF $\tilde{A}_x(\tau, \nu)$ of $x(t)$ is (effectively) supported on the rectangle $[\tau_0/2, \tau_0/2] \times [-\nu_0/2, \nu_0/2]$, then the process $x(t)$ is underspread if $\tau_0\nu_0 \ll 1$.

It can be shown that for a suitably chosen prototype function $g(t)$ (e.g., a Gaussian pulse) and grid constants $T$ and $F$, the Weyl-Heisenberg set $\{g^{(n,k)}(t) := g(t-nT)e^{-2\pi\nu Ft}\}_{n,k \in \mathbb{Z}}$ yields zero-mean analysis coefficients $\hat{x}[n,k] = \int_{-\infty}^{\infty} x(t)g^{(n,k)}(t)dt$ which are approximately uncorrelated and provide a complete representation of the process $x(t)$. Moreover, the covariance matrix of $\hat{x}[n,k]$ is approximately equal to the value $W_x(nT,kF)$ of the Wigner-Ville spectrum (WVS) \([33]\)

$$W_x(t,f) := \int_{-\infty}^{\infty} \tilde{A}_x(\tau, \nu) \exp(-2\pi(f\tau - \nu t))d\tau d\nu$$

which can be loosely interpreted as a time-varying power spectral density. For an underspread process whose EAF is effectively supported on $[\tau_0/2, \tau_0/2] \times [-\nu_0/2, \nu_0/2]$, the WVS $W_x(nT,kF)$ is approximately constant over a rectangle of area $\approx 1/(\tau_0\nu_0)$. Thus, the vectors $\hat{x}[n,k]$ approximately conform to the process model (2) with block length $L \approx 1/TF\tau_0\nu_0$.

**Conditional Independence Graph.** We now define a PGM for the observed samples $\{x[n]\}_{n=1}^{N}$ (cf. (2)) by identifying the individual components

$$x_i = (x_i[1], \ldots, x_i[N])^T$$

with the nodes $\mathcal{V} = \{1, \ldots, p\}$ of an undirected simple graph $G = (\mathcal{V}, \mathcal{E})$ (see Figure 1). This graph encodes conditional independence relations between the components $x_i$ and is hence called the conditional independence graph (CIG) of the process $x[n]$. In particular, an edge is absent between nodes $i,j \in \mathcal{V}$, i.e., $(i,j) \notin \mathcal{E}$, if the corresponding process
We highlight that the quantity \( n \) over all samples

Note that the CIG characterization (7) involves a coupling

matrices

read off conveniently from the inverse covariance (precision)

Using a global CIG between data points which are modelled as non-

stationary vector samples is useful for many applications (see

[14], 25–27 and references therein).

Since we model the observed samples \( x[n] \) as realizations

dependencies between the components \( \{ x_i \}_{i \in V \setminus \{i,j\}} \).

We highlight that the CIG \( G \) represents stochastic de-

Fig. 1 Example of a CIG underlying vector-valued samples

\( x[n] = (x_1[n], x_2[n], x_3[n])^T \), for \( n = 1, \ldots, N \) (see (2)).

components \( x_i \) and \( x_j \) are conditionally independent, given

the remaining components \( \{ x_r \}_{r \in V \setminus \{i,j\}} \).

The strength of a connection \( \{i,j\} \in E \) between process components \( x_m \) with \( m \in \{1, \ldots, p\} \setminus \{i,j\} \). The

average connection strength \( \rho_{i,j}^2 \) is closely related to the

average squared partial correlations (conditional correlation coefficients) [20], [21], [30]

\[
\frac{1}{B} \sum_{b=1}^{B} (K_{i,i}^{(b)}[n])^2 / (K_{i,i}^{(b)}[n]K_{j,j}^{(b)}[n]).
\]

In contrast to the squared partial correlation, the CIG

Assumption 1. The average connection strength \( \rho_{i,j}^2 \) (see

(8)) between any two connected components \( x_i \) and \( x_j \) with

\( \{i,j\} \in E \) is lower bounded

\[
\rho_{i,j}^2 \geq \rho_{\text{min}}^2 \quad \text{for every } \{i,j\} \in E,
\]

with some known lower bound \( \rho_{\text{min}}^2 > 0 \).

Given a node \( i \in V \) in the CIG \( G \), we denote its neighbour-

hood and degree as \( N(i) := \{ j \in V \setminus \{i\} : \{i,j\} \in E \} \) and

\( s_i = |N(i)| \), respectively. While in principle, our analysis of

GMS applies to processes with arbitrary CIG structure, our

results will be most useful if the underlying CIG \( G \) is sparse

in the sense of having a small (bounded) maximum node
degree.

Assumption 2. The node degrees in the CIG are bounded

by some sparsity level \( s \) as

\[
\max_{i \in V} s_i \leq s, \quad \text{with } s < (p/3) \wedge (L/3).
\]

We highlight that our approach to GMS for the process

(2) requires a known sparsity level \( s \) for the upper bound

(10). However, in contrast to (39), the sparsity \( s \) is only

required to form an upper bound on the node degrees \( s_i \). In

particular, our approach is able to handle nodes \( i \in G \) which

have smaller degrees \( s_i < s \).

The requirement (10) implies a trade-off between the

block length \( L \) of consecutive i.i.d. samples in (2) and

the sparsity \( s \) of the underlying CIG. In particular, for a
given sample size \( N \), we can tolerate less smoothness (smaller

block length \( L \) in (2), if the underlying CIG is more sparse

(having smaller maximum degree \( s \)).
It will be notationally convenient to assume the samples \( x[n] \) suitably scaled such that the eigenvalues of the covariance matrices \( C[n] \) are bounded with known constants.

**Assumption 3.** The eigenvalues of the covariance matrices \( C[n] \) are bounded as
\[
1 \leq \lambda_l(C[n]) \leq \beta \quad \text{for all } l \in \{1, \ldots, p\} \quad \text{and} \quad n \in \{1, \ldots, N\},
\]
with some known upper bound \( \beta \geq 1 \).

Fixing the lower bound in Assumption 3 to be equal to 1 is not restrictive since we assume the covariance matrices \( C[n] \) to be invertible.

### 3. SPARSE NEIGHBORHOOD REGRESSION

The CIG \( G \) of the process \( x[n] \) in (2) is fully specified by the neighbourhoods of the nodes in the CIG. Indeed, rather trivially, we can determine the CIG by determining the neighbourhoods \( \mathcal{N}(i) \) separately for each node \( i \in \mathcal{V} \). Thus, without loss of generality, we will focus on the sub-problem of determining the neighbourhood \( \mathcal{N}(i) \) of an arbitrary but fixed node \( i \in \mathcal{V} \).

In view of the process model (2) we define, for an arbitrary but fixed block \( b \in \{1, \ldots, B\} \), the \( i \)-th component process as
\[
x_i^{(b)} := (x_{i1}(b-1)L + 1, \ldots, x_{iL}bL)^T \in \mathbb{R}^L.
\]
The process components of different blocks are uncorrelated, i.e.,
\[
E\{x_i^{(b)}(x_j^{(b')})^T\} = 0 \quad \text{for } b \neq b' \quad \text{and any } i, j \in \mathcal{V}.
\]

Elementary properties of multivariate normal distributions (see, e.g., [40, Thm. 3.5.1]) and the fact \( K_{i,j}[n] = 0 \) for \( j \notin \mathcal{N}(i) \) (cf. (7)), yield
\[
x_i^{(b)} = \sum_{j \in \mathcal{N}(i)} a_{ij}x_j^{(b)} + \varepsilon_i^{(b)},
\]
(12)
with the coefficients \( a_{ij} = -K_{i,j}/K_{i,i}^{(b)} \). The error vector \( \varepsilon_i^{(b)} \sim \mathcal{N}(0, (1/K_{i,i}^{(b)})I_L) \) is uncorrelated with the vectors \( \{x_j^{(b)}\}_{j \in \mathcal{N}(i)} \). Note that the random vector \( \sum_{j \in \mathcal{N}(i)} a_{ij}x_j^{(b)} \) in (12) is the minimum mean squared error (MMSE) estimator of \( x_i^{(b)} \) using the random vectors \( \{x_j^{(b)}\}_{j \in \mathcal{N}(i)} \) as observations (see (11)).

Given some index set \( T \subseteq \{1, \ldots, p\} \) with \( \mathcal{N}(i) \setminus T \neq \emptyset \), another application of [40, Thm. 3.5.1] to the component \( \sum_{j \in \mathcal{N}(i)} a_{ij}x_j^{(b)} \) in the decomposition (12) yields
\[
x_i^{(b)} = \sum_{j \in T} c_{ij}x_j^{(b)} + \tilde{x}_i^{(b)} + \varepsilon_i^{(b)},
\]
(13)
with the random vectors \( \tilde{x}_i^{(b)} \), \( \{x_j^{(b)}\}_{j \in T} \) and \( \varepsilon_i^{(b)} \) being jointly Gaussian. Moreover, the random vectors \( \tilde{x}_i^{(b)} \) are uncorrelated with the random vectors \( \{x_j^{(b)}\}_{j \in T} \) and distributed as
\[
\tilde{x}_i^{(b)} \sim \mathcal{N}(0, \hat{\sigma}_i^2 I_L).
\]
Note that the vector \( \tilde{x}_i^{(b)} \) in (13) is the estimation error incurred by the MMSE estimator of the random vector \( \sum_{j \in \mathcal{N}(i)} a_{ij}x_j^{(b)} \) using \( \{x_j^{(b)}\}_{j \in T} \) as observations.

Using (40) Thm. 3.5.1, the variance \( \hat{\sigma}_i^2 \) of (the i.i.d. entries of) \( \tilde{x}_i^{(b)} \) can be obtained as
\[
\hat{\sigma}_i^2 = a^T(\tilde{K}^{(b)})^{-1}a
\]
(15)
with the matrix \( \tilde{K}^{(b)} = ((C^{(b)}_{\mathcal{N}(i)\setminus T})^{-1})_{\mathcal{N}(i)\setminus T} \) and the vector \( a \) as
\[
a_j = -K_{i,j}/K_{i,i}^{(b)}, \quad \text{for } j \in \mathcal{N}(i) \setminus T.
\]
In what follows we will make use of a lower bound on the variance \( \hat{\sigma}_i^2 \) which is due to Assumption 3. Indeed, by Assumption 3 we have \( \lambda_l((\tilde{K}^{(b)})^{-1}) \geq 1, \) for all \( l = 1, \ldots, |\mathcal{N}(i) \setminus T|, \) which implies (see (15)) the lower bound
\[
\hat{\sigma}_i^2 \geq \sum_{j \in \mathcal{N}(i) \setminus T} (K_{i,j}^{(b)}/K_{i,i}^{(b)})^2.
\]
(16)
On the other hand, we can use Assumption 3 to obtain (via (13) and (12)) the upper bound
\[
\hat{\sigma}_i^2 \leq \beta.
\]
(17)
It will be convenient to stack the vectors \( \tilde{x}_i^{(b)} \) (cf. (14)) into a single Gaussian random vector
\[
\tilde{x}_i := (\tilde{x}_i^{(1)})^T, \ldots, (\tilde{x}_i^{(B)})^T \sim \mathcal{N}(0, C_{\tilde{x}_i})
\]
with \( C_{\tilde{x}_i} = \text{blkdiag}(\hat{\sigma}_i^2 I_L)_{b=1}^B \).

The decompositions (12) and (13) suggest a simple strategy for estimating (selecting) the neighbourhoods \( \mathcal{N}(i) \) of the nodes \( i \in \mathcal{V} \) in the CIG \( G \). To this end, let \( P^{(b)}_{TT} \in \mathbb{R}^{L \times L} \) denote the orthogonal projection matrix for the complement of the subspace \( \mathcal{X}_T^{(b)} := \text{span}\{x_j^{(b)}\}_{j \in \mathcal{T} \subseteq \mathcal{I}} \subseteq \mathbb{R}^L \), i.e.,
\[
P^{(b)}_{TT} := I - P^{(b)}_T, \quad \text{with } P^{(b)}_T := \sum_{j=1}^{\dim X_T^{(b)}} u_j^{(b)}(u_j^{(b)})^T,
\]
(19)
with \( \{u_j^{(b)}\}_{j=1}^{\dim X_T^{(b)}} \) being an orthonormal basis for the subspace \( \mathcal{X}_T^{(b)} \subseteq \mathbb{R}^L \). The matrix \( P^{(b)}_T \) in (19) is an orthogonal projection matrix on the subspace \( \mathcal{X}_T^{(b)} \).

According to (12), for any index set \( T \supseteq \mathcal{N}(i) \) (such that \( \mathcal{N}(i) \setminus T = \emptyset \)),
\[
\|P_{TT}^{(b)}x_i^{(b)}\|_2^2 = \|P_{TT}^{(b)}\varepsilon_i^{(b)}\|_2^2 \quad \text{for all } b \in \{1, \ldots, B\}.
\]
(20)
On the other hand, for any index set $\mathcal{T}$ with $\mathcal{N}(i) \setminus \mathcal{T} \neq \emptyset$, \[(13)\] entails the component $\hat{\lambda}$ with some random vector $\tilde{x}_i$. Some of our efforts go into showing that
\[\| \mathbf{P}_\mathcal{T} (\tilde{x}_i + \tilde{\varepsilon}_i) \|_2^2 \approx \| \mathbf{P}_\mathcal{T} \tilde{x}_i \|_2^2 + \| \mathbf{P}_\mathcal{T} \tilde{\varepsilon}_i \|_2^2,\]
for all $b \in \{1, \ldots, B\}$. Thus, according to (20) and (21), if the component $\tilde{x}_i$ in (13) is not too small, the estimator
\[\hat{\mathcal{N}}(i) := \arg\min_{|\mathcal{T}| \leq s} (1/N) \frac{\sum_{b=1}^B \| \mathbf{P}_\mathcal{T} \tilde{x}_i \|_2^2 + \lambda |\mathcal{T}|}{b}, \]
delivers the true neighbourhood, i.e., $\hat{\mathcal{N}}(i) = \mathcal{N}(i)$, with high probability. The penalty term $\lambda |\mathcal{T}|$ in (22) is required since we allow nodes $i \in \mathcal{V}$ in the CIG to potentially have fewer than $s$ neighbours ($|N(i)| < s$)\footnotemark[3] Indeed, the statistic $\| \mathbf{P}_\mathcal{T} \tilde{x}_i \|_2^2$ does not allow to distinguish between different sets $\mathcal{T}$ which contain the neighbourhood $\mathcal{N}(i)$. Therefore, we need to add the penalty term $\lambda |\mathcal{T}|$ in (22) in order to prefer smaller sets $\mathcal{T}$ as an estimate for $\mathcal{N}(i)$.

The estimator (22) performs sparse block-wise least squares regression by approximating the $i$th component $x_i$ (cf. (9)) in a sparse manner (by allowing only $s$ active components) using the remaining process components. Indeed, the summations $\| \mathbf{P}_\mathcal{T} \tilde{x}_i \|_2^2$ in (22) are the errors obtained from the block-wise regression problems
\[\| \mathbf{P}_\mathcal{T} \tilde{x}_i \|_2^2 = \min_{w \in \mathcal{P}} \| x_i - \sum_{j \in \mathcal{T}} w_j x_j \|_2^2,\]

We highlight that the estimator (22) is mainly useful as a theoretical device which allows for a simple performance analysis and, in turn, a characterization of the required sample size for accurate GMS. Using a naive implementation of (22), by searching over all subsets of $\{1, \ldots, p\}$ with size at most $s$, has a complexity which grows exponentially in the sparsity level $s$. Thus, the estimator (22) is typically intractable except for very small sparsity levels $s$ (corresponding to a very sparse CIG). More tractable methods for GMS can be obtained by using convex relaxations of (22) which result in Lasso-type methods (see [27] and Section 4-A).

The estimator (22) itself only delivers an estimate for the neighbourhood of some node $i \in \mathcal{V}$ in the CIG $G$ underlying the process (2). In order to obtain an estimate of the entire CIG, we have to repeatedly apply the estimator (22) to each node $i \in \mathcal{V}$. It might then happen that due to estimation errors, we obtain $i \in \hat{\mathcal{N}}(j)$ but $j \notin \hat{\mathcal{N}}(i)$ for two different nodes $i, j \in \mathcal{V}$. There are different options how to handle such a situation such as insisting in consistency between the neighbourhoods when declaring the presence of an edge (see [30, Eq. (7)]). However, the implementation details for handling such cases are not relevant to our analysis, which aims at sufficient conditions such that (22) delivers the correct neighborhood for all nodes simultaneously (with high probability).

Our main result is an upper bound on the probability of the sparse neighbourhood regression (22) to fail in delivering the correct neighbourhood $\mathcal{N}(i)$, i.e., the error event
\[E_i := \{ \mathcal{N}(i) \neq \hat{\mathcal{N}}(i) \}. \]

**Theorem 1.** Consider the vector samples $x[n]$, for $n = 1, \ldots, N$, conforming to the process model (2) and such that Assumption [1, 2, and 3] are valid. We estimate the neighbourhood $\mathcal{N}(i)$ of an arbitrary but fixed node $i \in \mathcal{V}$ in the CIG via sparse regression (22) with $\lambda = \rho_{\min}^2/6$. Then, if the average connection strength between connected components are sufficiently large such that (see (9))
\[\rho_{\min}^2 \geq 24\beta/L,\]
for any sample size
\[N \geq 864(\beta/\rho_{\min}^2) \log(6ps^2/\eta),\]
the probability of the error event (23) is bounded as $\mathbf{P}\{E_i \} \leq \eta$.

By Theorem 1 the true neighbourhood $\mathcal{N}(i)$ of a node $i \in \mathcal{V}$ can be recovered via (22) with high probability if the samples size $N$ is on the order of $((\beta/\rho_{\min}^2) \log(ps^2)$ (for a fixed error tolerance $\eta$). Therefore, given sufficiently large computational power, GMS via sparse neighbourhood regression (22) is feasible in the high dimensional regime where $N \ll p$.

Since a CIG $G$ is entirely determined by the neighbourhoods $\mathcal{N}(i)$ of all nodes $i \in \mathcal{V}$, we obtain the following result on GMS as a direct consequence of Theorem 1.

**Corollary 2.** Consider a process (2) with underlying CIG $G$ and satisfying all the assumptions in Theorem 1. Then, for any sample size
\[N \geq 864(\beta/\rho_{\min}^2) \log(6ps^2/\eta),\]
there is a GMS method delivering a CIG estimate $\hat{G}$ with $\mathbf{P}\{\hat{G} \neq G\} \leq \eta$.

**Proof.** Using (22), we compute an estimate $\hat{\mathcal{N}}(i)$ for each node $i \in \mathcal{V}$. Then, we construct a CIG estimate $\hat{G}$ having an edge $\{i, j\}$ between nodes $i, j \in \mathcal{V}$ when $j \in \hat{\mathcal{N}}(i)$ and $i \in \hat{\mathcal{N}}(j)$. The estimate $\hat{G}$ is correct, i.e., $\hat{G} = G$ whenever all of the estimates $\hat{\mathcal{N}}(i)$ are correct, i.e., $\hat{\mathcal{N}}(i) = \mathcal{N}(i)$ for each node $i \in \mathcal{V}$. The result then follows by combining Theorem 1 with a union bound (over all nodes $i \in \mathcal{V}$).  

\footnotetext[3]{In contrast to our approach [22], the analysis of GMS presented in [39], for the special case of i.i.d. samples, requires all neighbourhoods $\mathcal{N}(i)$ to have exactly the same size $s$, i.e., $\mathcal{N}(i) = s$ for all $i \in \mathcal{V}$.}
The bound (26) indicates that accurate GMS (with prescribed small error rate \( \eta \)) from time series data conforming to the model (2) is possible for a sample size \( N \propto (1/\rho^2_{\text{min}}) \log(ps) \). We note that the bound (26) improves existing bounds on the sample size required for particular GMS methods based on convex optimization [28]. In particular, while the results in [28] indicate that the sample size \( N \) required for GMS scales with \( \log p \), they do not provide the explicit dependence of \( N \) on the guaranteed error rate \( \mathbb{P}\{\hat{G} \neq G\} \), sparsity \( s \) (see (10)) and minimum connection strength \( \rho^2_{\text{min}} \) (see (9)).

It turns out that the bound (26) is sharp since it matches a fundamental lower bound on the required sample size for any GMS method which performs uniformly well for any process of the form (2) and satisfying Assumption [1], [2] and [3]. This lower bound follows directly from the results in [16].

Lemma 3. [16] Theorem 3.1] Consider a GMS method which reads in vector samples \( x[1], \ldots, x[N] \) (see (2)) and delivers an estimate \( \hat{G} \) for the CIG \( G \) between the components \( x_i \), for \( i = 1, \ldots, p \) (see (3)). If the method achieves an error probability \( \mathbb{P}\{E_i\} \) uniformly bounded by some prescribed error level \( \eta \) for any process of the form (2) satisfying Assumption [1], [2] and [3] with \( \rho^2_{\text{min}} \leq 1/4 \), then the sample size must necessarily satisfy \( N > \frac{1}{4\rho^2_{\text{min}}} \).

Combining Theorem 1 with Lemma 3, we conclude that the bound (26) characterizes, up to a constant factor, the minimum required sample size for accurate GMS based on processes of the form (2).

It is instructive to compare the sufficient condition (26) on the sample size \( N \) with the results obtained in [20], [21], [39] for the special case of i.i.d. samples, which coincides with the model (2) for \( B = 1 \) and \( N = L \). We note that for this special case, the bound (26) matches the necessary condition on sample size derived in [20], which confirms the sparse regression method (22) to be optimal in terms of sample size requirement. However, this is already certified by Lemma 3 which is extends the results of [20] to non-stationary processes (2).

At first sight it appears that the bound (26) suggests a smaller required sample size compared to the bound \( N \propto s^2 \log p \) obtained in [21] Corollary 1. However, it is important to note that the lower bound \( \rho^2_{\text{min}} \) (see (9)) on the minimum connection strength \( \rho^2_{i,j} \) (between connected components) cannot be chosen arbitrarily in order to have at least one process (2) satisfying Assumption [1]. In particular, the off-diagonal entries \( K_{i,j}[n] \) of the precision matrices cannot take on arbitrary (large) values, since the precision matrix \( K[n] = (C[n])^{-1} \) (see (1)) must be positive definite.

A practically relevant regime for the minimum connection strength is \( \rho^2_{\text{min}} \leq c/s^2 \) with some constant \( c \) which may depend on \( \beta \) (see (11)). For this regime, which is also considered in [20], the bound (26) becomes \( N \propto s^3 \log p \) which closely resembles the sample size requirement for the convex GMS method in [21].

Finally, we note that Theorem 1 does not involve some incoherence condition, which requires sub-matrices of the covariance matrices \( C^{(b)} \) (see (3)) to be well-conditioned. Such incoherence conditions are typically required by convex relaxations of the sparse regression estimator (22). While convex (Lasso-based) methods are computationally more tractable than non-convex estimators such as (22), convex methods place more stringent conditions (such as some incoherence condition) on the process (2) in order to guarantee accurate estimation of the underlying CIG [42], [43].

4. NUMERICAL RESULTS

We verify the predictions of Theorem 1 by means of numerical experiments involving synthetic data (see Section 4-A) and data collected by pedestrian count devices located in the city of Turku in Finland (see Section 4-B). We also compare our results with the empirical performance obtained from computationally efficient convex optimization methods (see Section 4-A). In order to support reproducible research, we have made the source code for our experiments available under https://github.com/alexjungalo/ResearchPublic/tree/master/GMSNonStat.

4-A. Chain

Our first experiment revolves around a synthetic dataset \( x[n] \) which is generated according to the process model (2) such that the true underlying CIG \( G \) is a chain graph as depicted in Figure 2.

![Fig. 2. The CIG of a process (2) with a chain structure.](image)

In particular, we generated Gaussian random vectors conforming to the process model (2) with \( B = 4 \) blocks. The \( b \)-th block consists of \( L \) i.i.d. random vectors \( x[n] \sim \mathcal{N}(0, C^{(b)}) \) with \( C^{(b)} \) being chosen such that the marginal CIG \( G^{(b)} \) is a chain (see Figure 2) with the edge \( \{b, b+1\} \) missing (see Figure 3).

In order to estimate the neighbourhood \( \mathcal{N}(i) \) of a given node \( i \in V \) in the CIG from the generated vector samples, we use the sparse regression estimator (22) with \( s = 2 \). For such a small sparsity, it is still feasible to compute the estimator (22) by exhaustive search over all subsets of size at most \( s = 2 \). However, for larger sparsity \( s \), (22) becomes intractable and one has to use computationally cheaper methods such as convex optimization methods [7], [27].

We estimate the error probability (23) by an empirical average \( \hat{P}\{E_i\} \) over \( K = 100 \) i.i.d. simulation runs. In
particular, using the $j$-th realization of the process (2) as input to the sparse regression estimator (22), yielding the estimate $\hat{\mathcal{N}}^{(j)}(i)$, we compute the empirical error rate

$$\tilde{P}\{\mathcal{E}_i\} := (1/K) \sum_{j=1}^{K} \mathbb{I}(\hat{\mathcal{N}}^{(j)}(i) \neq \mathcal{N}(i)) (\approx P\{\mathcal{E}_i\}).$$

(27)

Here, $\hat{\mathcal{N}}^{(j)}(i)$ is the estimated neighbourhood of node $i \in \mathcal{V}$ during the $j$-th simulation run.

In Figure 4-(a) we depict the error rate $\tilde{P}\{\mathcal{E}_i\}$, achieved by the estimator (2) when estimating the neighbourhood or node $i = 2$ (see Figure 2), as a function of the sample size $N$. The three curves in Figure 4-(a) corresponds to three different processes. Each process is of the form (2) with CIG being a chain (see Figure 2), but with different $p$ and $\rho_{\text{min}}^2$ (see (9)).

As indicated by the upper bound (25) of Theorem 1, the error rate $P\{\mathcal{E}_i\}$ (see (27)) crucially depends on the scaled sample size $N' := N\rho_{\text{min}}^2/\log p$. Therefore, we plot in Figure 4-(b) the error rate $\tilde{P}\{\mathcal{E}_i\}$ as a function of the scaled sample size $N'$. In agreement with our theoretical findings, we observe that the curves in Figure 4-(b) are almost lying on top of each other.

The sparse regression estimator (22) implements a form of pooling of the samples $x[1], \ldots, x[N]$ across different blocks. Indeed, the objective function in (22) sums up the contributions from all blocks such that the required sample size depends on the average connection strength $\mathcal{G}$. A simple alternative approach would be to consider the samples of each block in (2) as i.i.d. samples from a marginal CIG $\mathcal{G}^{(b)}$ and apply existing GMS methods for i.i.d. samples to obtain estimates for the marginal CIGs. We can then obtain an estimate for the global CIG $\mathcal{G}$ by using the union of the edge sets in each marginal CIG $\mathcal{G}^{(b)}$. More precisely, a “naïve” estimate $\hat{\mathcal{N}}^{(\text{naïve})}(i)$ for the neighborhood $\mathcal{N}(i)$ of some node $i \in \mathcal{V}$ can be obtained from the union of the block-wise neighborhood estimates $\hat{\mathcal{N}}^{(b)}(i)$, for $b = 1, \ldots, B$.

In Figure 5 we compare the error rate achieved by our pooled approach to this naïve approach. In particular, we use a constrained $\ell_1$ minimization approach (referred to as “CLIME”) to estimate the support of the sparse precision matrix (44) within each block. From Figure 5 we obtain that the pooled approach (22) clearly outperforms the naïve approach. This result should not come as a surprise since the pooled estimator (22) allows to cope with few blocks with very small connection strength (of connected nodes in the CIG) as long as the average connection strength (see (8)) is large enough. In contrast, the naïve approach is likely to fail if there is at least one block of samples which does not allow accurate GMS.

As pointed out in Section 3, the sparse regression method (22) becomes intractable except for very small number $p$ of process components in (2) and sparsity level $s$ of the underlying CIG (see Assumption 2). A computationally
more tractable GMS method can be obtained by replacing (relaxing) the non-convex penalty term $\lambda |\mathcal{T}|$ in (22) by a convex approximation. The group Lasso is obtained by a particular choice for this convex approximation as (45)

$$\hat{w} = \arg \min_{\hat{w}^{(b)} \in \mathbb{R}^p} \sum_{b=1}^{B} \| \hat{x}_i^{(b)} - \sum_{j=1}^{p} \hat{w}^{(b)}_j \hat{x}_j^{(b)} \|^2_2 + \frac{\rho}{2} \sum_{j=1}^{p} \| \hat{w}_j \|^2_2. \quad (28)$$

To obtain an estimate for the neighbourhood $\mathcal{N}(i)$ from the estimator (28), we threshold the squared block norms $\| \hat{w}_j \|^2_2 = \sum_{b=1}^{B} (\hat{w}^{(b)}_j)^2$ at the level $\eta = \rho_{\text{min}}/2$.

$$\hat{N}^{(g\text{Lasso})}(i) := \{ j \in \{1, \ldots, p\} \setminus \{i\} : \| \hat{w}_j \|^2_2 \geq \eta \}. \quad (29)$$

In Figure 6, we depict the error rate incurred by (29) for a process with chain-structured CIG (see Figure 2). While Figure 6(a) shows the error rate as a function of the original sample size $N$, Figure 6(b) displays the error rate as a function of the scaled sample size $N' = N\rho_{\text{min}}^2/\log p$. In agreement with our theoretical findings (see Theorem 1), the error rate of the estimator (29) seems to be mainly determined by the scaled sample size $N'$ as indicated in Figure 6(b).

A comparison of Figure 6(b) with Figure 4(b) reveals that the estimator (29) requires more process samples than the estimator (22) to ensure a prescribed error rate. However, the estimator (29) can be implemented by applying computationally efficient convex optimization methods to solve the group Lasso (28) (see [7]).

Fig. 5. Error rate (27) achieved by estimating the neighborhood $\mathcal{N}(2)$ using the pooled estimator (22) (solid curves) and by the union of the neighborhoods obtained by applying CLIME [44] to each block in (2) separately and then forming a union (over all blocks $b = 1, \ldots, B$) of all block-wise neighborhood estimates (dotted curves).

Fig. 6. Empirical error rate $P_{\text{err}} = \hat{P}\{\hat{N}^{(g\text{Lasso})}(2) \neq \mathcal{N}(2)\}$ (see (27)) incurred by the neighborhood estimate (29) applied to a process with chain structured CIG of size $p = 64$ (“+”), $p = 128$ (“x”), $p = 256$ (“◦”) and $p = 512$ (“∗”). (a) Error rate as a function of the sample size $N$. (b) Error rate as a function of scaled sample size $N' = N/\log p$. Error rate has been obtained using $K = 100$ simulation runs.

4-B. Pedestrian Counts

In this experiment we applied the sparse regression estimator (22) to hourly pedestrian counts collected in the city of Turku (Finland). The city operates pedestrian counting devices at certain locations in the city center (see Figure 7). The counting devices (based on cameras) measure the number of pedestrians which pass one of two counting lines in a certain direction (see Figure 7).

We have been provided with hourly count data obtained from $p = 5$ different counting devices located in the city center of Turku (see Figure 7) and collected since 23rd of July, 2018. For each counting device, we compute the average count $z^{(i)}[n]$, for $i = 1, \ldots, p$ at time $n$. We depict the average count $z^{(i)}[n]$ in Figure 8 which indicates a seasonal component with period 24. This is not too surprising as we
expect the pedestrian movements for different days to be similar for the same daytime.

In order to remove the seasonal component we difference the time series \( z^{(i)}[n] \) at lag 24 to obtain the time series (see [34, Chapter 1.4])

\[
\ddot{z}^{(i)}[n] := z^{(i)}[n + 24] - z^{(i)}[n] \quad \text{for } i = 1, \ldots, p. \tag{30}
\]

We depict the time series \( \ddot{z}^{(i)}[n] \) in Figure 9 which suggests that is is reasonable to model \( \ddot{z}^{(i)}[n] \) as a stationary time series (or discrete time process).

As discussed in Section 2, we can transform a stationary process into a process conforming to our non-stationary model (2) by applying a DFT. We compute the DFT of the difference time series \( \ddot{z}^{(i)}[n] \) (see [30]) using a period \( N = 3072 \) to obtain the vector-valued samples

\[
x[n] = (x^{(1)}[n], \ldots, x^{(p)}[n])^T,
\]

with

\[
x^{(i)}[n] := \sum_{n' = 1}^{N} \ddot{z}^{(i)}[n'] \exp \left( -2\pi i (n' - 1)(n - 1)/N \right) \tag{31}
\]

for \( n = 1, \ldots, N \) and \( i := \sqrt{-1} \).

Fig. 8. Hourly pedestrian counts (averaged over two counting lines) at location \( j = 1 \) as indicated in Figure 7.

Fig. 9. Differenced (at lag 24) hourly pedestrian counts \( \ddot{z}^{(i)}[n] \) (see [30]) at location \( j = 1 \) (see Figure 7).

We model the samples \( x[n] \) using (2) with a block-length \( L = 12 \) which has been chosen based on the empirical autocorrelation functions of the differenced time series \( \ddot{z}^{(i)} \) (see [39]). In order to infer the neighbourhoods \( N(i) \) in the CIG underlying the count measurements, we compute the test statistic

\[
Z(T) := (1/N) \sum_{b=1}^{B} \| F^{(b)}_T x^{(b)}_1 \|_2^2
\]

\[
= (1/N) \sum_{b=1}^{B} \min_{w^{(b)} \in \mathcal{R}} \| x^{(b)}_1 - \sum_{j \in T} w^{(b)} \|_2^2, \tag{32}
\]

with the DFT samples (31) and varying candidate sets \( T \subseteq \)
Since we neither know the maximum node degree (sparsity) \( s \), nor a lower bound \( \rho_{\text{min}}^2 \) on the average connection strength, we cannot directly implement the sparse regression estimator \( (22) \). Instead, we try to estimate the neighborhood strength, we cannot directly implement the sparse regression estimator \( (22) \). Nevertheless, we believe that the fundamental dependencies (see \( (26) \)) are sufficiently strong for \( \| 1/N \sum_{i=1}^{B} \min_{w_j} \| x_i^{(b)} - ( \sum_{j \in T'} w_j^{(b)} x_j^{(b)} + cf^{(b)}) \|_2^2 \) (33)

Here, the index set \( T' \) is chosen as \( T' = \arg \min_{|T|} Z(T) \).

The idea behind comparing \( \tilde{E}(s) \) with \( \tilde{E}(s) \) is to test if adding another process component to the components in \( T' \) yields a reduction in the statistic \( \tilde{E}(s+1) \) which is at least twice as large as the reduction of \( \tilde{E}(s) \) achieved by adding a “fake” pedestrian count signal obtained by i.i.d. uniformly distributed random variables \( f[n] \sim U[0, U] \). The interval size \( U \) is chosen in order to match the empirical variance of the pedestrian counts \( x^{(i)}[n] \).

We have obtained the following estimates for the neighborhoods in the CIG underlying the pedestrian count data:

\[
\hat{N}(1) = \{2\}, \hat{N}(2) = \{3, 5\}, \hat{N}(3) = \{2, 4\}, \hat{N}(4) = \{3, 5\}, \hat{N}(5) = \{2, 4\}.
\]

While our analysis applies only to real-valued vector samples \( x[n] \) in \( \mathbb{C} \), the vector samples \( \bar{x}[n] \) obtained from a DFT are typically complex-valued. However, we expect our analysis to also apply to complex-valued samples in \( \mathbb{C} \) by applying straightforward modifications of our methods. In particular, we believe that the fundamental dependencies (see \( (26) \)) between required sample size \( N \) on number \( p \) of process components, sparsity \( s \) and average connection strength \( \rho_{\text{min}}^2 \) to remain valid when allowing the samples in \( \mathbb{C} \) to be complex-valued Gaussian vectors.

In Figure [11] we depicted the CIG estimate obtained by placing an edge between nodes \( i, j \in \hat{V} \) if either \( j \in \hat{N}(i) \) or \( i \in \hat{N}(j) \). The estimated graph structure seems well-aligned with the local road network.

Fig. 10. Score \( \tilde{E}(s) = \min_{|T|} Z(T) \) achieved by minimizing the statistic \( (32) \) (for node \( i = 1 \)) over all candidate sets \( T \) with a prescribed size \( s \).

In Figure [10] we depict the score \( \tilde{E}(s) \) obtained for node \( i = 2 \). We then choose the neighborhood size \( s \) as the smallest number such that \( \tilde{E}(s) - \tilde{E}(s+1) < 2(\tilde{E}(s) - \tilde{E}(s)) \) with the “auxiliary score”

\[
\tilde{E}(s) = (1/N \sum_{i=1}^{B} \min_{w_j} \| x_i^{(b)} - ( \sum_{j \in T'} w_j^{(b)} x_j^{(b)} + cf^{(b)}) \|_2^2.
\]

Fig. 11. Map of Turku city including the locations of pedestrian count devices (depicted as red dots). The links between the count devices indicate the presence of an edge in the estimated CIG underlying the count data.

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6. PROOF OF THE MAIN RESULT

We now verify Theorem 1 by analyzing the probability $P\{\mathcal{E}_i\}$ of the error event $\mathcal{E}_i$ (see (25)) when (22) fails to deliver the correct neighbourhood $\mathcal{N}(i)$ of a particular node $i \in \mathcal{V}$ of the CIG $\mathcal{G}$. Let us introduce the shorthands

$$\mathcal{E}_T := \{Z(\mathcal{N}(i)) + \lambda s_i > Z(\mathcal{T}) + \lambda |\mathcal{T}|, \text{ with } Z(\mathcal{T}) := \frac{1}{N} \sum_{b=1}^{B} \| \mathbf{x}_i^{(b)} \|_2^2. \} \quad (34)$$

It will be convenient to denote the set of all subsets of $\{1, \ldots, p\}$ of size at most $s$ but different from the true neighbourhood $\mathcal{N}(i)$ by

$$\Sigma^p_s := \{ \mathcal{T} \subseteq \{1, \ldots, p\} : |\mathcal{T}| \leq s, \mathcal{T} \neq \mathcal{N}(i) \}.$$ 

Moreover, for given $\ell_1, t \leq s$, denote

$$\mathcal{N}(\ell_1, t) := \{ \mathcal{T} \in \Sigma^p_s : |\mathcal{T}| = t, |\mathcal{N}(i) \setminus \mathcal{T}| = \ell_1 \}. \quad (35)$$

Thus, the set $\mathcal{N}(\ell_1, t) \subseteq \Sigma^p_s$ collects all the index sets in $\Sigma^p_s$ with a prescribed size $t = |\mathcal{T}|$ and overlap $\ell_1 = |\mathcal{N}(i) \setminus \mathcal{T}|$ with the true neighbourhood $\mathcal{N}(i)$.

An elementary combinatorial argument (see [39, Sec. IV]) reveals that the number of these index sets is

$$N(\ell_1, t) := |\mathcal{N}(\ell_1, t)| = \binom{s_i}{\ell_1} \binom{p-s_i}{\ell_2}. \quad (36)$$

with

$$\ell_2 := \ell_1 + (t - s_i). \quad (37)$$

Given a particular node $i \in \mathcal{V}$ with neighbourhood $\mathcal{N}(i)$, the quantities $\ell_1$ and $\ell_2$ are fully determined by the index set $\mathcal{T}$. For notational convenience we will not make this dependence on $\mathcal{T}$ explicit, i.e., we write $\ell_1$ and $\ell_2$ instead of $\ell_1(\mathcal{T})$ and $\ell_2(\mathcal{T})$. Note that

$$\ell_2 = |\mathcal{T} \setminus \mathcal{N}(i)| \text{ and } \ell_1 + \ell_2 > 0 \text{ for every index set } \mathcal{T} \in \mathcal{N}(\ell_1, t). \quad (38)$$

Using the index set

$$\mathcal{I} := \{ (\ell_1, t) \in \mathbb{Z}^2_+ : \ell_1 \leq s_i, t \leq s \} \setminus \{ (0, s_i) \} \text{ with cardinality } |\mathcal{I}| \leq s^2, \quad (39)$$

we can write

$$\Sigma^p_s \subseteq \bigcup_{(\ell_1, t) \in \mathcal{I}} \mathcal{N}(\ell_1, t). \quad (40)$$

Since the error event $\mathcal{E}_i$ (see (25)) can only occur if at least one of the events $\mathcal{E}_\mathcal{T}$, for some $\mathcal{T} \in \Sigma^p_s$, occurs,

$$\mathcal{E}_i \subseteq \bigcup_{\mathcal{T} \in \Sigma^p_s} \mathcal{E}_\mathcal{T}, \quad (41)$$

implying, in turn via a union bound,

$$P\{\mathcal{E}_i\} \leq \sum_{\mathcal{T} \in \Sigma^p_s} P\{\mathcal{E}_\mathcal{T}\} \leq \sum_{(\ell_1, t) \in \mathcal{I}} \sum_{\mathcal{T} \in \mathcal{N}(\ell_1, t)} P\{\mathcal{E}_\mathcal{T}\}. \quad (42)$$

We now derive an upper bound $M(\ell_1, t)$ on the individual probabilities $P\{\mathcal{E}_\mathcal{T}\}$ such that

$$P\{\mathcal{E}_\mathcal{T}\} \leq M(\ell_1, t) \text{ for any } \mathcal{T} \in \mathcal{N}(\ell_1, t). \quad (43)$$

As the notation already indicates, the upper bound $M(\ell_1, t)$ depends on the index set $\mathcal{T}$ only via the overlap $\ell_1 = |\mathcal{N}(i) \setminus \mathcal{T}|$ and the size $t = |\mathcal{T}|$.

Combining (40) with (42) implies, via a union bound,

$$\log P\{\mathcal{E}_i\} \leq \log \sum_{(\ell_1, t) \in \mathcal{I}} \sum_{\mathcal{T} \in \mathcal{N}(\ell_1, t)} P\{\mathcal{E}_\mathcal{T}\}, \quad (44)$$

where

$$\mathcal{A} := \{ \mathcal{T} \subseteq \mathcal{V} : |\mathcal{T}| \leq s, \mathcal{T} \neq \mathcal{N}(i) \}.$$
will play a prominent role in controlling the probabilities of the error events we will analyze the probabilities of those events for the choice where in step $x$ a linear combination of the process components into the single noise vector $E$ we can characterize the error event by introducing the projection matrix (SVD) follows, define the two error events By (47), an error $V \in \mathbb{R}^d$ with the singular values can only occur if either $x \in (\mathbb{R}^d \setminus 0) \cap \mathbb{R}^d$. To this end, we make (12) more handy by stacking the (block-wise) noise vectors $\varepsilon_i^{(b)} \in \mathbb{R}^L$ into the single noise vector

$$
E_i = (\varepsilon_i^{(1)^T}, \ldots, \varepsilon_i^{(B)^T})^T \sim \mathcal{N}(0, C_{\varepsilon_i}) , \text{ with } C_{\varepsilon_i} = \text{blkdiag}\{((1/K_{i,j}^{(b)})I_{L})\}^{B}_{j=1}.
$$

By introducing the projection matrix

$$P_{\perp} := \text{blkdiag}\{P_{\perp}^{(b)}\}^{B}_{b=1},$$

we can characterize the error event $E_T$ in (34), for any $T \in \mathcal{N}(\ell_1, t)$ (see (35)), as

$$E_T = \{Z(\mathcal{N}(i)) - (1/N)||P_{\perp}\varepsilon_i||_2^2 > Z(T) - (1/N)||P_{\perp}\varepsilon_i||_2^2 + \lambda(t - s_i)\}.$$

In order to derive the upper bound $M(\ell_1, t)$ let us, for some number $\delta > 0$ whose precise value to be chosen in what follows, define the two error events

$$E_1(\delta) := \{Z(\mathcal{N}(i)) - (1/N)||P_{\perp}\varepsilon_i||_2^2 \geq \delta + (\lambda/2)(t - s_i)\},$$

$$E_2(\delta) := \{Z(T) - (1/N)||P_{\perp}\varepsilon_i||_2^2 + (\lambda/2)(t - s_i) \leq 2\delta\}. \quad (48b)$$

By (47), an error $E_T$ can only occur if either $E_1(\delta)$ or $E_2(\delta)$ occurs, i.e., $E_T \subseteq E_1(\delta) \cup E_2(\delta)$. Therefore, by a union bound,

$$P\{E_T\} \leq P\{E_1(\delta)\} + P\{E_2(\delta)\} \quad (49)$$

where we condition on the components $x_T = \{x_i\}_{i \in T}$ (cf. (6)).

We will now bound each of the two summands in (49) separately. To this end, we will use the singular value decomposition (SVD)

$$P_{\perp} C_{2_i}^{1/2} = \text{Udiag}\{d_j\}_{j=1}^N V^{T} \quad (50)$$

with the singular values $d_j \in \mathbb{R}_+$ and the singular vectors in the columns of the orthonormal matrices $U \in \mathbb{R}^{N \times N}$ and $V \in \mathbb{R}^{N \times N}$ (i.e., $UU^{T} = VV^{T} = I$). The singular values $d_j$, which satisfy

$$d_j \geq \sqrt{\beta} \quad (51)$$

will play a prominent role in controlling the probabilities of the error events $E_1(\delta)$ and $E_2(\delta)$ (see (48a), (48b)). In particular, we will analyze the probabilities of those events for the choice $\delta = m_3/4$ with

$$m_3 := E\{(1/N)||P_{\perp}\xbar{X}_i||_2^2 | x_T\} \quad (a) \quad (52)$$

where in step (a) we used the statistical independence of $x_i$ and $x_T$ (cf. (13)).

The quantity $m_3$ measures the minimum achievable error when approximating the process component $x_i$ (see (6)) using a linear combination of the process components $x_T = \{x_j\}_{j \in T}$. A lower bound on $m_3$ can be obtained via the minimum
average connection strength $\rho_{\min}^2$ (see Assumption [1]). Indeed,

$$m_3 = \frac{1}{N} \text{tr}\{C_{S_i}^{1/2}P_{T\perp}C_{S_i}^{1/2}\}$$

$$= \frac{1}{N} \text{tr}\{C_{S_i}P_{T\perp}\}$$

$$= \frac{1}{N} \sum_{b=1}^B \text{tr}\{P_{T\perp}^{(b)}\tilde{\sigma}_b^2I\}$$

$$= \frac{1}{N} \sum_{b=1}^B \tilde{\sigma}_b^2(L-|T|).$$

This can be further developed by using the lower bound (16) for the variance $\tilde{\sigma}_b^2$,

$$m_3 \geq \sum_{j \in N(i)} \sum_{b=1}^B (K_{i,j}^{(b)}/K_{i,i}^{(b)})^2(L-|T|)/N$$

$$\geq \ell_1B\rho_{\min}^2(L-|T|)/N$$

$$\geq (2/3)\ell_1\rho_{\min}^2.$$

For the choice $\delta = m_3/4$ this implies, in turn,

$$\delta \geq (1/6)\ell_1\rho_{\min}^2.$$

In order to upper bound the probability of the event $E_1(\delta)$, observe

$$Z(N(i)) = \frac{1}{N} \sum_{b=1}^B \|P_{N(i)\perp}x_i^{(b)}\|^2_2$$

$$= \frac{1}{N} \sum_{b=1}^B \left\| P_{N(i)\perp} \left( \sum_{j \in N(i)} a_j x_j^{(b)} + \epsilon_i^{(b)} \right) \right\|^2_2$$

$$= \frac{1}{N} \| P_{N(i)\perp} \epsilon_i \|^2_2. $$

Hence,

$$P\{E_1(\delta) \mid x_T\} \leq P\left\{ \frac{1}{N} \| P_{T\perp} \epsilon_i \|^2_2 \geq \delta + \frac{\lambda}{2}(s_i-t_i) \mid x_T \right\}$$

$$= P\left\{ (1/N)||P_{N(i)\perp}\epsilon_i||^2_2 - (1/N)||P_{T\perp}\epsilon_i||^2_2 \geq \delta + \frac{\lambda}{2}(s_i-t_i) \mid x_T \right\}$$

$$\leq P\left\{ (1/N)||P_{N(i)\perp} \epsilon_i ||^2_2 - (1/N)||P_{T\perp} \epsilon_i ||^2_2 \geq \frac{\rho_{\min}^2}{12}(s_i-t_i) \mid x_T \right\}$$

$$\leq P\left\{ (1/N)||P_{N(i)\perp} \epsilon_i ||^2_2 - (1/N)||P_{T\perp} \epsilon_i ||^2_2 \geq \frac{\rho_{\min}^2}{12}(s_i-t_i) \mid x_T \right\}.$$
with
\[
(P_T - P_{T \cap N^\perp(i)}) = \text{blkdiag} \{ P_T^{(b)} - P_{T \cap N^\perp(i)}^{(b)} \}_{b=1}^B.
\]
The matrix \(P_T^{(b)} - P_{T \cap N^\perp(i)}^{(b)} \in \mathbb{R}^{L \times L}\) is a random (since it depends on \(x_T = \{x_j\}_{j \in T}\)) orthogonal projection matrix on a subspace of dimension at most \(\ell_2 = |T \setminus N(i)|\) (cf. (38)), i.e.,
\[
P_T^{(b)} - P_{T \cap N^\perp(i)}^{(b)} = \sum_{j=1}^{\ell_2} \tilde{a}_j^{(b)} u_j^{(b)} (u_j^{(b)})^T \tag{60}
\]
with some coefficients \(\tilde{a}_j^{(b)} \in \{0, 1\}\) and orthonormal vectors \(\{u_j^{(b)} \in \mathbb{R}^L\}_{j=1, \ldots, \ell_2}\). Inserting (60) into (59),
\[
P\{\mathcal{E}_1(\delta) \mid x_T\} \leq P\{(1/N) \| (P_T - P_{T \cap N^\perp(i)}) \epsilon_i \|_2^2 \geq \rho_{\min}^2(\ell_1 + \ell_2)/12 \mid x_T\}
\]
(61)

Then, as can be verified easily,
\[
(1/N) \sum_{b=1}^B \sum_{j=1}^{\ell_2} \tilde{a}_j^{(b)} (z_j^{(b)})^2 = \sum_{n=1}^N \hat{a}_n z_n^2, \text{ with } z_n \sim \mathcal{N}(0, 1) \tag{62}
\]
and coefficients \(\hat{a}_n \in [0, \beta/N]\) (cf. (11)) satisfying
\[
\sum_{n=1}^N \hat{a}_n \leq \ell_2 \beta B/N \leq \ell_2 \rho_{\min}^2/24. \tag{63}
\]

Here, step (a) can be verified by taking (conditional, w.r.t. \(x_T = \{x_j\}_{j \in T}\)) expectations of (62) and using \(|a_j^{(b)}| \leq 1, 1/K_j^{(b)} \leq \beta\).

Inserting (62) into (61),
\[
P\{\mathcal{E}_1(\delta) \mid x_T\} \leq P\{\sum_{j=1}^N \hat{a}_j z_j^2 \geq \rho_{\min}^2(\ell_1 + \ell_2)/12 \mid x_T\}
\]
(64)

We now apply Lemma 4 to (64) using the choice
\[
\eta := \rho_{\min}^2(\ell_1 + \ell_2)/24, \tag{65}
\]
\(a_j := \tilde{a}_j\) and \(b_j := 0\) (cf. (83)). This yields
\[
P\{\mathcal{E}_1(\delta) \mid x_T\} \leq 2 \exp\left( - \frac{\eta^2/8}{\sum_{j=1}^N \hat{a}_j^2 + \eta \max_{j=1, \ldots, N} \hat{a}_j} \right).
\]
where the second inequality uses \( \max_{j=1,\ldots,N} \tilde{a}_j \leq \beta/N \). Combining
\[
\sum_{j=1}^{N} \tilde{a}_j \leq \ell_2 \rho_{\text{min}}/24 \leq \eta
\]
with (66), we arrive at
\[
P\{ \mathcal{E}_1(\delta) \} = E\{ P\{ \mathcal{E}_1(\delta) \mid \mathbf{x}_T \} \}
\leq 2 \exp\left( -N \eta/(16\beta) \right)
\leq 2 \exp\left( -N \rho_{\text{min}}(\ell_1 + \ell_2)/(24 \cdot 16\beta) \right).
\]

To upper bound the probability of \( \mathcal{E}_2(\delta) \) (cf. (48b)), consider
\[
P\{ \mathcal{E}_2(\delta) \mid \mathbf{x}_T \} := P\left\{ \frac{1}{N} \sum_{j=1}^{N} v_j^2 d_j^2 + \frac{2}{N} \sum_{j=1}^{N} v_j d_j \varepsilon_j \leq 2\delta - (\lambda/2)(t-s_i) \mid \mathbf{x}_T \right\}
\leq P\left\{ \frac{1}{N} \sum_{j=1}^{N} v_j^2 d_j^2 + \frac{2}{N} \sum_{j=1}^{N} v_j d_j \varepsilon_j \leq -m_3/2 - (\lambda/2)(t-s_i) \mid \mathbf{x}_T \right\}.
\]

By defining the random vector
\[
\mathbf{v} = (v_1, \ldots, v_N)^T := \mathbf{V}^T \mathbf{C}_{\tilde{\mathbf{x}}}^{-1/2} \tilde{\mathbf{x}}_i,
\]
using the (random) orthonormal matrix \( \mathbf{V} \in \mathbb{R}^{N \times N} \) constituted by the singular vectors of the matrix \( \mathbf{P}_{\mathbf{T}} \mathbf{C}^{1/2}_{\tilde{\mathbf{x}}} \) (cf. (50)), we can rewrite (69) as
\[
P\{ \mathcal{E}_2(\delta) \mid \mathbf{x}_T \} \overset{\delta=m_3/4}{=} P\left\{ \frac{1}{N} \sum_{j=1}^{N} v_j^2 d_j^2 + \frac{2}{N} \sum_{j=1}^{N} v_j d_j \varepsilon_j \leq -m_3/2 - (\lambda/2)(t-s_i) \mid \mathbf{x}_T \right\}.
\]

Note that, conditioned on \( \mathbf{x}_T \), the vector \( \mathbf{v} \) is standard Gaussian, i.e., \( \mathbf{v} \sim \mathcal{N}(0, \mathbf{I}_N) \). We now consider (70) for the particular choice \( \lambda = \rho_{\text{min}}/6 \) which yields, using (54),
\[
P\{ \mathcal{E}_2(\delta) \mid \mathbf{x}_T \} \overset{\delta=m_3/4}{=} P\left\{ \frac{1}{N} \sum_{j=1}^{N} v_j^2 d_j^2 + \frac{2}{N} \sum_{j=1}^{N} v_j d_j \varepsilon_j \leq -m_3/2 - (\lambda/2)(t-s_i) \mid \mathbf{x}_T \right\}
\leq P\left\{ \frac{1}{N} \sum_{j=1}^{N} v_j^2 d_j^2 + \frac{2}{N} \sum_{j=1}^{N} v_j d_j \varepsilon_j \leq -3/8 m_3 - ((\lambda/2)(t-s_i) + m_3/8) \mid \mathbf{x}_T \right\}
\leq P\left\{ \frac{1}{N} \sum_{j=1}^{N} v_j^2 d_j^2 + \frac{2}{N} \sum_{j=1}^{N} v_j d_j \varepsilon_j \leq -3/8 m_3 - ((\lambda/2)(t-s_i) + \ell_1 \rho_{\text{min}}/12) \mid \mathbf{x}_T \right\}
\overset{\lambda=\rho_{\text{min}}/6}{=} P\left\{ \frac{1}{N} \sum_{j=1}^{N} v_j^2 d_j^2 + \frac{2}{N} \sum_{j=1}^{N} v_j d_j \varepsilon_j \leq -3/8 m_3 - (\rho_{\text{min}}/12)((t-s_i) + \ell_1) \mid \mathbf{x}_T \right\}
\]
\[ P \left\{ \frac{1}{N} \sum_{j=1}^{N} v_j^2 \varepsilon_j^2 + \frac{2}{N} \sum_{j=1}^{N} v_j d_j \varepsilon_j - m_3 \leq - (3/8) m_3 - (\rho_{\min}/12) \ell_2 | x_T \right\}. \tag{71} \]

We will invoke Lemma 4 to obtain an upper bound for \( P \{ E_2(\delta = m_3/4) | x_T \} \). To this end, in order to control the term \( (2/N) \sum_{j=1}^{N} v_j d_j \varepsilon_j \) in (70), we condition on the event

\[ A := \left\{ \frac{1}{N} \sum_{j=1}^{N} d_j^2 \varepsilon_j^2 \leq 2 (\beta/N) \sum_{j=1}^{N} d_j^2 + \beta \ell_2 (\rho_{\min}/12) \right\} \tag{72} \]

with the constant \( \beta \) of Assumption 3. The event \( A \) is, conditioned on \( x_T \), statistically independent of \( \tilde{x}_i \) (cf. (13)) since, loosely speaking, its definition (72) involves only the random variables \( \{ \varepsilon_j \}_{j=1,...,N} \) which are statistically independent of \( \tilde{x}_i \) (cf. (13)) and quantities (e.g., the singular values \( d_j \)) which are constant when conditioning on \( x_T \).

We can upper bound the probability \( P \{ E_2(\delta = m_3/4) | x_T \} \) as

\[
P \{ E_2(\delta) | x_T \} = P \{ E_2(\delta) | A, x_T \} P \{ A | x_T \} + P \{ E_2(\delta) | A^c, x_T \} P \{ A^c | x_T \} \leq 1
\]

\[ \leq P \{ E_2(\delta) | A, x_T \} + P \{ A^c | x_T \}. \tag{73} \]

In order to control the probability \( P \{ A^c | x_T \} \) in (73), we will invoke Lemma 4. To this end, observe

\[
P \{ A^c | x_T \} = P \{ (1/N) \sum_{j=1}^{N} d_j^2 \varepsilon_j^2 \geq 2 (\beta/N) \sum_{j=1}^{N} d_j^2 + \beta \ell_2 (\rho_{\min}/12) | x_T \}
\]

\[ \leq P \{ (1/N) \sum_{j=1}^{N} d_j^2 \varepsilon_j^2 - E \{ (1/N) \sum_{j=1}^{N} d_j^2 \varepsilon_j^2 | x_T \} \geq (\beta/N) \sum_{j=1}^{N} d_j^2 + \beta \ell_2 (\rho_{\min}/12) | x_T \}, \tag{74} \]

where \((a)\) is due to

\[ E \left\{ (1/N) \sum_{j=1}^{N} d_j^2 \varepsilon_j^2 | x_T \right\} = (1/N) \sum_{j=1}^{N} d_j^2 E \{ \varepsilon_j^2 | x_T \} \geq (\beta/N) \sum_{j=1}^{N} d_j^2. \]

The random variables \( \{ \varepsilon_j \}_{j=1,...,N} \) are, conditioned on \( x_T \), i.i.d. zero-mean Gaussian variables with variance \( \sigma_\varepsilon^2 \leq \beta \) (cf. (45)). Therefore, we can use the innovation representation

\[ \varepsilon_j = \tilde{b}_j z_j \tag{75} \]

with i.i.d. standard Gaussian random variables \( z_j \sim N(0,1) \) and some coefficients \( \tilde{b}_j \in [0, \sqrt{\beta}] \). Inserting (75) into (74), we have

\[
P \{ A^c | x_T \} \leq P \left\{ \sum_{j=1}^{N} \tilde{b}_j^2 d_j^2 z_j^2 - E \{ \sum_{j=1}^{N} \tilde{b}_j d_j^2 z_j^2 | x_T \} \geq \beta \left( \sum_{j=1}^{N} d_j^2 + N \ell_2 (\rho_{\min}/12) \right) | x_T \right\}. \tag{76} \]

Applying (84), using the choice \( \eta := \beta \left( \sum_{j=1}^{N} d_j^2 + N \ell_2 (\rho_{\min}/12) \right) \), \( a_j := \tilde{b}_j^2 d_j^2 \) and \( b_j = 0 \) (cf. (83)) to (76), yields

\[
P \{ A^c | x_T \} \leq \exp \left( - \frac{\beta^3 \left( \sum_{j=1}^{N} d_j^2 + N \ell_2 (\rho_{\min}/12) \right)^2}{8 \left( \sum_{j=1}^{N} \tilde{b}_j^4 d_j^4 + \beta \left( \sum_{j=1}^{N} d_j^2 + N \ell_2 (\rho_{\min}/12) \right) \right) \max_{j=1,...,N} \tilde{b}_j^2 d_j^2} \right) \tag{83} \]

\[ \leq \exp \left( - \frac{\left( \sum_{j=1}^{N} d_j^2 + N \ell_2 (\rho_{\min}/12) \right)^2}{16 \beta \left( \sum_{j=1}^{N} d_j^2 + N \ell_2 (\rho_{\min}/12) \right)} \right) \tag{82} \]

\[ \leq \exp \left( - \frac{N \left( m_3 + \ell_2 (\rho_{\min}/12) \right)}{16 \beta} \right) \tag{81} \]
\[ \leq \exp \left( - \frac{N(2/3)\ell_1 \rho_{\min} + \ell_2 (\rho_{\min}/12)}{16\beta} \right) \]
\[ \leq \exp \left( - \frac{N \rho_{\min} (\ell_1 + \ell_2)}{192\beta} \right). \quad (77) \]

In order to control the probability \( P\{E_2(\delta) | A, x_T \} \) appearing in (73), we will again use Lemma 4. To this end, note that
\[ E\left\{ \frac{1}{N} \sum_{j=1}^{N} v_j^2 \delta_j^2 + \frac{2}{N} \sum_{j=1}^{N} v_j d_j \varepsilon_j \mid A, x_T \right\} \overset{(a)}{=} \frac{1}{N} \sum_{j=1}^{N} d_j^2 \delta_j \]
with (a) due to the random variables \( \{v_j\}_{j=1, \ldots, N} \) being i.i.d standard Gaussian \( \mathcal{N}(0, 1) \), conditioned on \( x_T \) and \( A \) (see 72). Then,
\[ P\{E_2(\delta) | A, x_T \} \overset{71, 69}{=} P\left\{ \left| \frac{1}{N} \sum_{j=1}^{N} v_j^2 \delta_j^2 + \frac{2}{N} \sum_{j=1}^{N} v_j d_j \varepsilon_j - m_3 \right| \leq -(3/8) m_3 - (\rho_{\min}/12) \varepsilon_2 \mid A, x_T \right\} \]
\[ \leq P\left\{ \left| \frac{1}{N} \sum_{j=1}^{N} v_j^2 \delta_j^2 + \frac{2}{N} \sum_{j=1}^{N} v_j d_j \varepsilon_j - m_3 \right| \geq (3/8) m_3 + \rho_{\min} \varepsilon_2/12 \mid A, x_T \right\} \]
\[ \leq P\left\{ \left| \frac{1}{N} \sum_{j=1}^{N} v_j^2 \delta_j^2 + \frac{2}{N} \sum_{j=1}^{N} v_j d_j \varepsilon_j - N m_3 \right| \geq N((3/8) m_3 + \rho_{\min} \varepsilon_2/12) \mid A, x_T \right\}. \quad (79) \]

Applying Lemma 4 to (79), using \( \eta:= N(3m_3/8 + \rho_{\min} \varepsilon_2/12), a_j:= d_j^2 \leq \beta, b_j:= d_j \varepsilon_j \) yields
\[ P\{E_2(\delta) | A, x_T \} \overset{51}{\leq} 2 \exp \left( - \frac{N^2(3m_3/8 + \rho_{\min} \varepsilon_2/12)^2}{8} \right) \]
\[ \leq 2 \exp \left( - \frac{N^2(3m_3/8 + \rho_{\min} \varepsilon_2/12)^2}{8 \beta (N m_3 + (1/\beta) \sum_{j=1}^{N} d_j^2 \varepsilon_j^2 + N(3m_3/8 + \rho_{\min} \varepsilon_2/12))} \right) \]
\[ \overset{72}{\leq} 2 \exp \left( - \frac{N^2(3m_3/8 + \rho_{\min} \varepsilon_2/12)^2}{8 \beta (3N m_3 + N \rho_{\min} \varepsilon_2/12 + N(3m_3/8 + \rho_{\min} \varepsilon_2/12))} \right) \]
\[ \leq 2 \exp \left( - \frac{N^2(3m_3/8 + \rho_{\min} \varepsilon_2/12)^2}{8 \beta 9 N (3m_3/8 + \rho_{\min} \varepsilon_2/12)} \right) \]
\[ \overset{54}{\leq} 2 \exp \left( - \frac{N \rho_{\min} (\ell_1 + \ell_2)/12}{72\beta} \right). \quad (80) \]

By combining (80) and (77) with (73),
\[ P\{E_2(\delta)\} = E\{P\{E_2(\delta) \mid x_T\}\} \leq 4 \exp \left( - \frac{N \rho_{\min} (\ell_1 + \ell_2)}{864\beta} \right) \quad (81) \]

Summing (68) and (81) yields (cf. 49)
\[ P\{E_T\} \leq M(\ell_1, t) := 6 \exp \left( - \frac{N \rho_{\min} (\ell_1 + \ell_2)}{864\beta} \right). \quad (82) \]
Inserting the upper bound $82$ into $44$,
\[
\log P\{E_i\} \leq 2 \log s + \max_{(\ell_1, \ell_2) \in \mathcal{I}} \left[ (\ell_1 + \ell_2) \log p + \log 6 - \frac{N \rho_{\text{min}} (\ell_1 + \ell_2)}{864 \beta} \right].
\]
Thus, $P\{E_i\} \leq \eta$ whenever $N \geq 864 \log(p6s^2/\eta)(\beta/\rho_{\text{min}})$.

APPENDIX

The main device underlying our analysis is the following large deviation property of a quadratic form involving Gaussian random variables.

**Lemma 4.** Consider two vectors $a = (a_1, \ldots, a_N)^T \in \mathbb{R}^N$ and $b = (b_1, \ldots, b_N)^T \in \mathbb{R}^N$. For $N$ i.i.d. random variables $z_j \sim \mathcal{N}(0, 1)$, define
\[
y = \sum_{j=1}^{N} a_j z_j^2 + b_j z_j.
\]
Then,
\[
P\{\|y - E\{y\}\| \geq \eta\} \leq 2 \exp \left( -\frac{\eta^2/8}{\|a\|_2^2 + \|b\|_2^2 + \|a\|_\infty \eta} \right).
\]

**Proof.** An elementary calculation (see, e.g., [46, Lemma 7.6]) reveals
\[
E\{\exp(\lambda (a_i z_i^2 + b_i z_i))\} = \exp \left( \frac{\lambda^2 b_i^2/2}{1-2\lambda a_i} \right) \left( \frac{1}{\sqrt{1-2\lambda a_i}} \right),
\]
which holds for any $\lambda \in [0, 1/(4\|a\|_\infty)]$. Hence, for any $i \in \{1, \ldots, N\}$,
\[
\log E\{\exp(\lambda (a_i z_i^2 + b_i z_i - a_i))\} \leq \frac{\lambda^2 b_i^2/2}{1-2\lambda a_i} - (1/2) \log(1-2\lambda a_i) - \lambda a_i.
\]
Since $-\log(1-u) \leq u - \frac{u^2}{2(1-u)}$, for $0 \leq u \leq 1$, the RHS of (86) yields, for every $i \in [N]$,
\[
\log E\{\exp(\lambda (a_i z_i^2 + b_i z_i - a_i))\} \leq \lambda^2 b_i^2 - (1/2) \log(1-2\lambda |a_i|) - \lambda |a_i|.
\]
Summing (87) for $i = 1, \ldots, N$ and inserting into (83),
\[
\log E\{\exp(\lambda (y - E\{y\}))\} \leq 2 \lambda^2 (\|a\|_2^2 + (1/2)\|b\|_2^2).
\]
Now, consider the tail bound (see, e.g., [46, Remark 7.4]),
\[
P\{y - E\{y\} \geq \eta\} \leq \exp(-\lambda \eta) E\{\exp(\lambda (y - E\{y\}))\} \leq \exp(-\lambda \eta + 2 \lambda^2 (\|a\|_2^2 + (1/2)\|b\|_2^2)).
\]
Minimizing the RHS of (89) over $\lambda \in [0, 1/(4\|a\|_\infty)]$,
\[
P\{y - E\{y\} \geq \eta\} \leq \exp \left( -\frac{\eta^2/8}{\|a\|_2^2 + (1/2)\|b\|_2^2 + \|a\|_\infty \eta} \right),
\]
where $(a)$ is due to $x \vee y \leq x + y$ for $x, y \in \mathbb{R}_+$. Similar to (90), one can also verify
\[
P\{y - E\{y\} \leq -\eta\} \leq \exp \left( -\frac{\eta^2/8}{\|a\|_2^2 + (1/2)\|b\|_2^2 + \|a\|_\infty \eta} \right).
\]
Adding (90) and (91) yields (84) by union bound. \qed