

Conditional independence testing with a single realization of a multivariate nonstationary nonlinear time series

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Abstract

Identifying relationships among stochastic processes is a key goal in disciplines that deal with complex temporal systems, such as economics. While the standard toolkit for multivariate time series analysis has many advantages, it can be difficult to capture nonlinear dynamics using linear vector autoregressive models. This difficulty has motivated the development of methods for variable selection, causal discovery, and graphical modeling for nonlinear time series, which routinely employ nonparametric tests for conditional independence. In this paper, we introduce the first framework for conditional independence testing that works with a single realization of a nonstationary nonlinear process. The key technical ingredients are time-varying nonlinear regression, time-varying covariance estimation, and a distribution-uniform strong Gaussian approximation.

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1 Introduction

A great deal of work has been dedicated to developing tests for conditional independence. That is, testing whether two random vectors X and Y are independent given a third random vector Z . For example, there are conditional independence tests based on conditional densities [SW08], characteristic functions [SW07], empirical likelihood ratios [SW14], discretization [Mar05; Hua10], permutation [Dor+14; Sen+17], kernels [Fuk+07; Zha+11; SP11], copulas [BRT12], and conditional mutual information [Run18b]. Also, there are many conditional independence tests based on regressing X on Z and Y on Z followed by testing for independence between the residuals [Pat+09; Pet+14; Ram14; FFX20; ZZG17; Zha+19].

Unfortunately, conditional independence tests oftentimes struggle to control the Type-I error in finite samples, as shown by Shah and Peters [SP20]. In fact, Shah and Peters [SP20] prove that conditional independence testing is fundamentally impossible without making further assumptions. This issue has sparked significant interest in conditional independence testing over the last several years. We begin by providing an overview of recent advances in conditional independence testing. Afterwards, we discuss how our work addresses limitations in the existing literature. Finally, we motivate our work by reviewing key applications of conditional independence tests for time series in areas such as variable selection and causal discovery.

The hardness of conditional independence testing. The no-free-lunch result from Shah and Peters [SP20] states that if one wants to have a conditional independence test with Type-I error control for all absolutely continuous (with respect to the Lebesgue measure) triplets of random vectors (X, Y, Z) , then this conditional independence test cannot have power against any alternative hypothesis. To make the conditional independence testing problem feasible, we must consider a smaller subset

of the null hypothesis and use domain knowledge to select an appropriate conditional independence test. This hardness result was revisited by Neykov et al. [NBW21] and Kim et al. [Kim+22], and was extended to the time series setting by Bodik and Pasche [BP24].

Shah and Peters [SP20] proposed a conditional independence test based on the *generalized covariance measure* (GCM), which is a suitably normalized sum of the products of the residuals from the regressions of X on Z and Y on Z . In this case, the practitioner’s domain knowledge is used to select appropriate regression methods for the problem at hand. In contrast with the previously mentioned tests, Shah and Peters [SP20] show that the GCM test has asymptotic Type-I error control, *uniformly* over a large collection of distributions for which the null hypothesis of conditional independence holds.

Since then, numerous tests have been developed which draw inspiration from the original GCM test [SHB22; LSP22; CPH22; WR23; KKR24; CZK24; Lun+24]. Our conditional independence test can be considered a GCM-type test for the nonstationary nonlinear time series setting. As we will discuss, moving to this complex setting introduces several challenges and requires completely different theoretical tools than the original GCM test.

Limitations of the existing literature. Most of the previously discussed conditional independence tests lack Type-I error control guarantees outside the iid setting. Furthermore, the literature on conditional independence testing when given only a single realization of a nonstationary process remains strikingly limited. To the best of our knowledge, only two tests have been proposed for this setting.

First, Malinsky and Spirtes [MS19] introduce a conditional independence test for nonstationary linear vector autoregressions with iid Gaussian errors. Specifically, they study processes that exhibit “stochastic trends” so that the first difference of the process is stable. In contrast, our conditional independence test allows for nonlinear processes with very general forms of nonstationarity and time-varying regression functions with non-iid and non-Gaussian errors. Moreover, we demonstrate that our conditional independence test possesses uniformly asymptotic Type-I error control, as established for the GCM test from Shah and Peters [SP20].

Second, Flaxman et al. [FNS15] develop a conditional independence testing framework for non-iid data based on Gaussian process regression. The main idea is to pre-whiten the non-iid data using Gaussian process regression to control for dependencies (e.g. spatial, temporal, or network), which should yield iid residuals. The next step is to test for independence between these residuals using the Hilbert-Schmidt Independence Criterion (HSIC) [Gre+07]. The authors state that their framework could be used with nonstationary covariance functions, although this idea was not developed.

We also mention some conditional independence tests designed for the setting in which multiple realizations of a stochastic process are available. Manten et al. [Man+24] develop a conditional independence test for stochastic processes using the signature kernel. Christgau et al. [CPH22] introduced a framework for testing so-called “conditional local independence” relationships among point processes. Lundborg et al. [LSP22] introduce a conditional independence test for function-valued random variables. Also, we note that there is a growing literature on independence testing for nonstationary processes. Liu et al. [Liu+23] develop independence tests based on the HSIC [Gre+07]. These tests require multiple realizations of the nonstationary process, whereas the independence tests for locally stationary processes from [Bee21; Bru22] only require one realization.

Variable selection for forecasting. A central problem in statistics and machine learning is variable selection. Conditional independence tests can be used for variable selection when paired with multiple testing procedures to control the false discovery rate (FDR) [BH95; BY01]. In the context of forecasting, the goal is to identify a minimal subset $S \subseteq \{1, \dots, p\}$ out of p signals (including relevant lags) such that, for all times t , the forecasting target Y_{t+h} at horizon h is conditionally independent of the other signals $(X_t^i)_{i \notin S}$ given $(X_t^i)_{i \in S}$. See Pearl [Pea14] and Candés et al. [Can+18] for more discussion of variable selection. We contribute to this literature by providing a conditional independence test flexible enough to be used for identifying relevant forecasting signals in unstable environments.

Causal discovery for time series. The discovery of time-lagged causal relationships (see Figure 1) from observational time series is an important problem in numerous scientific domains [Run+19a]. Conditional independence tests for time series are a core component of constraint-based and hybrid causal discovery algorithms designed for temporally correlated data. For example, Runge et al. [Run+19b] used the conditional mutual information-based conditional independence test from Runge [Run18b]

in a causal discovery algorithm for time series called PCMCI, which builds on the foundational PC algorithm from Spirtes et al. [SGS01]. We discuss a different framework for assessing relationships between stochastic processes called Granger causality in Section C.11.

Over the last several years, causal discovery for nonstationary time series has become an increasingly active area of research [MS19; Hua+20; FHG23; Don+23; SGF24]. We emphasize that the conditional independence test used in the causal discovery algorithm must be appropriately tailored to the characteristics of the data. For instance, if the underlying conditional independence test fails to account for nonstationarity, then the causal discovery algorithm may produce incorrect conclusions about the causal structure of the process. Our work fills a gap in the literature on causal discovery for nonstationary time series by providing a practical conditional independence test for this setting.

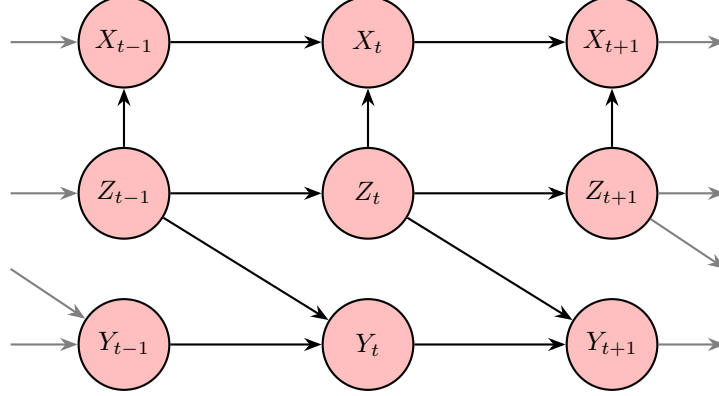


Figure 1: Causal graph depicting the time-lagged causal relationships among the stochastic processes $X = (X_t)_{t \in \mathbb{Z}}$, $Y = (Y_t)_{t \in \mathbb{Z}}$, and $Z = (Z_t)_{t \in \mathbb{Z}}$. The causal graph shows that Z is a common cause of both X and Y , directly influencing X in the same time period and affecting Y with a one time step delay. In this example, the causal graphical structure of the multivariate process (X, Y, Z) remains fixed over time, though the causal effects themselves may vary over time.

1.1 Our contributions

We summarize the key contributions of the paper here.

- We propose the first conditional independence test which can be used with a single realization of a nonstationary nonlinear process. In Theorem 3.1, we show that our test has asymptotic Type-I error control, *uniformly* over a large family of distributions for which the null hypothesis of conditional independence holds. For the multivariate version of our test, we provide ℓ_∞ -type and ℓ_2 -type test statistics so we can achieve high power against sparse or dense alternatives.
- Our test statistics are based on the sample covariance between the residuals from black-box time-varying regressions of X on Z and Y on Z . In contrast with other regression-based conditional independence tests which require iid errors, we allow the errors to be nonstationary nonlinear processes that satisfy a certain martingale difference sequence condition.
- We state a distribution-uniform version of the strong Gaussian approximation for high-dimensional nonstationary nonlinear time series from Mies and Steland [MS23]. We use this result with the products of the aforementioned error processes to justify our bootstrap procedure in Algorithm 1.
- In Theorem 4.1, we provide a guarantee for an instantiation of our test based on the sieve time-varying nonlinear regression estimator from Ding and Zhou [DZ21]. This test complements the work by the same authors on autoregressive approximations and the partial autocorrelation function for locally stationary time series, both of which use the method of sieves [DZ23; DZ25].
- In our simulations, we demonstrate the satisfactory performance of our test when using this sieve estimator. Also, we find that other regression-based conditional independence tests are extremely sensitive to even mild forms of nonstationarity and temporal dependence.

- In Section 5.1, we introduce a novel cross-validation procedure based on subsampling for non-parametric estimators of time-varying regression functions of nonstationary nonlinear processes.
- We show how our framework can be used for (unconditional) independence testing with multivariate nonstationary nonlinear time series in Section C.10. Thus, we supply the two essential components required for constraint-based causal discovery algorithms in this challenging setting.

1.2 Paper outline

The rest of the paper is structured as follows. In Section 2, we discuss the main ideas and implementation of our proposed test. In Section 3, we introduce the theoretical framework and state the theoretical guarantee for our test. In Section 4, we consider an instantiation of our test based on the sieve estimator from Ding and Zhou [DZ21]. In Section 5, we demonstrate the satisfactory performance of our test using this sieve estimator by conducting comprehensive simulations. In Section 6, we conduct a real data application. In Section 7, we summarize our findings and discuss future work.

In Section A, we prove the main results. In Section B, we state a distribution-uniform strong Gaussian approximation for nonstationary nonlinear processes. We provide extensions and additional discussions in Section C.

2 The Dynamic Generalized Covariance Measure (dGCM)

In this section, we give a high-level overview of our work. Specifically, we introduce the notation, main ideas, and implementation of our proposed *dynamic generalized covariance measure* (dGCM) test. For expository purposes, we delay the technical details of our theoretical framework until Section 3.

2.1 Setting and notation

We work in a triangular array framework for high-dimensional nonstationary nonlinear time series. Let $(X_{t,n}, Y_{t,n}, Z_{t,n})_{t \in [n]}$ be the observed sequence of length $n \in \mathbb{N}$, where $[n] = \{1, \dots, n\}$. We use the notation $X_n = (X_{t,n})_{t \in [n]}$, $Y_n = (Y_{t,n})_{t \in [n]}$, $Z_n = (Z_{t,n})_{t \in [n]}$ to refer to each observed sequence of length n , and we use the notation X, Y, Z to refer to each process with any length. Let $d_X = d_{X,n}$, $d_Y = d_{Y,n}$, $d_Z = d_{Z,n}$ denote the dimensions, which can grow with n . Denote dimension $i \in [d_X]$ of $X_{t,n}$ by $X_{t,n,i}$, dimension $j \in [d_Y]$ of $Y_{t,n}$ by $Y_{t,n,j}$, and dimension $k \in [d_Z]$ of $Z_{t,n}$ by $Z_{t,n,k}$.

Next, we introduce notation for the time-offsets of each dimension of $X_{t,n}$, $Y_{t,n}$, $Z_{t,n}$ because we want to infer time-lagged conditional dependencies. Negative time-offsets are called lags of the process, and positive time-offsets are called leads of the process. Time-offsets of zero are allowed so that contemporaneous conditional dependencies can be considered. Let

$$A_i \subset \{-n+1, \dots, n-1\}, B_j \subset \{-n+1, \dots, n-1\}, C_k \subset \{-n+1, \dots, 0\},$$

be the sets of time-offsets of $X_{t,n,i}$, $Y_{t,n,j}$, $Z_{t,n,k}$ under consideration. We require the time-offsets C_k to be non-positive so that the conditioning variables are known at time t . In practice, the largest (in magnitude) time-offsets should be selected small enough so that there is a sufficient amount of data to conduct the test.

Denote the time-offset $a \in A_i$ of $X_{t,n,i}$ by $X_{t,n,i,a} = X_{t+a,n,i}$, the time-offset $b \in B_j$ of $Y_{t,n,j}$ by $Y_{t,n,j,b} = Y_{t+b,n,j}$, and the time-offset $c \in C_k$ of $Z_{t,n,k}$ by $Z_{t,n,k,c} = Z_{t+c,n,k}$. Denote the sets of all time-offsets by $A = \bigcup_{i=1}^{d_X} A_i$, $B = \bigcup_{j=1}^{d_Y} B_j$, $C = \bigcup_{k=1}^{d_Z} C_k$, and largest (signed) time-offsets by $a_{\max} = \max(A)$, $b_{\max} = \max(B)$, $c_{\max} = \max(C)$, and the smallest (signed) time-offsets by $a_{\min} = \min(A)$, $b_{\min} = \min(B)$, $c_{\min} = \min(C)$.

Since we are interested in time-lagged conditional independence relationships, it is often useful to refer to the subset of original times,

$$\mathcal{T}_n = \{1 - \min(a_{\min}, b_{\min}, c_{\min}), n - \max(a_{\max}, b_{\max}, c_{\max})\} \subseteq \{1, \dots, n\},$$

in which all time-offsets of each dimension of $X_{t,n}$, $Y_{t,n}$, $Z_{t,n}$ are actually observed. Going forward, we will write $t \in \mathcal{T}_n$ instead of $t \in [n]$ because we are only using the subset of times in which all time-offsets are observed. Denote the first time of \mathcal{T}_n by $\mathbb{T}_n^- = \min(\mathcal{T}_n)$, the last time of \mathcal{T}_n by

$\mathbb{T}_n^+ = \max(\mathcal{T}_n)$, and the cardinality of \mathcal{T}_n by $T_n = |\mathcal{T}_n|$. Note that if no negative time-offsets (i.e. lags) are used then $\min(a_{\min}, b_{\min}, c_{\min}) = 0$, and if no positive time-offsets (i.e. leads) are used then $\max(a_{\max}, b_{\max}, c_{\max}) = 0$. Hence, if only time-offsets of zero are used, then $\mathcal{T}_n = [n]$.

For all $t \in \mathcal{T}_n$, denote the vectors with all dimensions and time-offsets of interest by

$$\mathbf{X}_{t,n} = (X_{t,n,i,a})_{i \in [d_X], a \in A_i}, \quad \mathbf{Y}_{t,n} = (Y_{t,n,j,b})_{j \in [d_Y], b \in B_j}, \quad \mathbf{Z}_{t,n} = (Z_{t,n,k,c})_{k \in [d_Z], c \in C_k}.$$

Denote the dimensions of $\mathbf{X}_{t,n}$, $\mathbf{Y}_{t,n}$, $\mathbf{Z}_{t,n}$ by $\mathbf{d}_X = \sum_{i=1}^{d_X} |A_i|$, $\mathbf{d}_Y = \sum_{j=1}^{d_Y} |B_j|$, $\mathbf{d}_Z = \sum_{k=1}^{d_Z} |C_k|$, respectively. Also, denote the entire processes by

$$\mathbf{X}_n = (\mathbf{X}_{t,n})_{t \in \mathcal{T}_n}, \quad \mathbf{Y}_n = (\mathbf{Y}_{t,n})_{t \in \mathcal{T}_n}, \quad \mathbf{Z}_n = (\mathbf{Z}_{t,n})_{t \in \mathcal{T}_n}.$$

We allow the number of time-offsets to grow with n , that is, $A_i = A_{i,n}$, $B_j = B_{j,n}$, $C_k = C_{k,n}$ and $A = A_n$, $B = B_n$, $C = C_n$. However, we require that the largest (in magnitude) time-offset grows at a slower rate than n such that as $n \rightarrow \infty$ we have $\min(a_{\min}, b_{\min}, c_{\min})/n \rightarrow 0$ and $\max(a_{\max}, b_{\max}, c_{\max})/n \rightarrow 0$ so that the number of observed times $T_n \rightarrow \infty$ arbitrarily slowly.

Since we allow both the number of time-offsets and the number of dimensions to grow with n , we introduce the index set

$$\mathcal{D}_n \subseteq \{(i, j, a, b) : i \in [d_X], j \in [d_Y], a \in A_i, b \in B_j\},$$

which contains all of the indices for the dimensions and time-offsets of interest. Note that \mathcal{D}_n is specified by the user, and need not contain all possible combinations. Going forward, we will often refer to the dimension/time-offset tuple by $m = (i, j, a, b) \in \mathcal{D}_n$ to lighten the notation. The index set \mathcal{D}_n depends on the sample size n through the dimensions and the time-offsets, so its cardinality $D_n = |\mathcal{D}_n|$ may grow with n . Note that D_n reflects the intrinsic dimensionality of the problem and will appear frequently in the rest of the paper. In the “best case” scenario, we allow $D_n = O(T_n^{\frac{1}{6}})$. See (22) and the rest of Section B.1 for the full details about how quickly D_n can grow.

For each $n \in \mathbb{N}$, let \mathcal{P}_n be a collection of distributions for the processes, which we allow to change with n . For expository purposes, we delay the technical details about \mathcal{P}_n until the end of Section 3.1.

2.2 The null hypothesis of conditional independence

Our univariate test is for the null hypothesis

$$X_{t,n,i,a} \perp\!\!\!\perp Y_{t,n,j,b} \mid \mathbf{Z}_{t,n} \text{ for all } t \in \mathcal{T}_n, \quad (1)$$

for a single dimension/time-offset tuple $(i, j, a, b) \in \mathcal{D}_n$. If domain knowledge suggests that we can restrict the alternative distributions so that the conditional dependencies are time-invariant, then we can use the alternative hypothesis

$$X_{t,n,i,a} \not\perp\!\!\!\perp Y_{t,n,j,b} \mid \mathbf{Z}_{t,n} \text{ for all } t \in \mathcal{T}_n. \quad (2)$$

We begin by focusing on time-invariant conditional independence relationships, and we address the time-varying case afterwards.

Consider the following forecasting example in the univariate setting with $d_X = 1$, $d_Y = 1$, $\mathbf{d}_Z \geq 1$.

Example 2.1 (Univariate test for time series forecasting). *Suppose we are interested in determining whether our existing forecasting signals $\mathbf{Z}_{t,n}$ render the current value of a new signal irrelevant for forecasting a target seven time steps ahead. Note that $\mathbf{Z}_{t,n}$ can consist of current values and lags of each signal. In this case, we would use our univariate test and the null hypothesis*

$$X_{t,n,1,0} \perp\!\!\!\perp Y_{t,n,1,7} \mid \mathbf{Z}_{t,n} \text{ for all } t \in \mathcal{T}_n.$$

We can conduct several of these univariate tests and use multiple testing procedures to control the false discovery rate. In high dimensions, we can group together correlated dimensions and neighboring time-offsets (i.e. similar leads and lags) and proceed as in Meinshausen [Mei08] and the related literature on hierarchical testing. Start by testing for conditional independence at a coarse resolution

$$X_{t,n,i,a} \perp\!\!\!\perp Y_{t,n,j,b} \mid \mathbf{Z}_{t,n} \text{ for all } t \in \mathcal{T}_n, \text{ for all } (i, j, a, b) \in \mathcal{D}_n, \quad (3)$$

using our multivariate test, and continue trying to attribute significance at finer resolutions while using multiple testing procedures to control the false discovery rate.

Next, suppose we have time series data from different countries, companies, or cities, as is common in economics. Write $X_{t,n,i,a}^\ell$, $Y_{t,n,j,b}^\ell$, $\mathbf{Z}_{t,n}^\ell$ to denote $X_{t,n,i,a}$, $Y_{t,n,j,b}$, $\mathbf{Z}_{t,n}$ at index $\ell \in \mathcal{L}_n$, where \mathcal{L}_n is an index set (e.g. different locations). We can often gain power by grouping together the time series in \mathcal{L}_n and using our multivariate test with the null hypothesis

$$X_{t,n,i,a}^\ell \perp\!\!\!\perp Y_{t,n,j,b}^\ell \mid \mathbf{Z}_{t,n}^\ell \text{ for all } t \in \mathcal{T}_n, \text{ for all } \ell \in \mathcal{L}_n, \quad (4)$$

for a single dimension/time-offset tuple $(i, j, a, b) \in \mathcal{D}_n$. We can use the alternative hypothesis

$$X_{t,n,i,a}^\ell \not\perp\!\!\!\perp Y_{t,n,j,b}^\ell \mid \mathbf{Z}_{t,n}^\ell \text{ for all } t \in \mathcal{T}_n, \text{ for some } \ell \in \mathcal{L}_n, \quad (5)$$

since we have restricted the alternative distributions to be such that the conditional dependencies are time-invariant. If we can further restrict the alternative distributions so that the conditional dependencies are time-invariant and index-invariant, then we can use the alternative hypothesis

$$X_{t,n,i,a}^\ell \not\perp\!\!\!\perp Y_{t,n,j,b}^\ell \mid \mathbf{Z}_{t,n}^\ell \text{ for all } t \in \mathcal{T}_n, \text{ for all } \ell \in \mathcal{L}_n. \quad (6)$$

The following example is about forecasting a group of time series with $d_X = 1$, $d_Y = 1$, $d_Z \geq 1$, and $|\mathcal{L}_n| > 1$. In many cases, the multivariate test used in Example 2.2 will have more power than the univariate test used in Example 2.1. Crucially, the processes at different indices (i.e. $\ell_1, \ell_2 \in \mathcal{L}_n$) can be correlated with one another and have different distributions, which is often the case in economics.

Example 2.2 (Multivariate test for forecasting a group of time series). *As in Example 2.1, we are interested in forecasting a target seven time steps ahead. We want to determine whether the current value of a new forecasting signal is relevant or not after accounting for the existing forecasting signals. However, now we have access to the same set of forecasting signals and targets for each index $\ell \in \mathcal{L}_n$. In this case, we would use the multivariate version of our test with the null hypothesis*

$$X_{t,n,1,0}^\ell \perp\!\!\!\perp Y_{t,n,1,7}^\ell \mid \mathbf{Z}_{t,n}^\ell \text{ for all } t \in \mathcal{T}_n, \text{ for all } \ell \in \mathcal{L}_n.$$

Going forward, we suppress the superscript $\ell \in \mathcal{L}_n$ and revert back to the original notation from $X_{t,n,i,a}^\ell$, $Y_{t,n,j,b}^\ell$, $\mathbf{Z}_{t,n}^\ell$ to $X_{t,n,i,a}$, $Y_{t,n,j,b}$, $\mathbf{Z}_{t,n}$. Note that this superscript can always be ignored outside of the “groups of time series” setting (i.e. when $|\mathcal{L}_n| = 1$).

To deal with the problem of time-varying conditional dependencies, we suggest modeling the conditional dependencies as though they are stable during certain time windows. If the breakpoints separating these time windows are known, then we can simply use our conditional independence test on each of these time windows and use multiple testing procedures to control the false discovery rate. However, this becomes more challenging if the breakpoints are unknown. In future work, we will develop a procedure for identifying time windows during which stable conditional dependencies hold while controlling the false discovery rate. That way, we can focus this manuscript on the main testing procedure. In Section C.6, we discuss how to test for time-varying conditional independence relationships at particular points in time by using the framework of locally stationary processes.

2.3 Time-varying regression functions

For a fixed sample size $n \in \mathbb{N}$, distribution $P \in \mathcal{P}_n$, time $t \in \mathcal{T}_n$ and dimension/time-offset tuple $(i, j, a, b) \in \mathcal{D}_n$, we can always decompose

$$X_{t,n,i,a} = f_{P,t,n,i,a}(\mathbf{Z}_{t,n}) + \varepsilon_{P,t,n,i,a}, \quad Y_{t,n,j,b} = g_{P,t,n,j,b}(\mathbf{Z}_{t,n}) + \xi_{P,t,n,j,b}, \quad (7)$$

where $f_{P,t,n,i,a}(\mathbf{z}) = \mathbb{E}_P(X_{t,n,i,a} \mid \mathbf{Z}_{t,n} = \mathbf{z})$ and $g_{P,t,n,j,b}(\mathbf{z}) = \mathbb{E}_P(Y_{t,n,j,b} \mid \mathbf{Z}_{t,n} = \mathbf{z})$ are the time-varying regression functions. The observed processes and error processes can all be nonstationary nonlinear processes. For $m = (i, j, a, b) \in \mathcal{D}_n$, denote the error products at time t by

$$R_{P,t,n,m} = \varepsilon_{P,t,n,i,a} \xi_{P,t,n,j,b}.$$

Next, let $\hat{f}_{t,n,i,a}$ and $\hat{g}_{t,n,j,b}$ be estimates of $f_{P,t,n,i,a}$ and $g_{P,t,n,j,b}$ created by time-varying nonlinear regressions of $(X_{t,n,i,a})_{t \in \mathcal{T}_n}$ on $(\mathbf{Z}_{t,n})_{t \in \mathcal{T}_n}$ and $(Y_{t,n,j,b})_{t \in \mathcal{T}_n}$ on $(\mathbf{Z}_{t,n})_{t \in \mathcal{T}_n}$, respectively. Let

$$\hat{\varepsilon}_{t,n,i,a} = X_{t,n,i,a} - \hat{f}_{t,n,i,a}(\mathbf{Z}_{t,n}), \quad \hat{\xi}_{t,n,j,b} = Y_{t,n,j,b} - \hat{g}_{t,n,j,b}(\mathbf{Z}_{t,n}),$$

be the corresponding residuals, and denote the product of these residuals at time t by

$$\hat{R}_{t,n,m} = \hat{\varepsilon}_{t,n,i,a} \hat{\xi}_{t,n,j,b}, \quad (8)$$

for $m = (i, j, a, b) \in \mathcal{D}_n$. Let $\hat{\mathbf{R}}_{t,n} = (\hat{R}_{t,n,m})_{m \in \mathcal{D}_n}$ be the high-dimensional vector process containing all the residual products for all dimension/time-offset combinations in \mathcal{D}_n .

2.4 Main ideas of our work and the algorithm

To begin, let us briefly summarize the main ideas behind the univariate version of the original generalized covariance measure (GCM) test from Shah and Peters [SP20]. For this paragraph, momentarily redefine X, Y to be two random variables and Z to be a random vector. The GCM test is based on the “weak” conditional independence criterion of Daudin [Dau80], which states that if $X \perp\!\!\!\perp Y \mid Z$, then $\mathbb{E}_P[\phi(X, Z)\varphi(Y, Z)] = 0$ for all functions $\phi \in L^2_{X,Z}$ and $\varphi \in L^2_{Y,Z}$ such that $\mathbb{E}_P[\phi(X, Z) \mid Z] = 0$ and $\mathbb{E}_P[\varphi(Y, Z) \mid Z] = 0$. Thus, under the null hypothesis of conditional independence, the expectation of the products of errors $\mathbb{E}_P(\varepsilon\xi)$ from the regressions $X = \phi(Z) + \varepsilon$ and $Y = \varphi(Z) + \xi$, or equivalently the expected conditional covariance $\mathbb{E}_P[\text{Cov}_P(X, Y \mid Z)]$, is equal to zero. As discussed in Shah and Peters [SP20], this can be seen as a generalization of the fact that the partial correlation coefficient, defined as the correlation between the residuals of linear regressions of X on Z and Y on Z , is equal to zero if and only if $X \perp\!\!\!\perp Y \mid Z$ when (X, Y, Z) are jointly Gaussian. The GCM test is based on the normalized sum of the products of residuals from the regressions of X on Z and Y on Z .

Now, let us translate the “weak” conditional independence criterion of Daudin [Dau80] into our setting using the notation from Section 2.1. If $X_{t,n,i,a} \perp\!\!\!\perp Y_{t,n,j,b} \mid \mathbf{Z}_{t,n}$, then

$$\mathbb{E}_P[\phi(X_{t,n,i,a}, \mathbf{Z}_{t,n})\varphi(Y_{t,n,j,b}, \mathbf{Z}_{t,n})] = 0,$$

for all functions $\phi \in L^2_{X_{t,n,i,a}, \mathbf{Z}_{t,n}}$ and $\varphi \in L^2_{Y_{t,n,j,b}, \mathbf{Z}_{t,n}}$ such that $\mathbb{E}_P[\phi(X_{t,n,i,a}, \mathbf{Z}_{t,n}) \mid \mathbf{Z}_{t,n}] = 0$ and $\mathbb{E}_P[\varphi(Y_{t,n,j,b}, \mathbf{Z}_{t,n}) \mid \mathbf{Z}_{t,n}] = 0$. Hence, under the null hypothesis, the expected conditional covariances,

$$\rho_{P,t,n,m} = \mathbb{E}_P[\text{Cov}_P(X_{t,n,i,a}, Y_{t,n,j,b} \mid \mathbf{Z}_{t,n})],$$

are always equal to zero for the dimension/time-offset combination $m = (i, j, a, b) \in \mathcal{D}_n$. Equivalently, the mean of the error products $\mathbb{E}_P(R_{P,t,n,m})$ from the time-varying nonlinear regressions of $(X_{t,n,i,a})_{t \in \mathcal{T}_n}$ on $(\mathbf{Z}_{t,n})_{t \in \mathcal{T}_n}$ and $(Y_{t,n,j,b})_{t \in \mathcal{T}_n}$ on $(\mathbf{Z}_{t,n})_{t \in \mathcal{T}_n}$ from Section 2.3 will always be zero under the null. This can be seen as a generalization of the partial correlation coefficient being equal to zero under conditional independence in the linear-Gaussian time series context; see the related discussion about Gaussian graphical models for nonstationary time series in Basu and Rao [BR23].

Crucially, the expected conditional covariances $\rho_{P,t,n,m}$ can be zero at all times, even under alternatives in which the corresponding conditional dependencies always hold. Consequently, we can only hope to have power against alternatives in which the time-varying expected conditional covariances $\rho_{P,t,n,m}$ are non-zero for at least some times. Hence, our test statistic (10) is designed to detect non-zero covariances between the errors $\varepsilon_{P,t,n,i,a}$ and $\xi_{P,t,n,j,b}$ at any point in time along the path.

We use a bootstrap-based testing procedure which appeals to the strong Gaussian approximation in Section B. The key ingredient of this bootstrap procedure is the time-varying covariance structure of the approximating nonstationary Gaussian process. Define the rolling window estimate of the time-varying covariance matrices of the vectors of error products by

$$\hat{\Sigma}_{t,n}^R = \frac{1}{L_n} \left(\sum_{s=t-L_n+1}^t \hat{\mathbf{R}}_{s,n} \right)^{\otimes 2}, \quad (9)$$

where $L_n \in \mathbb{N}$ is a lag-window size parameter and the outer product is denoted by $v^{\otimes 2} = vv^T$. In the univariate case with dimensions $d_X = 1, d_Y = 1, d_Z \geq 1$ and time-offsets $A = \{a\}, B = \{b\}$, we use the rolling-window estimate of the time-varying variances of the error products

$$\hat{\sigma}_{t,n}^R = \frac{1}{L_n} \left(\sum_{s=t-L_n+1}^t \hat{R}_{s,n,m} \right)^2,$$

where $m = (1, 1, a, b) \in \mathcal{D}_n$. We postpone the details about these covariances until Section 3.5, and we discuss how to select L_n in Section 5.1.

Next, we introduce our test statistic, which is based on the residual products from the time-varying regressions of $(X_{t,n,i,a})_{t \in \mathcal{T}_n}$ on $(\mathbf{Z}_{t,n})_{t \in \mathcal{T}_n}$ and $(Y_{t,n,j,b})_{t \in \mathcal{T}_n}$ on $(\mathbf{Z}_{t,n})_{t \in \mathcal{T}_n}$. Define the set of times

$$\mathcal{T}_{n,L} = \{L_n + \mathbb{T}_n^- - 1, \dots, \mathbb{T}_n^+ - 1, \mathbb{T}_n^+\},$$

and denote its cardinality by $T_{n,L} = |\mathcal{T}_{n,L}|$. Denote the entire process containing the residual products by $\hat{\mathbf{R}}_n = (\hat{\mathbf{R}}_{t,n})_{t \in \mathcal{T}_{n,L}}$. The test statistic based on the maximum ℓ_p -norm ($p \geq 2$) achieved by the partial sum process is given by

$$S_{n,p}(\hat{\mathbf{R}}_n) = \max_{s \in \mathcal{T}_{n,L}} \left\| \frac{1}{\sqrt{T_{n,L}}} \sum_{t \leq s} \hat{\mathbf{R}}_{t,n} \right\|_p. \quad (10)$$

For example, we can use ℓ_∞ -type or ℓ_2 -type test statistics to achieve high power against sparse or dense alternatives, respectively. In the univariate case with dimensions $d_X = 1$, $d_Y = 1$, $\mathbf{d}_Z \geq 1$ and time-offsets $A = \{a\}$, $B = \{b\}$, the test statistic reduces to the absolute value of the partial sum process of residual products

$$S_n(\hat{R}_{n,m}) = \max_{s \in \mathcal{T}_{n,L}} \left| \frac{1}{\sqrt{T_{n,L}}} \sum_{t \leq s} \hat{R}_{t,n,m} \right|,$$

where $m = (1, 1, a, b) \in \mathcal{D}_n$ and $\hat{R}_{n,m} = (\hat{R}_{t,n,m})_{t \in \mathcal{T}_{n,L}}$. See Sections C.2 and C.6 for discussions of alternative test statistics, namely those based on the full sum and those employing kernel smoothing.

The multivariate dGCM test is given by Algorithm 1. The main steps are time-varying regression, time-varying covariance estimation, and a bootstrap procedure justified by the distribution-uniform strong Gaussian approximation in Section B. The algorithm for the univariate setting is obtained by replacing $S_{n,p}(\cdot)$, $\hat{\Sigma}_{t,n}^{\mathbf{R}}$, $\hat{\mathbf{R}}_{t,n}$, $\hat{\mathbf{R}}_n$, $\check{\mathbf{R}}_{t,n}^{(r)}$, $\check{\mathbf{R}}_n^{(r)}$ with $S_n(\cdot)$, $\hat{\sigma}_{t,n}^R$, $\hat{R}_{t,n}$, \hat{R}_n , $\check{R}_{t,n}^{(r)}$, $\check{R}_n^{(r)}$, respectively.

Algorithm 1 The dynamic generalized covariance measure (dGCM) test

- 1: **Input:** Dimensions and time-offsets of interest \mathcal{D}_n , time points \mathcal{T}_n , data $(X_{t,n}, Y_{t,n}, Z_{t,n})_{t \in \mathcal{T}_n}$, test statistic $S_{n,p}(\cdot)$, α for the significance level, α' for the quantile $\hat{q}_{1-\alpha'}^{\text{boot}}$, number of simulations s
 - 2: **for** each time $t \in \mathcal{T}_n$ and dimension/time-offset tuple $m = (i, j, a, b) \in \mathcal{D}_n$ **do**
 - 3: Obtain estimates $\hat{f}_{t,n,i,a}$ and $\hat{g}_{t,n,j,b}$ of the time-varying regression functions from (7)
 - 4: Calculate the product of residuals $\hat{R}_{t,n,m} = \hat{\varepsilon}_{t,n,i,a} \hat{\xi}_{t,n,j,b}$ from (8)
 - 5: **end for**
 - 6: Select the lag-window size L_n for covariance estimation according to Section 5.1
 - 7: **for** each time $t \in \mathcal{T}_{n,L}$ **do**
 - 8: Calculate the rolling-window estimates $\hat{\Sigma}_{t,n}^{\mathbf{R}}$ of the time-varying covariance matrices from (9)
 - 9: **end for**
 - 10: **for** each simulation $r = 1, \dots, s$ **do**
 - 11: **for** each time $t \in \mathcal{T}_{n,L}$ **do**
 - 12: Simulate independent Gaussian random vectors $\check{\mathbf{R}}_{t,n}^{(r)} \sim \mathcal{N}(0, \hat{\Sigma}_{t,n}^{\mathbf{R}})$
 - 13: **end for**
 - 14: Calculate the test statistic $S_{n,p}(\check{\mathbf{R}}_n^{(r)})$ from (10) using the Gaussian process $\check{\mathbf{R}}_n^{(r)} = (\check{\mathbf{R}}_{t,n}^{(r)})_{t \in \mathcal{T}_{n,L}}$
 - 15: **end for**
 - 16: Calculate the $1 - \alpha'$ empirical quantile $\hat{q}_{1-\alpha'}^{\text{boot}}$ of $(S_{n,p}(\check{\mathbf{R}}_n^{(r)}))_{r=1}^s$
 - 17: Calculate the test statistic $S_{n,p}(\hat{\mathbf{R}}_n)$ from (10) using the residual products $\hat{\mathbf{R}}_n = (\hat{\mathbf{R}}_{t,n})_{t \in \mathcal{T}_{n,L}}$
 - 18: **if** $S_{n,p}(\hat{\mathbf{R}}_n) > \hat{q}_{1-\alpha'}^{\text{boot}}$ **then**
 - 19: Reject the null hypothesis at the level α
 - 20: **end if**
 - 21: **Output:** Decision to either reject or fail to reject the null hypothesis at the level α
-

3 Assumptions and the Theoretical Result for dGCM

In this section, we provide a theoretical guarantee for the dGCM test from Algorithm 1. To do this, we introduce a theoretical framework for high-dimensional nonstationary nonlinear processes. Our framework enables hypothesis testing based on the residuals formed from the predictions of black-box time-varying regression estimators.

We allow the processes to have long-range temporal dependence and very complicated forms of nonstationarity which can be both abrupt and smooth. We control the temporal dependence and nonstationarity of the processes *uniformly* over collections of distributions by employing versions of the functional dependence measure of Wu [Wu05] and the total variation-type nonstationarity condition of Mies and Steland [MS23]. These distribution-uniform assumptions are needed for the uniform level guarantee for the dGCM test in Theorem 3.1, which is our main theoretical result.

The framework we introduce in this section nests several well-studied classes of processes. In Section 4, we show how our framework nests a class of nonstationary processes called locally stationary processes, which allows for smooth changes over time. We refer interested readers to Dahlhaus [Dah12] and Dahlhaus et al. [DRW19] for more information about the linear and nonlinear cases, respectively. Stationary processes arise as a special case in our framework, precisely when there is no nonstationarity. Similarly, our framework accommodates temporally independent sequences with time-varying distributions. The fundamental setting of iid sequences arises as yet another special case when there is neither nonstationarity nor temporal dependence.

In Section C, we discuss how our framework is compatible with even more types of nonstationary processes. Notably, we explain how our framework nests a very general class of nonstationary processes called piecewise locally stationary processes. This class extends the framework for locally stationary processes by permitting both smooth changes and abrupt breakpoints. Also, we consider a class of nonstationary processes called cyclostationary processes which exhibit repetition over time. Additionally, we explain how our framework can leverage black-box simulators for the multivariate process (X, Z) by using simulation-and-regression techniques.

3.1 Nonstationary observed processes

To begin, we introduce the so-called “causal representations” of the processes. Specifically, we view each dimension of the observed sequence $(X_{t,n}, Y_{t,n}, Z_{t,n})_{t \in [n]}$ as the outputs of a time-varying nonlinear function that is given a sequence of iid inputs. This type of representation has a long history, tracing back to at least Rosenblatt [Ros61] and Wiener [Wie66], though its importance for the statistical analysis of time series was first elucidated by Wu [Wu05]. What follows is most similar to the framework for high-dimensional nonstationary nonlinear processes from Mies and Steland [MS23], which in turn builds on the framework from Zhou and Wu [ZW09]. For the following assumption, let

$$\mathcal{H}_t^X = (\eta_t^X, \eta_{t-1}^X, \dots), \quad \mathcal{H}_t^Y = (\eta_t^Y, \eta_{t-1}^Y, \dots), \quad \mathcal{H}_t^Z = (\eta_t^Z, \eta_{t-1}^Z, \dots),$$

where $(\eta_t^X, \eta_t^Y, \eta_t^Z)_{t \in \mathbb{Z}}$ is a sequence of iid random vectors. Denote the dimensions of $\eta_t^X = \eta_{t,n}^X$, $\eta_t^Y = \eta_{t,n}^Y$, $\eta_t^Z = \eta_{t,n}^Z$ respectively by $d_X^n = d_{X,n}^n$, $d_Y^n = d_{Y,n}^n$, $d_Z^n = d_{Z,n}^n$, which can change with n .

Assumption 3.1 (Causal representations of the observed processes). *Assume that for each time $t \in \mathcal{T}_n$ we can represent each dimension of each of the observed processes as the output of an evolving nonlinear system that was given a sequence of iid inputs:*

$$X_{t,n,i} = G_{t,n,i}^X(\mathcal{H}_t^X), \quad Y_{t,n,j} = G_{t,n,j}^Y(\mathcal{H}_t^Y), \quad Z_{t,n,k} = G_{t,n,k}^Z(\mathcal{H}_t^Z).$$

For each $n \in \mathbb{N}$, $(i, j, a, b) \in \mathcal{D}_n$, $t \in \mathcal{T}_n$, we assume that $G_{t,n,i}^X(\cdot)$, $G_{t,n,j}^Y(\cdot)$, $G_{t,n,k}^Z(\cdot)$ are each measurable functions from $(\mathbb{R}^{d_X^n})^\infty$, $(\mathbb{R}^{d_Y^n})^\infty$, $(\mathbb{R}^{d_Z^n})^\infty$ to \mathbb{R} —where we endow $(\mathbb{R}^{d_X^n})^\infty$, $(\mathbb{R}^{d_Y^n})^\infty$, $(\mathbb{R}^{d_Z^n})^\infty$ with the σ -algebra generated by all finite projections—such that $G_{t,n,i}^X(\mathcal{H}_s^X)$, $G_{t,n,j}^Y(\mathcal{H}_s^Y)$, $G_{t,n,k}^Z(\mathcal{H}_s^Z)$ are each well-defined random variables for each $s \in \mathbb{Z}$ and $(G_{t,n,i}^X(\mathcal{H}_s^X))_{s \in \mathbb{Z}}$, $(G_{t,n,j}^Y(\mathcal{H}_s^Y))_{s \in \mathbb{Z}}$, $(G_{t,n,k}^Z(\mathcal{H}_s^Z))_{s \in \mathbb{Z}}$ are each stationary ergodic processes.

To simplify the notation, we have not defined the input sequences for the observed processes separately for each dimension. Without loss of generality, we can define the measurable functions $G_{t,n,i}^X(\cdot)$, $G_{t,n,j}^Y(\cdot)$, $G_{t,n,k}^Z(\cdot)$ and the inputs η_t^X , η_t^Y , η_t^Z so that each dimension of the observed processes can have idiosyncratic inputs.

We will introduce several more causal representations throughout this paper. Let us state some properties that all causal representations will have to avoid repeating the same ideas each time. The causal representations will all be measurable functions on $(\mathbb{R}^{d^n})^\infty$ for some $d^n \in \mathbb{N}$, where we will always endow $(\mathbb{R}^{d^n})^\infty$ with the σ -algebra generated by all finite projections. The causal mechanism at a particular time $t \in \mathcal{T}_n$ with the input sequence up to a particular $s \in \mathbb{Z}$ is a well-defined random variable or vector. Similarly, the process induced by considering the input sequence up to each $s \in \mathbb{Z}$ with a fixed causal mechanism is a stationary ergodic process, as in Assumption 3.1.

In view of Assumption 3.1, we have the following causal representations for the observed processes with all dimensions

$$\begin{aligned} X_{t,n} &= G_{t,n}^X(\mathcal{H}_t^X) = (G_{t,n,i}^X(\mathcal{H}_t^X))_{i \in [d_X]}, \\ Y_{t,n} &= G_{t,n}^Y(\mathcal{H}_t^Y) = (G_{t,n,j}^Y(\mathcal{H}_t^Y))_{j \in [d_Y]}, \\ Z_{t,n} &= G_{t,n}^Z(\mathcal{H}_t^Z) = (G_{t,n,k}^Z(\mathcal{H}_t^Z))_{k \in [d_Z]}. \end{aligned}$$

Also, we have causal representations for each of the dimensions $i \in [d_X]$, $j \in [d_Y]$, $k \in [d_Z]$ with time-offsets $a \in A_i$, $b \in B_j$, $c \in C_k$

$$\begin{aligned} X_{t,n,i,a} &= G_{t,n,i,a}^X(\mathcal{H}_{t,a}^X) = G_{t+a,n,i}^X(\mathcal{H}_{t+a}^X), \\ Y_{t,n,j,b} &= G_{t,n,j,b}^Y(\mathcal{H}_{t,b}^Y) = G_{t+b,n,j}^Y(\mathcal{H}_{t+b}^Y), \\ Z_{t,n,k,c} &= G_{t,n,k,c}^Z(\mathcal{H}_{t,c}^Z) = G_{t+c,n,k}^Z(\mathcal{H}_{t+c}^Z), \end{aligned}$$

where $\mathcal{H}_{t,a}^X = (\eta_{t+a}^X, \eta_{t-1+a}^X, \dots)$, $\mathcal{H}_{t,b}^Y = (\eta_{t+b}^Y, \eta_{t-1+b}^Y, \dots)$, and $\mathcal{H}_{t,c}^Z = (\eta_{t+c}^Z, \eta_{t-1+c}^Z, \dots)$. We can then write the causal representation of the vectors with all dimensions and time-offsets as

$$\begin{aligned} \mathbf{X}_{t,n} &= \mathbf{G}_{t,n}^X(\mathcal{H}_t^X) = (G_{t,n,i,a}^X(\mathcal{H}_{t,a}^X))_{i \in [d_X], a \in A_i}, \\ \mathbf{Y}_{t,n} &= \mathbf{G}_{t,n}^Y(\mathcal{H}_t^Y) = (G_{t,n,j,b}^Y(\mathcal{H}_{t,b}^Y))_{j \in [d_Y], b \in B_j}, \\ \mathbf{Z}_{t,n} &= \mathbf{G}_{t,n}^Z(\mathcal{H}_t^Z) = (G_{t,n,k,c}^Z(\mathcal{H}_{t,c}^Z))_{k \in [d_Z], c \in C_k}, \end{aligned}$$

where $\mathcal{H}_t^X = (\eta_t^X, \eta_{t-1}^X, \dots)$, $\mathcal{H}_t^Y = (\eta_t^Y, \eta_{t-1}^Y, \dots)$, $\mathcal{H}_t^Z = (\eta_t^Z, \eta_{t-1}^Z, \dots)$, and $\eta_t^X = \eta_{t+a_{\max}}^X$, $\eta_t^Y = \eta_{t+b_{\max}}^Y$, $\eta_t^Z = \eta_{t+c_{\max}}^Z$.

Let Ω be a sample space, \mathcal{B} the Borel sigma-algebra, and (Ω, \mathcal{B}) a measurable space. For fixed $n \in \mathbb{N}$, let (Ω, \mathcal{B}) be equipped with a family of probability measures $(\mathbb{P}_P)_{P \in \mathcal{P}_n}$ so that the joint distribution of the nonlinear stochastic systems

$$(G_{t,n}^X(\mathcal{H}_s^X))_{t \in [n], s \in \mathbb{Z}}, (G_{t,n}^Y(\mathcal{H}_s^Y))_{t \in [n], s \in \mathbb{Z}}, (G_{t,n}^Z(\mathcal{H}_s^Z))_{t \in [n], s \in \mathbb{Z}},$$

under \mathbb{P}_P is $P \in \mathcal{P}_n$, where the collection of distributions \mathcal{P}_n can change with n . The family of probability measures $(\mathbb{P}_P)_{P \in \mathcal{P}_n}$ is defined with respect to the same measurable space (Ω, \mathcal{B}) , but need not have the same dominating measure. Denote the family of probability spaces by $(\Omega, \mathcal{B}, \mathbb{P}_P)_{P \in \mathcal{P}_n}$ and a sequence of such families of probability spaces by $((\Omega, \mathcal{B}, \mathbb{P}_P)_{P \in \mathcal{P}_n})_{n \in \mathbb{N}}$.

For a given sample size $n \in \mathbb{N}$ and distribution $P \in \mathcal{P}_n$, let $\mathbb{E}_P(\cdot)$ denote the expectation of a random variable with distribution determined by P . Let $\mathbb{P}_P(E)$ denote the probability of an event $E \in \mathcal{B}$. We use the notation $o_P(\cdot)$ and $O_P(\cdot)$ in the same way that Shah and Peters [SP20] do, so we replicate their notation here. Let $(V_{P,n})_{n \in \mathbb{N}, P \in \mathcal{P}_n}$ be a family of sequences of random variables with distributions determined by $P \in \mathcal{P}_n$ for some collection of distributions \mathcal{P}_n which will be made clear from the context. We write $V_{P,n} = o_P(1)$ to mean that for all $\epsilon > 0$, we have

$$\sup_{P \in \mathcal{P}_n} \mathbb{P}_P(|V_{P,n}| > \epsilon) \rightarrow 0.$$

Also, by $V_{P,n} = O_P(1)$ we mean for all $\epsilon > 0$, there exists a constant $K > 0$ such that

$$\sup_{n \in \mathbb{N}} \sup_{P \in \mathcal{P}_n} \mathbb{P}_P(|V_{P,n}| > K) < \epsilon.$$

Let $(W_{P,n})_{n \in \mathbb{N}, P \in \mathcal{P}_n}$ be another family of sequences of random variables. By $V_{P,n} = o_P(W_{P,n})$ we mean $V_{P,n} = W_{P,n} R_{P,n}$ and $R_{P,n} = o_P(1)$, and by $V_{P,n} = O_P(W_{P,n})$ we mean $V_{P,n} = W_{P,n} R_{P,n}$ and $R_{P,n} = O_P(1)$.

In the rest of this section, we will state distribution-uniform assumptions with respect to a generic sequence of collections of distributions $(\mathcal{P}_n)_{n \in \mathbb{N}}$ for the observed processes. Let $\mathcal{P}_{0,n}^{\text{CI}}$ be a collection of distributions for the observed processes such that the null hypothesis is true, and let $(\mathcal{P}_{0,n}^{\text{CI}})_{n \in \mathbb{N}}$ be a sequence of such collections of distributions. In our main result, which we state as Theorem 3.1, we will assume that these distribution-uniform assumptions hold for a sequence of collections of distributions $(\mathcal{P}_{0,n}^*)_{n \in \mathbb{N}}$, where $\mathcal{P}_{0,n}^* \subset \mathcal{P}_{0,n}^{\text{CI}}$ for each $n \in \mathbb{N}$. That is, we will make these assumptions for a sequence of subcollections of distributions for which the global null hypothesis of conditional independence holds.

3.2 Prediction processes

Next, we introduce causal representations for the prediction processes. For each $n \in \mathbb{N}$, $t \in \mathcal{T}_n$, $(i, j, a, b) \in \mathcal{D}_n$, let $\eta_{t,n,i,a}^{\text{algo}}$, $\eta_{t,n,j,b}^{\text{algo}}$ be random variables that encode the (possible) stochasticity of the statistical learning algorithms. If the learning algorithms are not stochastic, then these random variables can be ignored without loss of generality. Going forward, we will suppress the dependence of the predictors on $\eta_{t,n,i,a}^{\text{algo}}$, $\eta_{t,n,j,b}^{\text{algo}}$ to simplify the notation.

Let $\mathfrak{D}_{t,n,i,a}^{\hat{f}}$, $\mathfrak{D}_{t,n,j,b}^{\hat{g}}$ be the datasets containing the observations used to form the predictors $\hat{f}_{t,n,i,a}$, $\hat{g}_{t,n,j,b}$, and let $\mathcal{H}_{t,a}^{\mathfrak{D}_{t,n,i,a}^{\hat{f}}}$, $\mathcal{H}_{t,b}^{\mathfrak{D}_{t,n,j,b}^{\hat{g}}}$ be the corresponding input sequences. For example, if only the observations in \mathcal{T}_n up to time $t \in \mathcal{T}_n$ are used to form the predictor $\hat{g}_{t,n,j,b}$, then $\mathfrak{D}_{t,n,j,b}^{\hat{g}} = (Y_{s,n,j,b}, \mathbf{Z}_{s,n})_{s \leq t}$ and $\mathcal{H}_{t,b}^{\mathfrak{D}_{t,n,j,b}^{\hat{g}}} = (\mathcal{H}_{t,b}^Y, \mathcal{H}_t^Z)$. Similarly, if all of the observations in \mathcal{T}_n are used (i.e. to time \mathbb{T}_n^+) to form the predictor $\hat{g}_{t,n,j,b}$, then $\mathfrak{D}_{t,n,j,b}^{\hat{g}} = (Y_{t,n,j,b}, \mathbf{Z}_{t,n})_{t \in \mathcal{T}_n}$ and $\mathcal{H}_{t,b}^{\mathfrak{D}_{t,n,j,b}^{\hat{g}}} = (\mathcal{H}_{\mathbb{T}_n^+,b}^Y, \mathcal{H}_{\mathbb{T}_n^+}^Z)$.

Denote the sets of times corresponding to $\mathfrak{D}_{t,n,i,a}^{\hat{f}}$, $\mathfrak{D}_{t,n,j,b}^{\hat{g}}$ by $\mathcal{T}_{t,n,i,a}^{\hat{f}}$, $\mathcal{T}_{t,n,j,b}^{\hat{g}}$, respectively, and let $T_{t,n,i,a}^{\hat{f}} = |\mathcal{T}_{t,n,i,a}^{\hat{f}}|$, $T_{t,n,j,b}^{\hat{g}} = |\mathcal{T}_{t,n,j,b}^{\hat{g}}|$ be the cardinalities. For each $n \in \mathbb{N}$, $t \in \mathcal{T}_n$ let $\mathcal{M}(\mathcal{Z}, \mathcal{Y}) \subseteq \mathcal{Y}^{\mathcal{Z}}$ and $\mathcal{M}(\mathcal{Z}, \mathcal{X}) \subseteq \mathcal{X}^{\mathcal{Z}}$, where $\mathcal{X} = \mathbb{R}$, $\mathcal{Y} = \mathbb{R}$, and $\mathcal{Z} = \mathbb{R}^{\mathbf{d}_Z}$. Note that \mathbf{d}_Z can grow with n as discussed in Section 2.1, although we suppress this in the notation.

Assumption 3.2 (Causal representations of the predictors). *For each $n \in \mathbb{N}$, $(i, j, a, b) \in \mathcal{D}_n$, assume that the sequences of statistical learning algorithms $\mathcal{A}_{n,i,a}^{\hat{f}} = (\mathcal{A}_{t,n,i,a}^{\hat{f}})_{t \in \mathcal{T}_n}$, $\mathcal{A}_{n,j,b}^{\hat{g}} = (\mathcal{A}_{t,n,j,b}^{\hat{g}})_{t \in \mathcal{T}_n}$ consist of the Borel measurable functions*

$$\mathcal{A}_{t,n,i,a}^{\hat{f}} : \begin{cases} (\mathcal{Z} \times \mathcal{X})^{T_{t,n,i,a}^{\hat{f}}} \rightarrow \mathcal{M}(\mathcal{Z}, \mathcal{X}) \\ \mathfrak{D}_{t,n,i,a}^{\hat{f}} \rightarrow \hat{f}_{t,n,i,a}, \end{cases}$$

and

$$\mathcal{A}_{t,n,j,b}^{\hat{g}} : \begin{cases} (\mathcal{Z} \times \mathcal{Y})^{T_{t,n,j,b}^{\hat{g}}} \rightarrow \mathcal{M}(\mathcal{Z}, \mathcal{Y}) \\ \mathfrak{D}_{t,n,j,b}^{\hat{g}} \mapsto \hat{g}_{t,n,j,b}, \end{cases}$$

such that the predictors have the causal representations

$$\begin{aligned} \hat{f}_{t,n,i,a} &= G_{t,n,i,a}^{\mathcal{A}_{t,n,i,a}^{\hat{f}}}(\mathcal{H}_{t,a}^{\mathfrak{D}_{t,n,i,a}^{\hat{f}}}), \\ \hat{g}_{t,n,j,b} &= G_{t,n,j,b}^{\mathcal{A}_{t,n,j,b}^{\hat{g}}}(\mathcal{H}_{t,b}^{\mathfrak{D}_{t,n,j,b}^{\hat{g}}}), \end{aligned}$$

in view of Assumption 3.1. $G_{t,n,i,a}^{\mathcal{A}_{t,n,i,a}^{\hat{f}}}(\cdot)$, $G_{t,n,j,b}^{\mathcal{A}_{t,n,j,b}^{\hat{g}}}(\cdot)$ are measurable functions so that $G_{t,n,i,a}^{\mathcal{A}_{t,n,i,a}^{\hat{f}}}(\mathcal{H}_{t,a}^{\mathfrak{D}_{t,n,i,a}^{\hat{f}}})$, $G_{t,n,j,b}^{\mathcal{A}_{t,n,j,b}^{\hat{g}}}(\mathcal{H}_{t,b}^{\mathfrak{D}_{t,n,j,b}^{\hat{g}}})$ are well-defined function-valued random variables.

We make the following assumption for the predictions and prediction errors for some sequence of collections of distributions $(\mathcal{P}_n)_{n \in \mathbb{N}}$.

Assumption 3.3 (Causal representations of the predictions and prediction errors). *Assume that the predictors $\hat{f}_{t,n,i,a}$, $\hat{g}_{t,n,j,b}$ are Borel measurable functions from $\mathbb{R}^{\mathbf{d}_Z}$ to \mathbb{R} such that for each $n \in \mathbb{N}$, $t \in \mathcal{T}_n$, $(i, j, a, b) \in \mathcal{D}_n$ we can represent the predictions and prediction errors as*

$$\begin{aligned} \hat{f}_{t,n,i,a}(\mathbf{Z}_{t,n}) &= G_{t,n,i,a}^{\hat{f}}(\mathcal{H}_{t,a}^{\hat{f}}) = [\mathcal{A}_{t,n,i,a}^{\hat{f}}(X_{n,i,a}, \mathbf{Z}_n)](\mathbf{Z}_{t,n}), \\ \hat{g}_{t,n,j,b}(\mathbf{Z}_{t,n}) &= G_{t,n,j,b}^{\hat{g}}(\mathcal{H}_{t,b}^{\hat{g}}) = [\mathcal{A}_{t,n,j,b}^{\hat{g}}(Y_{n,j,b}, \mathbf{Z}_n)](\mathbf{Z}_{t,n}), \end{aligned}$$

and

$$\begin{aligned}\hat{w}_{P,t,n,i,a}^f &= G_{P,t,n,i,a}^{\hat{w}^f}(\mathcal{H}_{t,a}^f) = f_{P,t,n,i,a}(\mathbf{Z}_{t,n}) - \hat{f}_{t,n,i,a}(\mathbf{Z}_{t,n}), \\ \hat{w}_{P,t,n,j,b}^g &= G_{P,t,n,j,b}^{\hat{w}^g}(\mathcal{H}_{t,b}^g) = g_{P,t,n,j,b}(\mathbf{Z}_{t,n}) - \hat{g}_{t,n,j,b}(\mathbf{Z}_{t,n}),\end{aligned}$$

in view of Assumptions 3.1, 3.2, where the input sequences are

$$\mathcal{H}_{t,a}^f = (\mathcal{H}_{t,a}^{\mathfrak{D}^f}, \mathcal{H}_t^Z), \quad \mathcal{H}_{t,b}^g = (\mathcal{H}_{t,b}^{\mathfrak{D}^g}, \mathcal{H}_t^Z).$$

Also, assume that for all $n \in \mathbb{N}$, $t \in \mathcal{T}_n$, $(i, j, a, b) \in \mathcal{D}_n$ there exists some $q \geq 2$ such that

$$\sup_{P \in \mathcal{P}_n} \mathbb{E}_P(|\hat{w}_{P,t,n,i,a}^f|^q) < \infty, \quad \sup_{P \in \mathcal{P}_n} \mathbb{E}_P(|\hat{w}_{P,t,n,j,b}^g|^q) < \infty.$$

$G_{t,n,i,a}^f(\cdot)$, $G_{P,t,n,i,a}^{\hat{w}^f}(\cdot)$ and $G_{t,n,j,b}^g(\cdot)$, $G_{P,t,n,j,b}^{\hat{w}^g}(\cdot)$ are measurable functions such that $G_{t,n,i,a}^f(\mathcal{H}_{t,a}^f)$, $G_{t,n,j,b}^g(\mathcal{H}_{t,b}^g)$ and $G_{P,t,n,i,a}^{\hat{w}^f}(\mathcal{H}_{t,a}^f)$, $G_{P,t,n,j,b}^{\hat{w}^g}(\mathcal{H}_{t,b}^g)$ are well-defined real-valued random variables.

In view of Assumption 3.3, we have the following causal representation for all dimensions and time-offsets of the prediction errors

$$\begin{aligned}\hat{w}_{P,t,n}^f &= \mathbf{G}_{P,t,n}^{\hat{w}^f}(\mathcal{H}_t^f) = (\hat{w}_{P,t,n,i,a}^f)_{i \in [d_X], a \in A_i}, \\ \hat{w}_{P,t,n}^g &= \mathbf{G}_{P,t,n}^{\hat{w}^g}(\mathcal{H}_t^g) = (\hat{w}_{P,t,n,j,b}^g)_{j \in [d_Y], b \in B_j},\end{aligned}$$

where $\mathcal{H}_t^f = (\mathcal{H}_{t,a}^f)_{a \in A}$ and $\mathcal{H}_t^g = (\mathcal{H}_{t,b}^g)_{b \in B}$.

3.3 Nonstationary error processes

The most important part of our theoretical framework is the causal representation of the process of error products. For the next assumption, for each $a \in A$, $b \in B$, define the input sequences

$$\mathcal{H}_{t,a}^\varepsilon = (\eta_{t,a}^\varepsilon, \eta_{t,a-1}^\varepsilon, \dots), \quad \mathcal{H}_{t,b}^\xi = (\eta_{t,b}^\xi, \eta_{t,b-1}^\xi, \dots), \quad (11)$$

where $(\eta_{t,a}^\varepsilon, \eta_{t,b}^\xi)_{t \in \mathbb{Z}}$ is a sequence of iid random vectors. For the following assumption, denote the dimension of $\eta_{t,a}^\varepsilon = \eta_{t,a,n}^\varepsilon$ by $d_\varepsilon^\eta = d_{\varepsilon,n}^\eta$, and the dimension of $\eta_{t,b}^\xi = \eta_{t,b,n}^\xi$ by $d_\xi^\eta = d_{\xi,n}^\eta$, both of which can change with n .

Assumption 3.4 (Causal representations of the error processes). *Assume that for each $n \in \mathbb{N}$, $P \in \mathcal{P}_n$, $(i, j, a, b) \in \mathcal{D}_n$, $t \in \mathcal{T}_n$, we can represent the error processes from Section 2.3 as*

$$\varepsilon_{P,t,n,i,a} = G_{P,t,n,i,a}^\varepsilon(\mathcal{H}_{t,a}^\varepsilon), \quad \xi_{P,t,n,j,b} = G_{P,t,n,j,b}^\xi(\mathcal{H}_{t,b}^\xi),$$

with $\mathbb{E}_P(\varepsilon_{P,t,n,i,a} | \mathcal{H}_t^g) = 0$ and $\mathbb{E}_P(\xi_{P,t,n,j,b} | \mathcal{H}_t^f) = 0$, where the input sequences \mathcal{H}_t^g , \mathcal{H}_t^f are defined following Assumption 3.3. $G_{P,t,n,i,a}^\varepsilon(\cdot)$ and $G_{P,t,n,j,b}^\xi(\cdot)$ are measurable functions from $(\mathbb{R}^{d_\varepsilon^\eta})^\infty$ and $(\mathbb{R}^{d_\xi^\eta})^\infty$, respectively, to \mathbb{R} —where we endow $(\mathbb{R}^{d_\varepsilon^\eta})^\infty$ and $(\mathbb{R}^{d_\xi^\eta})^\infty$ with the σ -algebra generated by all finite projections—so that $G_{P,t,n,i,a}^\varepsilon(\mathcal{H}_{s,a}^\varepsilon)$, $G_{P,t,n,j,b}^\xi(\mathcal{H}_{s,b}^\xi)$ are well-defined random variables for each $s \in \mathbb{Z}$ and $(G_{P,t,n,i,a}^\varepsilon(\mathcal{H}_{s,a}^\varepsilon))_{s \in \mathbb{Z}}$, $(G_{P,t,n,j,b}^\xi(\mathcal{H}_{s,b}^\xi))_{s \in \mathbb{Z}}$ are stationary ergodic processes.

We have not defined the input sequences for the error processes separately for each dimension. Without loss of generality, the measurable functions $G_{P,t,n,i,a}^\varepsilon(\cdot)$, $G_{P,t,n,j,b}^\xi(\cdot)$ and inputs $\eta_{t,a}^\varepsilon, \eta_{t,b}^\xi$ can be defined so that each dimension of the error processes has idiosyncratic inputs.

In view of the causal representations of the univariate error processes, we have the following causal representations for the high-dimensional nonstationary vector-valued error processes

$$\begin{aligned}\varepsilon_{P,t,n} &= \mathbf{G}_{P,t,n}^\varepsilon(\mathcal{H}_t^\varepsilon) = (G_{P,t,n,i,a}^\varepsilon(\mathcal{H}_{t,a}^\varepsilon))_{i \in [d_X], a \in A_i}, \\ \xi_{P,t,n} &= \mathbf{G}_{P,t,n}^\xi(\mathcal{H}_t^\xi) = (G_{P,t,n,j,b}^\xi(\mathcal{H}_{t,b}^\xi))_{j \in [d_Y], b \in B_j},\end{aligned}$$

where $\mathcal{H}_t^\varepsilon = (\eta_t^\varepsilon, \eta_{t-1}^\varepsilon, \dots)$, $\mathcal{H}_t^\xi = (\eta_t^\xi, \eta_{t-1}^\xi, \dots)$ with $\eta_t^\varepsilon = (\eta_{t,a}^\varepsilon)_{a \in A}$, $\eta_t^\xi = (\eta_{t,b}^\xi)_{b \in B}$ for each $t \in \mathbb{Z}$. Similarly, for each dimension/time-offset tuple $m = (i, j, a, b) \in \mathcal{D}_n$ the error products at time t can be represented as

$$R_{P,t,n,m} = G_{P,t,n,m}^R(\mathcal{H}_{t,m}^R) = G_{P,t,n,i,a}^\varepsilon(\mathcal{H}_{t,a}^\varepsilon)G_{P,t,n,j,b}^\xi(\mathcal{H}_{t,b}^\xi),$$

where $\mathcal{H}_{t,m}^R = (\eta_{t,m}^R, \eta_{t-1,m}^R, \dots)$ with $\eta_{t,m}^R = (\eta_{t,a}^\varepsilon, \eta_{t,b}^\xi)^\top$ for each $t \in \mathbb{Z}$. Also, we have the following representation for the high-dimensional nonstationary \mathbb{R}^{D_n} -valued process of all the products of errors

$$\mathbf{R}_{P,t,n} = \mathbf{G}_{P,t,n}^R(\mathcal{H}_t^R) = (G_{P,t,n,m}^R(\mathcal{H}_{t,m}^R))_{m=(i,j,a,b) \in \mathcal{D}_n},$$

where $\mathcal{H}_t^R = (\eta_t^R, \eta_{t-1}^R, \dots)$ and $\eta_t^R = (\eta_t^\varepsilon, \eta_t^\xi)^\top$ for each $t \in \mathbb{Z}$. Note that for a fixed $P \in \mathcal{P}_n$, $t \in \mathcal{T}_n$, and $n \in \mathbb{N}$ we have that $\mathbf{G}_{P,t,n}^R(\mathcal{H}_s^R)$ is a well-defined high-dimensional random vector for each $s \in \mathbb{Z}$ and $(\mathbf{G}_{P,t,n}^R(\mathcal{H}_s^R))_{s \in \mathbb{Z}}$ is a high-dimensional stationary ergodic \mathbb{R}^{D_n} -valued process.

In view of Assumptions 3.3 and 3.4, for $m = (i, j, a, b) \in \mathcal{D}_n$, we can represent the products of the errors and prediction errors as

$$\begin{aligned}\hat{w}_{P,t,n,m}^{g,\varepsilon} &= G_{P,t,n,m}^{\hat{w}^{g,\varepsilon}}(\mathcal{H}_{t,m}^{\hat{w}^{g,\varepsilon}}) = \hat{w}_{P,t,n,j,b}^g \varepsilon_{P,t,n,i,a}, \\ \hat{w}_{P,t,n,m}^{f,\xi} &= G_{P,t,n,m}^{\hat{w}^{f,\xi}}(\mathcal{H}_{t,m}^{\hat{w}^{f,\xi}}) = \hat{w}_{P,t,n,i,a}^f \xi_{P,t,n,j,b},\end{aligned}$$

with $\mathcal{H}_{t,m}^{\hat{w}^{g,\varepsilon}} = (\mathcal{H}_{t,b}^{\hat{g}}, \mathcal{H}_{t,a}^\varepsilon)$ and $\mathcal{H}_{t,m}^{\hat{w}^{f,\xi}} = (\mathcal{H}_{t,a}^{\hat{f}}, \mathcal{H}_{t,b}^\xi)$. Putting it all together, we have the following causal representation for all dimensions and time-offsets of the products of errors and prediction errors

$$\begin{aligned}\hat{w}_{P,t,n}^{g,\varepsilon} &= \mathbf{G}_{P,t,n}^{\hat{w}^{g,\varepsilon}}(\mathcal{H}_t^{\hat{w}^{g,\varepsilon}}) = (\hat{w}_{P,t,n,m}^{g,\varepsilon})_{m=(i,j,a,b) \in \mathcal{D}_n}, \\ \hat{w}_{P,t,n}^{f,\xi} &= \mathbf{G}_{P,t,n}^{\hat{w}^{f,\xi}}(\mathcal{H}_t^{\hat{w}^{f,\xi}}) = (\hat{w}_{P,t,n,m}^{f,\xi})_{m=(i,j,a,b) \in \mathcal{D}_n},\end{aligned}$$

with $\mathcal{H}_t^{\hat{w}^{g,\varepsilon}} = (\mathcal{H}_t^{\hat{g}}, \mathcal{H}_t^\varepsilon)$ and $\mathcal{H}_t^{\hat{w}^{f,\xi}} = (\mathcal{H}_t^{\hat{f}}, \mathcal{H}_t^\xi)$, where we have suppressed the dependence on n . $\mathbf{G}_{P,t,n}^{\hat{w}^{g,\varepsilon}}(\cdot)$ and $\mathbf{G}_{P,t,n}^{\hat{w}^{f,\xi}}(\cdot)$ are measurable functions such that $\mathbf{G}_{P,t,n}^{\hat{w}^{g,\varepsilon}}(\mathcal{H}_s^{\hat{w}^{g,\varepsilon}})$, $\mathbf{G}_{P,t,n}^{\hat{w}^{f,\xi}}(\mathcal{H}_s^{\hat{w}^{f,\xi}})$ are well-defined high-dimensional random vectors for each $s \in \mathbb{Z}$ and $(\mathbf{G}_{P,t,n}^{\hat{w}^{g,\varepsilon}}(\mathcal{H}_s^{\hat{w}^{g,\varepsilon}}))_{s \in \mathbb{Z}}$, $(\mathbf{G}_{P,t,n}^{\hat{w}^{f,\xi}}(\mathcal{H}_s^{\hat{w}^{f,\xi}}))_{s \in \mathbb{Z}}$ are high-dimensional stationary ergodic processes.

3.4 Assumptions on dependence and nonstationarity

We impose mild assumptions on the rate of decay in temporal dependence and the degree of non-stationarity of the error processes. Crucially, these assumptions are stated in a distribution-uniform manner, which is essential for applying the strong Gaussian approximation in Section B. This will be further elaborated upon in Section 3.5.

We quantify temporal dependence using the functional dependence measure of Wu [Wu05]. Let $(\tilde{\eta}_{t,a}^\varepsilon, \tilde{\eta}_{t,b}^\xi)_{t \in \mathbb{Z}}$ be an iid copy of $(\eta_{t,a}^\varepsilon, \eta_{t,b}^\xi)_{t \in \mathbb{Z}}$. Denote the set of well-defined tuples of error processes, dimensions, and time-offsets by

$$\mathbb{E} = \{(\varepsilon, i, a) : i \in [d_X], a \in A_i\} \cup \{(\xi, j, b) : j \in [d_Y], b \in B_j\}.$$

For any tuple $(e, l, d) \in \mathbb{E}$ corresponding to a well-defined combination of an error process, dimension, and time-offset, define

$$\tilde{\mathcal{H}}_{t,d,h}^e = (\eta_{t,d}^e, \dots, \eta_{t-h+1,d}^e, \tilde{\eta}_{t-h,d}^e, \eta_{t-h-1,d}^e, \dots)$$

to be $\mathcal{H}_{t,d}^e$ with the input $\eta_{t-h,d}^e$ replaced with the iid copy $\tilde{\eta}_{t-h,d}^e$. Similarly, define $\tilde{\mathcal{H}}_{t,m,h}^R$ as $\mathcal{H}_{t,m}^R$ with the input $\eta_{t-h,m}^R$ replaced with the iid copy $\tilde{\eta}_{t-h,m}^R$ for $m = (i, j, a, b) \in \mathcal{D}_n$, and define $\tilde{\mathcal{H}}_{t,h}^R$ as \mathcal{H}_t^R with the input η_{t-h}^R replaced with the iid copy $\tilde{\eta}_{t-h}^R$. Next, we define measures of dependence.

Definition 3.1 (Functional dependence measure). *We define the following measures of temporal dependence for each $n \in \mathbb{N}$, $P \in \mathcal{P}_n$, and $t \in \mathcal{T}_n$. First, define the L^∞ version of the functional dependence measure for the error processes $G_{P,t,n,l,d}^e(\mathcal{H}_{t,d}^e)$ for each $(e, l, d) \in \mathbb{E}$ with $h \in \mathbb{N}_0$ as*

$$\theta_{P,t,n,l,d}^{e,\infty}(h) = \inf\{K \geq 0 : \mathbb{P}_P(|G_{P,t,n,l,d}^e(\mathcal{H}_{t,d}^e) - G_{P,t,n,l,d}^e(\tilde{\mathcal{H}}_{t,d,h}^e)| > K) = 0\}.$$

Second, define the functional dependence measures for the processes of error products $G_{P,t,n,m}^R(\mathcal{H}_{t,m}^R)$ for each $m = (i, j, a, b) \in \mathcal{D}_n$ with $h \in \mathbb{N}_0$, and some $q \geq 1$ as

$$\theta_{P,t,n,m}^R(h, q) = [\mathbb{E}_P(|G_{P,t,n,m}^R(\mathcal{H}_{t,m}^R) - G_{P,t,n,m}^R(\tilde{\mathcal{H}}_{t,m,h}^R)|^q)]^{1/q},$$

and for the vector-valued process $\mathbf{G}_{P,t,n}^R(\mathcal{H}_t^R)$ with $h \in \mathbb{N}_0$, and some $q \geq 1$, $r \geq 1$ as

$$\theta_{P,t,n}^R(h, q, r) = [\mathbb{E}_P(\|\mathbf{G}_{P,t,n}^R(\mathcal{H}_t^R) - \mathbf{G}_{P,t,n}^R(\tilde{\mathcal{H}}_{t,h}^R)\|_r^q)]^{1/q}.$$

For some sequence of collections of distributions $(\mathcal{P}_n)_{n \in \mathbb{N}}$, we make the following assumption about the temporal dependence. We only require the relatively mild assumption that it decays polynomially, rather than geometrically. Note that we will often write the time of the input sequence as 0 when it does not matter due to stationarity.

Assumption 3.5 (Distribution-uniform decay of temporal dependence). *Assume that there exist $\bar{\Theta}^\infty > 0$, $\bar{\beta}^\infty > 1$ such that for all $n \in \mathbb{N}$, $t \in \mathcal{T}_n$, and error processes $(e, l, d) \in \mathbb{E}$, it holds that*

$$\sup_{P \in \mathcal{P}_n} \|G_{P,t,n,l,d}^e(\mathcal{H}_{0,d}^e)\|_{L^\infty(P)} \leq \bar{\Theta}^\infty, \quad \sup_{P \in \mathcal{P}_n} \theta_{P,t,n,l,d}^{e,\infty}(h) \leq \bar{\Theta}^\infty \cdot (h \vee 1)^{-\bar{\beta}^\infty}, \quad h \geq 0.$$

For additional control in terms of the product of errors alone, also assume that there exist $\bar{\Theta}^R > 0$, $\bar{\beta}^R > 3$, $\bar{q}^R > 4$, such that for all $n \in \mathbb{N}$, $t \in \mathcal{T}_n$, $m = (i, j, a, b) \in \mathcal{D}_n$, it holds that

$$\sup_{P \in \mathcal{P}_n} [\mathbb{E}_P(|G_{P,t,n,m}^R(\mathcal{H}_{0,m}^R)|^{q^R})]^{1/q^R} \leq \bar{\Theta}^R, \quad \sup_{P \in \mathcal{P}_n} \theta_{P,t,n,m}^R(h, \bar{q}^R) \leq \bar{\Theta}^R \cdot (h \vee 1)^{-\bar{\beta}^R}, \quad h \geq 0.$$

A few remarks are in order. First, the constants in Assumption 3.5 do not depend on n . Second, the assumptions on the individual error processes can be weakened; see Section C.8 for more discussion. Third, for all $n \in \mathbb{N}$, $t \in \mathcal{T}_n$, by Jensen's inequality we have

$$\sup_{P \in \mathcal{P}_n} [\mathbb{E}_P(\|\mathbf{G}_{P,t,n}^R(\mathcal{H}_0^R)\|_2^{q^R})]^{1/q^R} \leq D_n^{\frac{1}{2}} \bar{\Theta}^R, \quad \sup_{P \in \mathcal{P}_n} \theta_{P,t,n}^R(h, \bar{q}^R, 2) \leq D_n^{\frac{1}{2}} \bar{\Theta}^R \cdot (h \vee 1)^{-\bar{\beta}^R}, \quad h \geq 0.$$

Next, for some sequence of collections of distributions $(\mathcal{P}_n)_{n \in \mathbb{N}}$, we make the following assumption to control the nonstationarity of the process of error products.

Assumption 3.6 (Distribution-uniform total variation condition for nonstationarity). *Recall $\bar{\Theta}^R > 0$ from Assumption 3.5. Assume that for each $n \in \mathbb{N}$, there exists a constant $\bar{\Gamma}_n^R \geq 1$ such that*

$$\sup_{P \in \mathcal{P}_n} \left(\sum_{t=\mathbb{T}_n+1}^{\mathbb{T}_n^+} (\mathbb{E}_P \|\mathbf{G}_{P,t,n}^R(\mathcal{H}_0^R) - \mathbf{G}_{P,t-1,n}^R(\mathcal{H}_0^R)\|_2^2)^{1/2} \right) \leq \bar{\Theta}^R \bar{\Gamma}_n^R.$$

3.5 Theoretical result for dGCM

We present the theoretical result that justifies the bootstrap procedure described in Algorithm 1. This result relies on time-varying nonlinear regression for nonstationary processes and the distribution-uniform strong Gaussian approximation from Section B applied to the process of error products. The approximating nonstationary Gaussian process has a time-varying covariance structure, which is explicitly characterized by local long-run covariance matrices.

Definition 3.2 (Local long-run covariance matrices of the process of error products). *For each $P \in \mathcal{P}_n$, $t \in \mathcal{T}_n$, $n \in \mathbb{N}$, define the local long-run covariance matrix $\Sigma_{P,t,n}^R \in \mathbb{R}^{D_n \times D_n}$ of the \mathbb{R}^{D_n} -valued stationary process $(\mathbf{G}_{P,t,n}^R(\mathcal{H}_t^R))_{t \in \mathbb{Z}}$ by*

$$\Sigma_{P,t,n}^R = \sum_{h \in \mathbb{Z}} \text{Cov}_P(\mathbf{G}_{P,t,n}^R(\mathcal{H}_0^R), \mathbf{G}_{P,t,n}^R(\mathcal{H}_h^R)).$$

In view of the Gaussian approximation theory developed in Mies and Steland [MS23], we only require an estimator of the cumulative covariance matrices of the error products

$$Q_{P,t,n}^R = \sum_{s=\mathbb{T}_n^-}^t \Sigma_{P,s,n}^R,$$

rather than the local long-run covariance matrices at each time individually. This is critical for the practical applicability of our method, as estimating individual local long-run covariance matrices can be extremely challenging in practice. Specifically, we use the estimator

$$\hat{Q}_{t,n}^R = \sum_{s=L_n+\mathbb{T}_n^--1}^t \frac{1}{L_n} \left(\sum_{r=s-L_n+1}^s \hat{R}_{r,n} \right)^{\otimes 2}, \quad (12)$$

for some lag-window size $L_n \in \mathbb{N}$. We discuss how to select L_n in practice using the minimum volatility method in Section 5.1. Going forward, denote $Q_{P,n}^R = (Q_{P,t,n}^R)_{t \in \mathcal{T}_{n,L}}$ and $\hat{Q}_n^R = (\hat{Q}_{t,n}^R)_{t \in \mathcal{T}_{n,L}}$, where

$$\mathcal{T}_{n,L} = \{L_n + \mathbb{T}_n^- - 1, \dots, \mathbb{T}_n^+ - 1, \mathbb{T}_n^+\}.$$

To account for the estimation errors for the time-varying regression functions and the cumulative covariance matrices, as well as the error for the Gaussian approximation, we introduce offsets $\tau_n \rightarrow 0$, $\nu_n \rightarrow 0$ so that $\tau_n = o(\log^{-(1+\delta)}(T_n))$ for some $\delta > 0$ and

$$\nu_n \gg \log(T_n) D_n \left[\left(\frac{D_n}{T_n} \right)^{2\xi(\bar{q}^R, \bar{\beta}^R)} + \tau_n^{-2} (\varphi_{n,1} + \varphi_{n,2}) \right], \quad (13)$$

where

$$\varphi_{n,1} = T_n^{-\frac{1}{2}} (\bar{\Gamma}_n^R)^{\frac{1}{2}} L_n^{\frac{1}{4}} + T_n^{-\frac{1}{4}} D_n^{\frac{1}{4}} L_n^{\frac{1}{4}} + L_n^{-\frac{1}{2}} + L_n^{1-\frac{\bar{\beta}^R}{2}} + T_n^{-1},$$

comes from the covariance estimation error, and

$$\varphi_{n,2} = \tau_n^{\frac{7}{2}} D_n^{-\frac{5}{4}} + \tau_n^7 D_n^{-\frac{5}{2}},$$

comes from the time-varying regression function estimation errors. Also, the lag-window size L_n from (12) must satisfy $L_n \asymp T_n^\zeta$ for some $\zeta \in (0, \frac{1}{2})$ so that $\tau_n^{-6} D_n^2 L_n^{-1} = o(1)$ and $\bar{\Gamma}_n^R T_n^{-1} D_n^2 \tau_n^{-6} L_n^{\frac{1}{2}} = o(1)$, where $\bar{\Gamma}_n^R$ is from Assumption 3.6. We see that the offsets depend on the number of observations T_n from Section 2.1, the intrinsic dimensionality D_n from Section 2.1, the degree of nonstationarity $\bar{\Gamma}_n^R$ from Assumption 3.6, and the lag-window parameter L_n from (12). $\xi(\bar{q}^R, \bar{\beta}^R)$ is a rate defined in Section B that depends on the constants $\bar{\beta}^R, \bar{q}^R$ from Assumption 3.5.

The following result establishes the validity of our bootstrap-based testing procedure described in Algorithm 1, provided that the previously stated assumptions hold and the prediction errors

$$\begin{aligned} \hat{w}_{P,t,n,i,a}^f &= f_{P,t,n,i,a}(\mathbf{Z}_{t,n}) - \hat{f}_{t,n,i,a}(\mathbf{Z}_{t,n}), \\ \hat{w}_{P,t,n,j,b}^g &= g_{P,t,n,j,b}(\mathbf{Z}_{t,n}) - \hat{g}_{t,n,j,b}(\mathbf{Z}_{t,n}), \end{aligned}$$

converge to zero sufficiently fast, in some sense. If it were known, we could correctly calibrate our test with the (random) quantile function \hat{q} of $S_{n,p}(\check{\mathbf{R}}_n)$, where $\check{\mathbf{R}}_n = (\check{\mathbf{R}}_{t,n})_{t \in \mathcal{T}_{n,L}}$ and $\check{\mathbf{R}}_{t,n} \sim \mathcal{N}(0, \hat{\Sigma}_{t,n}^R)$ for all $t \in \mathcal{T}_{n,L}$. In practice, \hat{q} is numerically approximated by conducting a large number of Monte Carlo simulations, and we use \hat{q}^{boot} from Algorithm 1 in its place.

Theorem 3.1. *Suppose that Assumptions 3.1, 3.2, 3.3, 3.4, 3.5, 3.6 related to the temporal dependence and nonstationarity of the processes all hold for the sequence of collections of distributions $(\mathcal{P}_{0,n}^*)_{n \in \mathbb{N}}$, where $\mathcal{P}_{0,n}^* \subset \mathcal{P}_{0,n}^{\text{CI}}$ for each $n \in \mathbb{N}$. Further, suppose that*

$$\begin{aligned} \sup_{P \in \mathcal{P}_{0,n}^*} \max_{(i,j,a,b) \in \mathcal{D}_n} \max_{t \in \mathcal{T}_n} \mathbb{E}_P \left(\left| \hat{w}_{P,t,n,i,a}^f \right|^2 \right)^{\frac{1}{2}} \mathbb{E}_P \left(\left| \hat{w}_{P,t,n,j,b}^g \right|^2 \right)^{\frac{1}{2}} &= o(T_n^{-\frac{1}{2}} \tau_n^7 D_n^{-\frac{3}{2}}), \\ \sup_{P \in \mathcal{P}_{0,n}^*} \max_{i \in [d_X], a \in A_i} \max_{t \in \mathcal{T}_n} \mathbb{E}_P \left(\left| \hat{w}_{P,t,n,i,a}^f \right|^2 \right)^{\frac{1}{2}} &= o(\tau_n^7 D_n^{-\frac{5}{2}}), \\ \sup_{P \in \mathcal{P}_{0,n}^*} \max_{j \in [d_Y], b \in B_j} \max_{t \in \mathcal{T}_n} \mathbb{E}_P \left(\left| \hat{w}_{P,t,n,j,b}^g \right|^2 \right)^{\frac{1}{2}} &= o(\tau_n^7 D_n^{-\frac{5}{2}}). \end{aligned}$$

If the offsets $\tau_n \rightarrow 0$ and $\nu_n \rightarrow 0$ are chosen such that condition (13) holds, then we have

$$\limsup_{n \rightarrow \infty} \sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{P}_P \left(S_{n,p}(\hat{\mathbf{R}}_n) > \hat{q}_{1-\alpha+\nu_n} + \tau_n \right) \leq \alpha.$$

The above result demonstrates that the dGCM test possesses a property known as rate double robustness. This property means that we only require modest convergence rates for the products of the prediction errors, rather than for each prediction error individually. This feature of the dGCM test can be especially useful in the contexts of causal discovery for time-lagged effects and variable selection in time series forecasting. In these applications, a faster convergence rate of a nowcasting model can compensate for a slower convergence rate of a forecasting model, or vice versa.

4 dGCM with Sieve Time-Varying Regression (Sieve-dGCM)

The purpose of this section is to demonstrate that the convergence rates required by Theorem 3.1 for estimating the time-varying regression functions can be achieved. To show this, we consider an instantiation of the dynamic generalized covariance measure (dGCM) test based on the sieve time-varying nonlinear regression estimator from Ding and Zhou [DZ21]. We refer to this instantiation of the dGCM test as the Sieve-dGCM test.

We prove that, under mild assumptions about the temporal dependence and nonstationarity of the processes, the sieve estimator achieves the required convergence rates and the Sieve-dGCM test has asymptotic Type-I error control. In Section 5, we study the finite sample performance of the Sieve-dGCM test. Along the way, in Section 5.1, we introduce a novel cross-validation scheme which we use for selecting the parameters of the sieve estimator.

In this section, we use the framework of locally stationary processes [Dah97; ZW09; Dah12; DRW19]. This is a well-studied class of nonstationary processes that fits within the general triangular array framework for nonstationary processes from Section 3. We note that there are several other time-varying regression estimators for locally stationary processes; see [ZW15; YN21; CSW22].

4.1 Setting and notation

We follow Dahlhaus [Dah97] in rescaling time to the unit interval $t/n \in [0, 1]$, so that infill asymptotics can be used to study nonstationary processes. In this setting, the sample size n no longer corresponds to getting information about the future. Instead, as n increases we get more observations about each local structure of the nonstationary process. Zhou and Wu [ZW09] introduced the framework for representing locally stationary processes as nonlinear functions of iid inputs as in Wu [Wu05].

We use the same notation as Section 2.1, with the only difference being that we fix the number of dimensions d_Z and time-offsets C_k for each dimension $k \in [d_Z]$. We still allow the number of dimensions $d_X = d_{X,n}$, $d_Y = d_{Y,n}$ and time-offsets A_i , B_j for each $i \in [d_X]$, $j \in [d_Y]$ to grow with n . Define A , B , C as the collection of all time-offsets as in Section 2.1, where $A = A_n$, $B = B_n$ and C is fixed. We emphasize that there is no inherent necessity for fixing the number of dimensions d_Z and time-offsets C_k . Our reason for doing this is because we want to leverage the existing theoretical results for the sieve estimator from Ding and Zhou [DZ21]. Future investigations can study the performance of the sieve estimator in the high-dimensional setting, so that we can allow the number of dimensions d_Z and time-offsets C_k to grow with n .

We will still use the notation \mathcal{T}_n for the subset of original times in which all time-offsets of each dimension of $X_{t,n}$, $Y_{t,n}$, and $Z_{t,n}$ are actually observed,

$$\mathcal{T}_n = \{1 - \min(a_{\min}, b_{\min}, c_{\min}), n - \max(a_{\max}, b_{\max}, c_{\max})\} \subseteq \{1, \dots, n\}.$$

Also, we will still denote $T_n = |\mathcal{T}_n|$, $\mathbb{T}_n^- = \min(\mathcal{T}_n)$, and $\mathbb{T}_n^+ = \max(\mathcal{T}_n)$. Similarly, denote the corresponding interval of rescaled times in which all time-offsets are well-defined by

$$\mathcal{U}_n = \left[\frac{1}{n} - \frac{\min(a_{\min}, b_{\min}, c_{\min})}{n}, 1 - \frac{\max(a_{\max}, b_{\max}, c_{\max})}{n} \right] \subset [0, 1],$$

and denote $\mathbb{U}_n^- = \min(\mathcal{U}_n)$, and $\mathbb{U}_n^+ = \max(\mathcal{U}_n)$.

Recall the index set containing the dimensions and time-offsets of interest

$$\mathcal{D}_n \subseteq \{(i, j, a, b) : i \in [d_X], j \in [d_Y], a \in A_i, b \in B_j\},$$

where A_i , B_j are the time-offsets for dimensions $i \in [d_X]$, $j \in [d_Y]$. Again, we will often refer to the dimension/time-offset tuple by $m = (i, j, a, b) \in \mathcal{D}_n$ to lighten the notation. Denote cardinality $D_n = |\mathcal{D}_n|$ which may grow with n .

4.2 Locally stationary observed processes

Next, we introduce the causal representation of locally stationary processes, which is most similar to Ding and Zhou [DZ21] and Example 3 in Mies and Steland [MS23]. This representation is different than the previous causal representation from Assumption 3.1, because we now assume that the nonlinear stochastic system is well-defined for all rescaled times. For the following assumption, let

$$\mathcal{H}_t^X = (\eta_t^X, \eta_{t-1}^X, \dots), \quad \mathcal{H}_t^Y = (\eta_t^Y, \eta_{t-1}^Y, \dots), \quad \mathcal{H}_t^Z = (\eta_t^Z, \eta_{t-1}^Z, \dots),$$

where $(\eta_t^X, \eta_t^Y, \eta_t^Z)_{t \in \mathbb{Z}}$ is an iid sequence of random vectors. Denote the dimensions of $\eta_t^X = \eta_{t,n}^X$, $\eta_t^Y = \eta_{t,n}^Y$, $\eta_t^Z = \eta_{t,n}^Z$ respectively by $d_X^\eta = d_{X,n}^\eta$, $d_Y^\eta = d_{Y,n}^\eta$, $d_Z^\eta = d_{Z,n}^\eta$, which can change with n .

Assumption 4.1 (Causal representations of the observed processes). *Assume that we can represent each dimension of each of the observed processes as the output of an evolving nonlinear system that was given a sequence of iid inputs:*

$$X_{t,n,i} = \tilde{G}_{n,i}^X(t/n, \mathcal{H}_t^X), \quad Y_{t,n,j} = \tilde{G}_{n,j}^Y(t/n, \mathcal{H}_t^Y), \quad Z_{t,n,k} = \tilde{G}_{n,k}^Z(t/n, \mathcal{H}_t^Z),$$

where the systems are defined for all $u \in [0, 1]$ by

$$\tilde{X}_{t,n,i}(u) = \tilde{G}_{n,i}^X(u, \mathcal{H}_t^X), \quad \tilde{Y}_{t,n,j}(u) = \tilde{G}_{n,j}^Y(u, \mathcal{H}_t^Y), \quad \tilde{Z}_{t,n,k}(u) = \tilde{G}_{n,k}^Z(u, \mathcal{H}_t^Z),$$

so that we have $X_{t,n,i} = \tilde{X}_{t,n,i}(t/n)$, $Y_{t,n,j} = \tilde{Y}_{t,n,j}(t/n)$, $Z_{t,n,k} = \tilde{Z}_{t,n,k}(t/n)$.

For each $n \in \mathbb{N}$, $(i, j, a, b) \in \mathcal{D}_n$, $t \in \mathcal{T}_n$, we assume that $\tilde{G}_{n,i}^X(u, \cdot)$, $\tilde{G}_{n,j}^Y(u, \cdot)$, $\tilde{G}_{n,k}^Z(u, \cdot)$ are measurable functions from $(\mathbb{R}^{d_X^\eta})^\infty$, $(\mathbb{R}^{d_Y^\eta})^\infty$, $(\mathbb{R}^{d_Z^\eta})^\infty$, respectively, to \mathbb{R} —where we endow $(\mathbb{R}^{d_X^\eta})^\infty$, $(\mathbb{R}^{d_Y^\eta})^\infty$, $(\mathbb{R}^{d_Z^\eta})^\infty$ with the σ -algebra generated by all finite projections—such that $\tilde{G}_{n,i}^X(u, \mathcal{H}_s^X)$, $\tilde{G}_{n,j}^Y(u, \mathcal{H}_s^Y)$, $\tilde{G}_{n,k}^Z(u, \mathcal{H}_s^Z)$ are each well-defined random variables for each $s \in \mathbb{Z}$ and $(\tilde{G}_{n,i}^X(u, \mathcal{H}_s^X))_{s \in \mathbb{Z}}$, $(\tilde{G}_{n,j}^Y(u, \mathcal{H}_s^Y))_{s \in \mathbb{Z}}$, $(\tilde{G}_{n,k}^Z(u, \mathcal{H}_s^Z))_{s \in \mathbb{Z}}$ are each stationary ergodic processes.

As in Section 3.1, we have not defined the input sequences for the observed processes separately for each dimension. However, without loss of generality, we can define the measurable functions $\tilde{G}_{n,i}^X(u, \cdot)$, $\tilde{G}_{n,j}^Y(u, \cdot)$, $\tilde{G}_{n,k}^Z(u, \cdot)$ and the inputs η_t^X , η_t^Y , η_t^Z so that each dimension of the observed processes can have idiosyncratic inputs.

In light of Assumption 4.1, we have the following causal representations for all dimensions with no time-offsets:

$$\begin{aligned} \tilde{X}_{t,n}(u) &= \tilde{G}_n^X(u, \mathcal{H}_t^X) = (\tilde{G}_{n,i}^X(u, \mathcal{H}_t^X))_{i \in [d_X]}, \\ \tilde{Y}_{t,n}(u) &= \tilde{G}_n^Y(u, \mathcal{H}_t^Y) = (\tilde{G}_{n,j}^Y(u, \mathcal{H}_t^Y))_{j \in [d_Y]}, \\ \tilde{Z}_{t,n}(u) &= \tilde{G}_n^Z(u, \mathcal{H}_t^Z) = (\tilde{G}_{n,k}^Z(u, \mathcal{H}_t^Z))_{k \in [d_Z]}, \end{aligned}$$

so that we have $X_{t,n} = \tilde{X}_{t,n}(t/n)$, $Y_{t,n} = \tilde{Y}_{t,n}(t/n)$, $Z_{t,n} = \tilde{Z}_{t,n}(t/n)$. For each $n \in \mathbb{N}$, we have causal representations for dimensions $i \in [d_X]$, $j \in [d_Y]$, $k \in [d_Z]$ with time-offsets $a \in A_i$, $b \in B_j$, $c \in C_k$

$$\begin{aligned} \tilde{X}_{t,n,i,a}(u) &= \tilde{G}_{n,i,a}^X(u, \mathcal{H}_{t,a}^X) = \tilde{G}_{n,i}^X(u + a/n, \mathcal{H}_{t+a}^X), \\ \tilde{Y}_{t,n,j,b}(u) &= \tilde{G}_{n,j,b}^Y(u, \mathcal{H}_{t,b}^Y) = \tilde{G}_{n,j}^Y(u + b/n, \mathcal{H}_{t+b}^Y), \\ \tilde{Z}_{t,n,k,c}(u) &= \tilde{G}_{n,k,c}^Z(u, \mathcal{H}_{t,c}^Z) = \tilde{G}_{n,k}^Z(u + c/n, \mathcal{H}_{t+c}^Z), \end{aligned}$$

where $\mathcal{H}_{t,a}^X = (\eta_{t+a}^X, \eta_{t-1+a}^X, \dots)$, $\mathcal{H}_{t,b}^Y = (\eta_{t+b}^Y, \eta_{t-1+b}^Y, \dots)$, and $\mathcal{H}_{t,c}^Z = (\eta_{t+c}^Z, \eta_{t-1+c}^Z, \dots)$, so that we have $X_{t,n,i,a} = \tilde{X}_{t,n,i,a}(t/n)$, $Y_{t,n,j,b} = \tilde{Y}_{t,n,j,b}(t/n)$, $Z_{t,n,k,c} = \tilde{Z}_{t,n,k,c}(t/n)$ for each dimension of the observed sequence with time-offset. We can then write the causal representation of the vectors with all dimensions and time-offsets as

$$\begin{aligned} \tilde{\mathbf{X}}_{t,n}(u) &= \tilde{\mathbf{G}}_n^X(u, \mathcal{H}_t^X) = (\tilde{G}_{n,i,a}^X(u, \mathcal{H}_{t,a}^X))_{i \in [d_X], a \in A_i}, \\ \tilde{\mathbf{Y}}_{t,n}(u) &= \tilde{\mathbf{G}}_n^Y(u, \mathcal{H}_t^Y) = (\tilde{G}_{n,j,b}^Y(u, \mathcal{H}_{t,b}^Y))_{j \in [d_Y], b \in B_j}, \\ \tilde{\mathbf{Z}}_{t,n}(u) &= \tilde{\mathbf{G}}_n^Z(u, \mathcal{H}_t^Z) = (\tilde{G}_{n,k,c}^Z(u, \mathcal{H}_{t,c}^Z))_{k \in [d_Z], c \in C_k}, \end{aligned}$$

where $\mathcal{H}_t^X = (\eta_t^X, \eta_{t-1}^X, \dots)$, $\mathcal{H}_t^Y = (\eta_t^Y, \eta_{t-1}^Y, \dots)$, $\mathcal{H}_t^Z = (\eta_t^Z, \eta_{t-1}^Z, \dots)$, and $\eta_t^X = \eta_{t+a_{\max}}^X$, $\eta_t^Y = \eta_{t+b_{\max}}^Y$, $\eta_t^Z = \eta_{t+c_{\max}}^Z$, so that we have $\mathbf{X}_{t,n} = \tilde{\mathbf{X}}_{t,n}(t/n)$, $\mathbf{Y}_{t,n} = \tilde{\mathbf{Y}}_{t,n}(t/n)$, $\mathbf{Z}_{t,n} = \tilde{\mathbf{Z}}_{t,n}(t/n)$ for the observed sequence including all dimensions and time-offsets.

Let Ω be a sample space, \mathcal{B} the Borel sigma-algebra, and (Ω, \mathcal{B}) a measurable space. For fixed $n \in \mathbb{N}$, let (Ω, \mathcal{B}) be equipped with a family of probability measures $(\mathbb{P}_P)_{P \in \mathcal{P}_n}$ so that the joint distribution of the nonlinear stochastic systems

$$(\tilde{G}_n^X(u, \mathcal{H}_t^X))_{u \in [0,1], t \in \mathbb{Z}}, (\tilde{G}_n^Y(u, \mathcal{H}_t^Y))_{u \in [0,1], t \in \mathbb{Z}}, (\tilde{G}_n^Z(u, \mathcal{H}_t^Z))_{u \in [0,1], t \in \mathbb{Z}}$$

under \mathbb{P}_P is $P \in \mathcal{P}_n$, where the collection of distributions \mathcal{P}_n can change with n . The family of probability measures $(\mathbb{P}_P)_{P \in \mathcal{P}_n}$ is defined with respect to the same measurable space (Ω, \mathcal{B}) , but need not have the same dominating measure.

We use the same null hypotheses of conditional independence as those in Section 2.2. Again, for each $n \in \mathbb{N}$, we denote the collection of distributions such that the null hypothesis is true by $\mathcal{P}_{0,n}^{\text{CI}}$. In the locally stationary setting, the null hypothesis

$$X_{t,n,i,a} \perp\!\!\!\perp Y_{t,n,j,b} \mid \mathbf{Z}_{t,n} \text{ for all } t \in \mathcal{T}_n, \text{ for all } (i, j, a, b) \in \mathcal{D}_n, \quad (14)$$

can be written equivalently as

$$\tilde{X}_{t,n,i,a}(t/n) \perp\!\!\!\perp \tilde{Y}_{t,n,j,b}(t/n) \mid \tilde{\mathbf{Z}}_{t,n}(t/n) \text{ for all } t \in \mathcal{T}_n, \text{ for all } (i, j, a, b) \in \mathcal{D}_n,$$

where \mathcal{D}_n only contains a single dimension/time-offset tuple in the univariate setting.

We will state more assumptions in the rest of this section for a generic sequence of collections of distributions $(\mathcal{P}_n)_{n \in \mathbb{N}}$. In Theorem 4.1, we will assume that these conditions hold for the sequence of collections of distributions $(\mathcal{P}_{0,n}^*)_{n \in \mathbb{N}}$, where $\mathcal{P}_{0,n}^* \subset \mathcal{P}_{0,n}^{\text{CI}}$ for each $n \in \mathbb{N}$. Note that we make stronger assumptions in this section than in Section 3 to ensure that the sieve estimators satisfy the convergence rate requirements of Theorem 3.1.

4.3 Sieve time-varying nonlinear regression estimator

For a given sample size $n \in \mathbb{N}$, distribution $P \in \mathcal{P}_n$, time $t \in \mathcal{T}_n$, and dimension/time-offset tuple $(i, j, a, b) \in \mathcal{D}_n$, we consider the time-varying nonlinear regression model

$$X_{t,n,i,a} = f_{P,n,i,a}(t/n, \mathbf{Z}_{t,n}) + \varepsilon_{P,t,n,i,a}, \quad Y_{t,n,j,b} = g_{P,n,j,b}(t/n, \mathbf{Z}_{t,n}) + \xi_{P,t,n,j,b},$$

where $f_{P,n,i,a}(u, \mathbf{z})$ and $g_{P,n,j,b}(u, \mathbf{z})$ are smooth functions of rescaled time u and covariate values \mathbf{z} with $f_{P,n,i,a}(t/n, \mathbf{z}) = \mathbb{E}_P(X_{t,n,i,a} \mid \mathbf{Z}_{t,n} = \mathbf{z})$ and $g_{P,n,j,b}(t/n, \mathbf{z}) = \mathbb{E}_P(Y_{t,n,j,b} \mid \mathbf{Z}_{t,n} = \mathbf{z})$. We emphasize that the functions $f_{P,n,i,a}(u, \mathbf{z})$ and $g_{P,n,j,b}(u, \mathbf{z})$ depend on rescaled time u rather than “real time” t , as in the literature on nonparametric regression for locally stationary processes [Vog12; ZW15; YN21; CSW22; DZ21]. For $m = (i, j, a, b) \in \mathcal{D}_n$, denote the error products at time t by

$$R_{P,t,n,m} = \varepsilon_{P,t,n,i,a} \xi_{P,t,n,j,b},$$

and the corresponding residual products by

$$\hat{R}_{t,n,m} = \hat{\varepsilon}_{t,n,i,a} \hat{\xi}_{t,n,j,b},$$

where $\hat{\varepsilon}_{t,n,i,a} = X_{t,n,i,a} - \hat{f}_{t,n,i,a}(t/n, \mathbf{Z}_{t,n})$ and $\hat{\xi}_{t,n,j,b} = Y_{t,n,j,b} - \hat{g}_{t,n,j,b}(t/n, \mathbf{Z}_{t,n})$.

The estimates $\hat{f}_{t,n,i,a}$ and $\hat{g}_{t,n,j,b}$ of the functions $f_{P,n,i,a}$ and $g_{P,n,j,b}$ are formed by regressing $(X_{t,n,i,a})_{t \in \mathcal{T}_n}$ on $(\mathbf{Z}_{t,n})_{t \in \mathcal{T}_n}$ and $(Y_{t,n,j,b})_{t \in \mathcal{T}_n}$ on $(\mathbf{Z}_{t,n})_{t \in \mathcal{T}_n}$, respectively, using the time-varying nonlinear sieve regression estimator introduced below. The subscript t in $\hat{f}_{t,n,i,a}$ and $\hat{g}_{t,n,j,b}$ is to indicate that we allow for sequential estimation, which will be discussed in Remark 4.1. Let $\hat{\mathbf{R}}_{t,n} = (\hat{R}_{t,n,m})_{m \in \mathcal{D}_n}$ be the high-dimensional vector process containing the residual products for all dimension/time-offset combinations in \mathcal{D}_n . The observed processes X, Y, Z and error processes ε, ξ can all be locally stationary processes; see Section 4.5 for the details.

For some sequence of collections of distributions $(\mathcal{P}_n)_{n \in \mathbb{N}}$, we make the following assumption.

Assumption 4.2 (Additive form and regularity). *For each sample size $n \in \mathbb{N}$, distribution $P \in \mathcal{P}_n$, rescaled time $u \in \mathcal{U}_n$, and dimension/time-offset tuple $(i, j, a, b) \in \mathcal{D}_n$, assume that*

$$f_{P,n,i,a}(u, \mathbf{z}) = \sum_{k=1}^{d_Z} \sum_{c=1}^{C_k} f_{P,n,i,a,k,c}(u, z_{k,c}),$$

$$g_{P,n,j,b}(u, \mathbf{z}) = \sum_{k=1}^{d_Z} \sum_{c=1}^{C_k} g_{P,n,j,b,k,c}(u, z_{k,c}),$$

where $f_{P,n,i,a,k,c} : \mathcal{U}_n \times \mathbb{R} \rightarrow \mathbb{R}$ and $g_{P,n,j,b,k,c} : \mathcal{U}_n \times \mathbb{R} \rightarrow \mathbb{R}$ are time-varying partial response functions, so that we have

$$\mathbb{E}_P(X_{t,n,i,a} | \mathbf{Z}_{t,n} = \mathbf{z}) = \sum_{k=1}^{d_Z} \sum_{c=1}^{C_k} f_{P,n,i,a,k,c}(t/n, z_{k,c}),$$

$$\mathbb{E}_P(Y_{t,n,j,b} | \mathbf{Z}_{t,n} = \mathbf{z}) = \sum_{k=1}^{d_Z} \sum_{c=1}^{C_k} g_{P,n,j,b,k,c}(t/n, z_{k,c}),$$

for each time $t \in \mathcal{T}_n$.

Further, assume for all $n \in \mathbb{N}$, $i \in [d_X]$, $a \in A_i$, $j \in [d_Y]$, $b \in B_j$, $k \in [d_Z]$, $c \in C_k$, $u \in \mathcal{U}_n$, there exists some $q \geq 2$ such that

$$\sup_{P \in \mathcal{P}_n} \mathbb{E}_P(|f_{P,n,i,a,k,c}(u, Z_{t,n,k,c})|^q) < \infty,$$

$$\sup_{P \in \mathcal{P}_n} \mathbb{E}_P(|g_{P,n,j,b,k,c}(u, Z_{t,n,k,c})|^q) < \infty.$$

To fix ideas, we use the algebraic mapping $h : [-1, 1] \rightarrow \mathbb{R}$ from Example 3.1 in Ding and Zhou [DZ21] with positive scaling factor $s = 1$,

$$h(\tilde{z}) = \begin{cases} -\infty, & \tilde{z} = -1, \\ \frac{\tilde{z}}{\sqrt{1-\tilde{z}^2}}, & \tilde{z} \in (-1, 1), \\ \infty, & \tilde{z} = 1. \end{cases}$$

See the discussion preceding Definition 3.1 in Ding and Zhou [DZ21] for additional details. For some sequence of collections of distributions $(\mathcal{P}_n)_{n \in \mathbb{N}}$, for each $n \in \mathbb{N}$, $i \in [d_X]$, $a \in A_i$, $j \in [d_Y]$, $b \in B_j$, $k \in [d_Z]$, $c \in C_k$, and $P \in \mathcal{P}_n$, we relate the time-varying partial response functions $f_{P,n,i,a,k,c} : \mathcal{U}_n \times \mathbb{R} \rightarrow \mathbb{R}$ and $g_{P,n,j,b,k,c} : \mathcal{U}_n \times \mathbb{R} \rightarrow \mathbb{R}$ to $\tilde{f}_{P,n,i,a,k,c} : [0, 1] \times [0, 1] \rightarrow \mathbb{R}$ and $\tilde{g}_{P,n,j,b,k,c} : [0, 1] \times [0, 1] \rightarrow \mathbb{R}$, respectively, where

$$\tilde{f}_{P,n,i,a,k,c}(u^*, z^*) = f_{P,n,i,a,k,c}(\mathbb{U}_n^- + u^*(\mathbb{U}_n^+ - \mathbb{U}_n^-), h(2z^* - 1)),$$

$$\tilde{g}_{P,n,j,b,k,c}(u^*, z^*) = g_{P,n,j,b,k,c}(\mathbb{U}_n^- + u^*(\mathbb{U}_n^+ - \mathbb{U}_n^-), h(2z^* - 1)),$$

with $\mathbb{U}_n^- = \min(\mathcal{U}_n)$ and $\mathbb{U}_n^+ = \max(\mathcal{U}_n)$.

For some sequence of collections of distributions $(\mathcal{P}_n)_{n \in \mathbb{N}}$, we make the following assumption.

Assumption 4.3 (Smoothness). *For each $n \in \mathbb{N}$, $i \in [d_X]$, $a \in A_i$, $j \in [d_Y]$, $b \in B_j$, $k \in [d_Z]$, $c \in C_k$, and $P \in \mathcal{P}_n$, assume that for each fixed $u^* \in [0, 1]$ we have*

$$\tilde{f}_{P,n,i,a,k,c}(u^*, \cdot) \in C^\infty([0, 1]), \quad \tilde{g}_{P,n,j,b,k,c}(u^*, \cdot) \in C^\infty([0, 1]),$$

and for each fixed $z^* \in [0, 1]$ we have

$$\tilde{f}_{P,n,i,a,k,c}(\cdot, z^*) \in C^\infty([0, 1]), \quad \tilde{g}_{P,n,j,b,k,c}(\cdot, z^*) \in C^\infty([0, 1]),$$

where $C^\infty([0, 1])$ denotes the space of functions on $[0, 1]$ that are infinitely differentiable.

If Assumption 4.3 holds, then by Theorem 3.1 of Ding and Zhou [DZ21] we can approximate the time-varying partial response functions by

$$\begin{aligned} f_{P,n,i,a,k,c}(u, z) &\approx \sum_{\ell_1=1}^{\tilde{c}_n} \sum_{\ell_2=1}^{\tilde{d}_n} \beta_{P,n,i,a,k,c,\ell_1,\ell_2}^f b_{\ell_1,\ell_2}(u, z), \\ g_{P,n,j,b,k,c}(u, z) &\approx \sum_{\ell_1=1}^{\tilde{c}_n} \sum_{\ell_2=1}^{\tilde{d}_n} \beta_{P,n,j,b,k,c,\ell_1,\ell_2}^g b_{\ell_1,\ell_2}(u, z), \end{aligned}$$

where $\{b_{\ell_1,\ell_2}(u, z)\} = \{\phi_{\ell_1}(u)\varphi_{\ell_2}(z)\}$ are basis functions and $\{\beta_{P,n,i,a,k,c,\ell_1,\ell_2}^f\}$, $\{\beta_{P,n,j,b,k,c,\ell_1,\ell_2}^g\}$ are coefficients which we can estimate with OLS. The numbers of basis functions for time and the covariate values—denoted by \tilde{c}_n and \tilde{d}_n , respectively—are chosen to increase with the sample size n at some rate. To fix ideas, we will use Legendre polynomials as the basis functions for both the theoretical analysis in this section and the numerical simulations in Section 5. Specifically, for each $\ell_1 \in [\tilde{c}_n]$ and $\ell_2 \in [\tilde{d}_n]$, let the basis functions for time $\{\phi_{\ell_1}(u)\}$ and the covariate values $\{\varphi_{\ell_2}(z)\}$ be *mapped* Legendre polynomials as in Example C.2 and Section 3.1.1 of Ding and Zhou [DZ21]. It is straightforward to replace the Legendre polynomials used in our theoretical analysis and simulations with trigonometric polynomials, wavelets, or other Jacobi polynomials.

Next, we introduce the sieve estimators for the time-varying regression functions. Although we do not discuss this topic in detail here, we point interested readers to further discussions of asymptotically optimal linear forecasting for locally stationary processes [DZ23; KR24; CZ25]. We expect similar results for asymptotically optimal nonlinear forecasting to be developed over the next few years.

Remark 4.1 (Sequential sieve estimation). *Our formulation of the sieve estimator from Ding and Zhou [DZ21] accommodates sequential estimation, in the sense that the predictors for rescaled time t/n are only constructed using the information up to rescaled time t/n . We emphasize that sequential estimation is not required for all settings, particularly when certain exogeneity conditions hold. The need for sequential estimation in some settings is due to the martingale difference sequence condition imposed on the error processes in Assumption 3.4 (c.f. Assumption 4.4), which becomes relevant when using our test for variable selection for forecasting and causal inference for time-lagged effects. Note that the same convergence rates are attained whether or not sequential estimation is used, due to the infill asymptotic framework of locally stationary processes. That is, because more observations for each local structure become available as n grows.*

Recall the following notation from Section 3.2. Let $\mathfrak{D}_{t,n,i,a}^f$, $\mathfrak{D}_{t,n,j,b}^g$ be the datasets used to form the estimators $\hat{f}_{t,n,i,a}(t/n, \cdot)$, $\hat{g}_{t,n,j,b}(t/n, \cdot)$ of the time-varying regression functions at rescaled time $t/n \in \mathcal{U}_n$, let $\mathcal{H}_{t,a}^f$, $\mathcal{H}_{t,b}^g$ be the corresponding input sequences, and let $\mathcal{T}_{t,n,i,a}^f$, $\mathcal{T}_{t,n,j,b}^g$ be the corresponding sets of times with $T_{t,n,i,a}^f = |\mathcal{T}_{t,n,i,a}^f|$, $T_{t,n,j,b}^g = |\mathcal{T}_{t,n,j,b}^g|$. Note that each of the estimators $\hat{f}_{t,n,i,a,k,c}(t/n, \cdot)$, $\hat{g}_{t,n,j,b,k,c}(t/n, \cdot)$ of the corresponding time-varying partial response functions at rescaled time $t/n \in \mathcal{U}_n$ may have different numbers of basis functions. Without confusion, we will write the numbers of basis functions as \tilde{c}_n and \tilde{d}_n instead of $\tilde{c}_{t,n,i,a,k,c}^f$, $\tilde{c}_{t,n,j,b,k,c}^g$ and $\tilde{d}_{t,n,i,a,k,c}^f$, $\tilde{d}_{t,n,j,b,k,c}^g$ to simplify the presentation below.

For some fixed $n \in \mathbb{N}$, $t \in \mathcal{T}_n$, and $(i, j, a, b) \in \mathcal{D}_n$, denote the design matrices by $\bar{\mathbf{Z}}_{t,n,i,a} \in \mathbb{R}^{T_{t,n,i,a}^f \times d_Z \tilde{c}_n \tilde{d}_n}$ and $\bar{\mathbf{Z}}_{t,n,j,b} \in \mathbb{R}^{T_{t,n,j,b}^g \times d_Z \tilde{c}_n \tilde{d}_n}$. The (s, p) -th entries of $\bar{\mathbf{Z}}_{t,n,i,a}$ and $\bar{\mathbf{Z}}_{t,n,j,b}$ are

$$\begin{aligned} \bar{\mathbf{Z}}_{t,n,i,a}^{(s,p)} &= \phi_{\ell_{1,p}}(t_s/n) \varphi_{\ell_{2,p}}(Z_{t_s,n,k_p,c_p}), \\ \bar{\mathbf{Z}}_{t,n,j,b}^{(s,p)} &= \phi_{\ell_{1,p}}(t_s/n) \varphi_{\ell_{2,p}}(Z_{t_s,n,k_p,c_p}), \end{aligned}$$

where we use mappings for the rows $s \mapsto t_s \in \mathcal{T}_{t,n,i,a}^f$ and $s \mapsto t_s \in \mathcal{T}_{t,n,j,b}^g$ which maintain the sequential order of time (i.e. $t_{s_1} < t_{s_2}$ if $s_1 < s_2$), and some mappings for the columns $p \mapsto (k_p, c_p, \ell_{1,p}, \ell_{2,p})$ which determine orderings for the dimension/time-offset/basis-index combinations, where $k_p \in [d_Z]$, $c_p \in C_{k_p}$, $\ell_{1,p} \in [\tilde{c}_n]$, $\ell_{2,p} \in [\tilde{d}_n]$. That is, each row corresponds to one time and each column corresponds to one dimension/time-offset combination with a particular basis-index combination. For

each $n \in \mathbb{N}$, $P \in \mathcal{P}_n$, $i \in [d_X]$, $a \in A_i$, $j \in [d_Y]$, $b \in B_j$ the (time-invariant) coefficient vectors

$$\begin{aligned}\beta_{P,n,i,a}^f &= (\beta_{P,n,i,a,k,c,\ell_1,\ell_2}^f)_{k,c,\ell_1,\ell_2}^\top \in \mathbb{R}^{d_Z \tilde{c}_n \tilde{d}_n}, \\ \beta_{P,n,j,b}^g &= (\beta_{P,n,j,b,k,c,\ell_1,\ell_2}^g)_{k,c,\ell_1,\ell_2}^\top \in \mathbb{R}^{d_Z \tilde{c}_n \tilde{d}_n},\end{aligned}$$

have the following OLS estimators

$$\begin{aligned}\hat{\beta}_{t,n,i,a}^f &= (\bar{\mathbf{Z}}_{t,n,i,a}^\top \bar{\mathbf{Z}}_{t,n,i,a})^{-1} \bar{\mathbf{Z}}_{t,n,i,a}^\top \bar{\mathbf{X}}_{t,n,i,a} = (\hat{\beta}_{t,n,i,a,k,c,\ell_1,\ell_2}^f)_{k,c,\ell_1,\ell_2}^\top \in \mathbb{R}^{d_Z \tilde{c}_n \tilde{d}_n}, \\ \hat{\beta}_{t,n,j,b}^g &= (\bar{\mathbf{Z}}_{t,n,j,b}^\top \bar{\mathbf{Z}}_{t,n,j,b})^{-1} \bar{\mathbf{Z}}_{t,n,j,b}^\top \bar{\mathbf{Y}}_{t,n,j,b} = (\hat{\beta}_{t,n,j,b,k,c,\ell_1,\ell_2}^g)_{k,c,\ell_1,\ell_2}^\top \in \mathbb{R}^{d_Z \tilde{c}_n \tilde{d}_n},\end{aligned}$$

where

$$\bar{\mathbf{X}}_{t,n,i,a} = (X_{t,n,i,a})_{t \in \mathcal{T}_{t,n,i,a}^f}^\top \in \mathbb{R}^{T_{t,n,i,a}^f}, \quad \bar{\mathbf{Y}}_{t,n,j,b} = (Y_{t,n,j,b})_{t \in \mathcal{T}_{t,n,j,b}^g}^\top \in \mathbb{R}^{T_{t,n,j,b}^g}.$$

Finally, the estimators of the time-varying regression functions $f_{P,n,i,a}(t/n, \cdot)$ and $g_{P,n,j,b}(t/n, \cdot)$ at rescaled time $t/n \in \mathcal{U}_n$ are given by

$$\begin{aligned}\hat{f}_{t,n,i,a}(t/n, \cdot) &= \sum_{k=1}^{d_Z} \sum_{c=1}^{C_k} \hat{f}_{t,n,i,a,k,c}(t/n, \cdot), \\ \hat{g}_{t,n,j,b}(t/n, \cdot) &= \sum_{k=1}^{d_Z} \sum_{c=1}^{C_k} \hat{g}_{t,n,j,b,k,c}(t/n, \cdot),\end{aligned}$$

where the estimators of the time-varying partial response functions $f_{P,n,i,a,k,c}(t/n, \cdot)$ and $g_{P,n,j,b,k,c}(t/n, \cdot)$ at rescaled time $t/n \in \mathcal{U}_n$ are given by

$$\begin{aligned}\hat{f}_{t,n,i,a,k,c}(t/n, \cdot) &= \sum_{\ell_1=1}^{\tilde{c}_n} \sum_{\ell_2=1}^{\tilde{d}_n} \hat{\beta}_{t,n,i,a,k,c,\ell_1,\ell_2}^f b_{\ell_1,\ell_2}(t/n, \cdot), \\ \hat{g}_{t,n,j,b,k,c}(t/n, \cdot) &= \sum_{\ell_1=1}^{\tilde{c}_n} \sum_{\ell_2=1}^{\tilde{d}_n} \hat{\beta}_{t,n,j,b,k,c,\ell_1,\ell_2}^g b_{\ell_1,\ell_2}(t/n, \cdot).\end{aligned}$$

Although we only discuss the sieve estimator here, we emphasize that any black-box time-varying regression estimator can be used with the dGCM test. For example, we can use time-varying regression estimators based on kernel smoothing [Vog12; ZW15; YN21; CSW22; DZ21]. To use kernel smoothing estimators for sequential estimation, we can use one-sided temporal kernels so that observations after rescaled time t/n receive a weight of zero. This is practically important because “local” nonparametric estimators are naturally far more computationally efficient for sequential estimation than “global” nonparametric estimators in the absence of efficient online estimation procedures for the latter.

4.4 Locally stationary error processes

We will now introduce the causal representations of the locally stationary error processes from Section 4.3. For each $a \in A$, $b \in B$, define the input sequences

$$\mathcal{H}_{t,a}^\varepsilon = (\eta_{t,a}^\varepsilon, \eta_{t,a-1}^\varepsilon, \dots), \quad \mathcal{H}_{t,b}^\xi = (\eta_{t,b}^\xi, \eta_{t,b-1}^\xi, \dots),$$

where $(\eta_{t,a}^\varepsilon, \eta_{t,b}^\xi)_{t \in \mathbb{Z}}$ is a sequence of iid random vectors. Denote the dimension of $\eta_{t,a}^\varepsilon = \eta_{t,a,n}^\varepsilon$ by $d_\varepsilon^\eta = d_{\varepsilon,n}^\eta$ and the dimension of $\eta_{t,b}^\xi = \eta_{t,b,n}^\xi$ by $d_\xi^\eta = d_{\xi,n}^\eta$, both of which can change with n . For the next assumption, let $\mathcal{H}_t^{\hat{f}} = (\mathcal{H}_{t,a}^{\hat{f}})_{a \in A}$, $\mathcal{H}_t^{\hat{g}} = (\mathcal{H}_{t,b}^{\hat{g}})_{b \in B}$ and $\mathcal{H}_t^{\hat{f}} = (\mathcal{H}_{t,a}^{\hat{f}}, \mathcal{H}_t^Z)$, $\mathcal{H}_t^{\hat{g}} = (\mathcal{H}_{t,b}^{\hat{g}}, \mathcal{H}_t^Z)$, where the input sequences $\mathcal{H}_{t,a}^{\hat{f}}$, $\mathcal{H}_{t,b}^{\hat{g}}$ were defined in Section 4.3 and \mathcal{H}_t^Z was defined in Section 4.2.

Assumption 4.4 (Causal representations of the error processes). *Assume that for each $n \in \mathbb{N}$, $P \in \mathcal{P}_n$, $(i, j, a, b) \in \mathcal{D}_n$, $t \in \mathcal{T}_n$, the error processes from Section 4.3 can be represented as*

$$\varepsilon_{P,t,n,i,a} = \tilde{G}_{P,n,i,a}^\varepsilon(t/n, \mathcal{H}_{t,a}^\varepsilon), \quad \xi_{P,t,n,j,b} = \tilde{G}_{P,n,j,b}^\xi(t/n, \mathcal{H}_{t,b}^\xi),$$

with $\mathbb{E}_P(\varepsilon_{P,t,n,i,a}|\mathcal{H}_t^{\hat{g}}) = 0$ and $\mathbb{E}_P(\xi_{P,t,n,j,b}|\mathcal{H}_t^{\hat{f}}) = 0$, where the input sequences $\mathcal{H}_t^{\hat{g}}$, $\mathcal{H}_t^{\hat{f}}$ were defined above. The causal representations

$$\tilde{\varepsilon}_{P,t,n,i,a}(u) = \tilde{G}_{P,n,i,a}^{\varepsilon}(u, \mathcal{H}_{t,a}^{\varepsilon}), \quad \tilde{\xi}_{P,t,n,j,b}(u) = \tilde{G}_{P,n,j,b}^{\xi}(u, \mathcal{H}_{t,b}^{\xi}),$$

are defined at all $u \in \mathcal{U}_n$, so that we have $\varepsilon_{P,t,n,i,a} = \tilde{\varepsilon}_{P,t,n,i,a}(t/n)$, $\xi_{P,t,n,j,b} = \tilde{\xi}_{P,t,n,j,b}(t/n)$. $\tilde{G}_{P,n,i,a}^{\varepsilon}(u, \cdot)$ and $\tilde{G}_{P,n,j,b}^{\xi}(u, \cdot)$ are measurable functions from $(\mathbb{R}^{d_{\varepsilon}^n})^{\infty}$ and $(\mathbb{R}^{d_{\xi}^n})^{\infty}$, respectively, to \mathbb{R} —where we endow $(\mathbb{R}^{d_{\varepsilon}^n})^{\infty}$ and $(\mathbb{R}^{d_{\xi}^n})^{\infty}$ with the σ -algebra generated by all finite projections—so that $\tilde{G}_{P,n,i,a}^{\varepsilon}(u, \mathcal{H}_{s,a}^{\varepsilon})$, $\tilde{G}_{P,n,j,b}^{\xi}(u, \mathcal{H}_{s,b}^{\xi})$ are well-defined random variables for each $s \in \mathbb{Z}$ and $(\tilde{G}_{P,n,i,a}^{\varepsilon}(u, \mathcal{H}_{s,a}^{\varepsilon}))_{s \in \mathbb{Z}}$, $(\tilde{G}_{P,n,j,b}^{\xi}(u, \mathcal{H}_{s,b}^{\xi}))_{s \in \mathbb{Z}}$ are stationary ergodic processes.

As in Section 3.3, we have not defined the input sequences for the error processes separately for each dimension, because without loss of generality we may define the measurable functions $\tilde{G}_{P,n,i,a}^{\varepsilon}(u, \cdot)$, $\tilde{G}_{P,n,j,b}^{\xi}(u, \cdot)$ and inputs $\eta_{t,a}^{\varepsilon}, \eta_{t,b}^{\xi}$ so that each dimension of the error processes has idiosyncratic inputs.

Using the causal representations of the univariate error processes, we have the following causal representations of the vector-valued error processes

$$\begin{aligned} \tilde{\varepsilon}_{P,t,n}(u) &= \tilde{\mathbf{G}}_{P,n}^{\varepsilon}(u, \mathcal{H}_t^{\varepsilon}) = (\tilde{G}_{P,n,i,a}^{\varepsilon}(u, \mathcal{H}_{t,a}^{\varepsilon}))_{i \in [d_X], a \in A_i}, \\ \tilde{\xi}_{P,t,n}(u) &= \tilde{\mathbf{G}}_{P,n}^{\xi}(u, \mathcal{H}_t^{\xi}) = (\tilde{G}_{P,n,j,b}^{\xi}(u, \mathcal{H}_{t,b}^{\xi}))_{j \in [d_Y], b \in B_j}, \end{aligned}$$

so that we have $\varepsilon_{P,t,n} = \tilde{\varepsilon}_{P,t,n}(t/n)$, $\xi_{P,t,n} = \tilde{\xi}_{P,t,n}(t/n)$, where $\mathcal{H}_t^{\varepsilon} = (\eta_t^{\varepsilon}, \eta_{t-1}^{\varepsilon}, \dots)$, $\mathcal{H}_t^{\xi} = (\eta_t^{\xi}, \eta_{t-1}^{\xi}, \dots)$ with $\eta_t^{\varepsilon} = (\eta_{t,a}^{\varepsilon})_{a \in A}$, $\eta_t^{\xi} = (\eta_{t,b}^{\xi})_{b \in B}$ for each $t \in \mathbb{Z}$. Similarly, for each $m = (i, j, a, b) \in \mathcal{D}_n$ the error products can be represented as

$$\tilde{R}_{P,t,n,m}(u) = \tilde{G}_{P,n,m}^R(u, \mathcal{H}_{t,m}^R) = \tilde{G}_{P,n,i,a}^{\varepsilon}(u, \mathcal{H}_{t,a}^{\varepsilon}) \tilde{G}_{P,n,j,b}^{\xi}(u, \mathcal{H}_{t,b}^{\xi}),$$

so that we have $R_{P,t,n,m} = \tilde{R}_{P,t,n,m}(t/n)$, where $\mathcal{H}_{t,m}^R = (\eta_{t,m}^R, \eta_{t-1,m}^R, \dots)$ with $\eta_{t,m}^R = (\eta_{t,a}^{\varepsilon}, \eta_{t,b}^{\xi})^{\top}$ for each $t \in \mathbb{Z}$. Also, we have the following causal representation of the nonstationary \mathbb{R}^{D_n} -valued process of all the products of errors

$$\tilde{\mathbf{R}}_{P,n,t}(u) = \tilde{\mathbf{G}}_{P,n}^R(u, \mathcal{H}_t^R) = (\tilde{G}_{P,n,m}^R(u, \mathcal{H}_{t,m}^R))_{m \in \mathcal{D}_n},$$

so that we have $\mathbf{R}_{P,t,n} = \tilde{\mathbf{R}}_{P,n,t}(t/n)$, where $\mathcal{H}_t^R = (\eta_t^R, \eta_{t-1}^R, \dots)$ and $\eta_t^R = (\eta_t^{\varepsilon}, \eta_t^{\xi})^{\top}$ for each $t \in \mathbb{Z}$. We emphasize that for a fixed $P \in \mathcal{P}_n$, $u \in \mathcal{U}_n$, and $n \in \mathbb{N}$, we have that $\tilde{\mathbf{G}}_{P,n}^R(u, \mathcal{H}_s^R)$ is a well-defined random vector for each $s \in \mathbb{Z}$ and $(\tilde{\mathbf{G}}_{P,n}^R(u, \mathcal{H}_s^R))_{s \in \mathbb{Z}}$ is a stationary ergodic \mathbb{R}^{D_n} -valued process.

4.5 Assumptions on dependence and nonstationarity

We impose assumptions on the rate of decay in temporal dependence and the degree of nonstationarity of the observed processes and error processes. We emphasize that the assumptions here are strictly stronger than those in Section 3.4. We impose these stronger assumptions to guarantee that the sieve time-varying nonlinear regression estimator achieves the convergence rates required by Theorem 3.1. Note that the assumptions here require that the nonstationary processes evolve “smoothly” in time, which excludes nonstationary processes with abrupt changes. We do this mainly to simplify the presentation, and we discuss extensions to nonstationary processes with both smooth and abrupt changes in Section C.7.

Denote the set of well-defined tuples of observed processes, dimensions, and time-offsets by

$$\mathbb{W} = \{(X, i, a) : i \in [d_X], a \in A_i\} \cup \{(Y, j, b) : j \in [d_Y], b \in B_j\} \cup \{(Z, k, c) : k \in [d_Z], c \in C_k\},$$

so that we may conveniently refer to such well-defined combinations by $(W, l, d) \in \mathbb{W}$. Also, denote the set of well-defined tuples of error processes, dimensions, and time-offsets by

$$\mathbb{E} = \{(\varepsilon, i, a) : i \in [d_X], a \in A_i\} \cup \{(\xi, j, b) : j \in [d_Y], b \in B_j\},$$

so that we may write $(e, l, d) \in \mathbb{E}$ to refer to any such combination.

Again, we quantify temporal dependence via the functional dependence measure of Wu [Wu05]. Let $(\tilde{\eta}_t^X, \tilde{\eta}_t^Y, \tilde{\eta}_t^Z)_{t \in \mathbb{Z}}$ be an iid copy of $(\eta_t^X, \eta_t^Y, \eta_t^Z)_{t \in \mathbb{Z}}$. Going forward, the inputs with the tilde are from $(\tilde{\eta}_t^X, \tilde{\eta}_t^Y, \tilde{\eta}_t^Z)_{t \in \mathbb{Z}}$. For any tuple $(W, l, d) \in \mathbb{W}$ corresponding to a well-defined combination of an observed process, dimension, and time-offset, define

$$\tilde{\mathcal{H}}_{t,d,h}^W = (\eta_{t+d}^W, \dots, \eta_{t-h+1+d}^W, \tilde{\eta}_{t-h+d}^W, \eta_{t-h-1+d}^W, \dots)$$

to be $\mathcal{H}_{t,d}^W$ with the input η_{t-h+d}^W replaced with the iid copy $\tilde{\eta}_{t-h+d}^W$. For example, for $i \in [d_X]$, $a \in A_i$, we have that $\tilde{\eta}_{t-h+a}^X$ is the copy of the input η_{t-h+a}^X in the input sequence $\mathcal{H}_{t,a}^X$ used in the causal representation of $X_{t,n,i,a}$. Analogously, for $\mathbf{W} \in \{\mathbf{X}, \mathbf{Y}, \mathbf{Z}\}$ define $\tilde{\mathcal{H}}_{t,h}^{\mathbf{W}}$ as $\mathcal{H}_t^{\mathbf{W}}$ with the input $\eta_{t-h}^{\mathbf{W}}$ replaced with the iid copy $\tilde{\eta}_{t-h}^{\mathbf{W}}$ as in Section 4.2.

For any tuple $(e, l, d) \in \mathbb{E}$ corresponding to a well-defined combination of an error process, dimension, and time-offset, define

$$\tilde{\mathcal{H}}_{t,d,h}^e = (\eta_{t,d}^e, \dots, \eta_{t-h+1,d}^e, \tilde{\eta}_{t-h,d}^e, \eta_{t-h-1,d}^e, \dots)$$

to be $\mathcal{H}_{t,d}^e$ with the input $\eta_{t-h,d}^e$ replaced with the iid copy $\tilde{\eta}_{t-h,d}^e$. Analogously, for $e \in \{\varepsilon, \xi\}$ define $\tilde{\mathcal{H}}_{t,h}^e$ as \mathcal{H}_t^e with the input η_{t-h}^e replaced with the iid copy $\tilde{\eta}_{t-h}^e$ as in Section 4.4. Also, for the product of errors define $\tilde{\mathcal{H}}_{t,m,h}^R$ as $\mathcal{H}_{t,m}^R$ with the input $\eta_{t-h,m}^R$ replaced with the iid copy $\tilde{\eta}_{t-h,m}^R$ for $m = (i, j, a, b) \in \mathcal{D}_n$. Analogously, define $\tilde{\mathcal{H}}_{t,h}^R$ as \mathcal{H}_t^R with the input η_{t-h}^R replaced with the iid copy $\tilde{\eta}_{t-h}^R$ as in Section 4.4. Now, we define the functional dependence measures of the processes.

Definition 4.1 (Functional dependence measures). *We define the following measures of temporal dependence for each $n \in \mathbb{N}$, $P \in \mathcal{P}_n$, $u \in \mathcal{U}_n$, $t \in \mathcal{T}_n$. First, define the functional dependence measures of the observed processes $\tilde{G}_{n,l,d}^W(u, \mathcal{H}_{t,d}^W)$ for each $(W, l, d) \in \mathbb{W}$ with $h \in \mathbb{N}_0$, and some $q \geq 1$ as*

$$\theta_{P,u,t,n,l,d}^W(h, q) = [\mathbb{E}_P(|\tilde{G}_{n,l,d}^W(u, \mathcal{H}_{t,d}^W) - \tilde{G}_{n,l,d}^W(u, \tilde{\mathcal{H}}_{t,d,h}^W)|^q)]^{1/q},$$

and for the vector-valued process $\tilde{\mathbf{G}}_n^W(u, \mathcal{H}_t^W)$ for each $\mathbf{W} \in \{\mathbf{X}, \mathbf{Y}, \mathbf{Z}\}$ with $h \in \mathbb{N}_0$, and some $q \geq 1$, $r \geq 1$ as

$$\theta_{P,u,t,n}^{\mathbf{W}}(h, q, r) = [\mathbb{E}_P(\|\tilde{\mathbf{G}}_n^W(u, \mathcal{H}_t^W) - \tilde{\mathbf{G}}_n^W(u, \tilde{\mathcal{H}}_{t,h}^W)\|_r^q)]^{1/q}.$$

Second, define the L^∞ versions of the functional dependence measures of the error processes $\tilde{G}_{P,n,l,d}^e(u, \mathcal{H}_{t,d}^e)$ for each $(e, l, d) \in \mathbb{E}$ with $h \in \mathbb{N}_0$ as

$$\theta_{P,u,t,n,l,d}^{e,\infty}(h) = \inf\{K \geq 0 : \mathbb{P}_P(|\tilde{G}_{P,n,l,d}^e(u, \mathcal{H}_{t,d}^e) - \tilde{G}_{P,n,l,d}^e(u, \tilde{\mathcal{H}}_{t,d,h}^e)| > K) = 0\},$$

and for the vector-valued process $\tilde{\mathbf{G}}_{P,n}^e(u, \mathcal{H}_t^e)$ for each $e \in \{\varepsilon, \xi\}$ with $h \in \mathbb{N}_0$, and some $r \geq 1$ as

$$\theta_{P,u,t,n}^{e,\infty}(h, r) = \inf\{K \geq 0 : \mathbb{P}_P(\|\tilde{\mathbf{G}}_{P,n}^e(u, \mathcal{H}_t^e) - \tilde{\mathbf{G}}_{P,n}^e(u, \tilde{\mathcal{H}}_{t,h}^e)\|_r > K) = 0\}.$$

Third, define the functional dependence measures of the processes of error products $\tilde{G}_{P,n,m}^R(u, \mathcal{H}_{t,m}^R)$ for each $m = (i, j, a, b) \in \mathcal{D}_n$ with $h \in \mathbb{N}_0$, and some $q \geq 1$ as

$$\theta_{P,u,t,n,m}^R(h, q) = [\mathbb{E}_P(|\tilde{G}_{P,n,m}^R(u, \mathcal{H}_{t,m}^R) - \tilde{G}_{P,n,m}^R(u, \tilde{\mathcal{H}}_{t,m,h}^R)|^q)]^{1/q},$$

and for the vector-valued process $\tilde{\mathbf{G}}_{P,n}^R(u, \mathcal{H}_t^R)$ with $h \in \mathbb{N}_0$, and some $q \geq 1$, $r \geq 1$ as

$$\theta_{P,u,t,n}^R(h, q, r) = [\mathbb{E}_P(\|\tilde{\mathbf{G}}_{P,n}^R(u, \mathcal{H}_t^R) - \tilde{\mathbf{G}}_{P,n}^R(u, \tilde{\mathcal{H}}_{t,h}^R)\|_r^q)]^{1/q}.$$

Next, we introduce an assumption imposing a uniform polynomial decay of the temporal dependence. Note that we will often write the time as 0 when the time of the input sequence does not matter because of stationarity. For some sequence of collections of distributions $(\mathcal{P}_n)_{n \in \mathbb{N}}$, we make the following assumption.

Assumption 4.5 (Distribution-uniform decay of temporal dependence). *Assume that there exist $\bar{\Theta} > 0$, $\bar{\beta} > 2$, $\bar{q} > 4$, such that for all $n \in \mathbb{N}$, $u \in \mathcal{U}_n$, and observed processes $(W, l, d) \in \mathbb{W}$, it holds that*

$$\sup_{P \in \mathcal{P}_n} [\mathbb{E}_P(|\tilde{G}_{n,l,d}^W(u, \mathcal{H}_{0,d}^W)|^{\bar{q}})]^{1/\bar{q}} \leq \bar{\Theta}, \quad \sup_{P \in \mathcal{P}_n} \theta_{P,u,0,n,l,d}^W(h, \bar{q}) \leq \bar{\Theta} \cdot (h \vee 1)^{-\bar{\beta}}, \quad h \geq 0.$$

Also, assume that there exist $\bar{\Theta}^\infty > 0$, $\bar{\beta}^\infty > 2$, such that for all $n \in \mathbb{N}$, $u \in \mathcal{U}_n$, and error processes $(e, l, d) \in \mathbb{E}$, it holds that

$$\sup_{P \in \mathcal{P}_n} \|G_{P,n,l,d}^e(u, \mathcal{H}_{0,d}^e)\|_{L^\infty(P)} \leq \bar{\Theta}^\infty, \quad \sup_{P \in \mathcal{P}_n} \theta_{P,u,0,n,l,d}^{e,\infty}(h) \leq \bar{\Theta}^\infty \cdot (h \vee 1)^{-\bar{\beta}^\infty}, \quad h \geq 0.$$

For additional control in terms of the product of errors alone, also assume that there exist $\bar{\Theta}^R > 0$, $\bar{\beta}^R > 3$, $\bar{q}^R > 4$, such that for all $n \in \mathbb{N}$, $u \in \mathcal{U}_n$, $m = (i, j, a, b) \in \mathcal{D}_n$, it holds that

$$\sup_{P \in \mathcal{P}_n} [\mathbb{E}_P(|\tilde{G}_{P,n,m}^R(u, \mathcal{H}_{0,m}^R)|^{\bar{q}^R})]^{1/\bar{q}^R} \leq \bar{\Theta}^R, \quad \sup_{P \in \mathcal{P}_n} \theta_{P,u,0,n,m}^R(h, \bar{q}^R) \leq \bar{\Theta}^R \cdot (h \vee 1)^{-\bar{\beta}^R}, \quad h \geq 0.$$

In view of Assumption 4.5, we have the following bounds on the functional dependence measures of the corresponding vector-valued processes for each $n \in \mathbb{N}$, $u \in \mathcal{U}_n$ by Jensen's inequality. For the vector-valued process of error products, we have

$$\sup_{P \in \mathcal{P}_n} [\mathbb{E}_P(\|\tilde{G}_{P,n}^R(u, \mathcal{H}_0^R)\|_2^{\bar{q}^R})]^{1/\bar{q}^R} \leq D_n^{\frac{1}{2}} \bar{\Theta}^R, \quad \sup_{P \in \mathcal{P}_n} \theta_{P,u,0,n}^R(h, \bar{q}^R, 2) \leq D_n^{\frac{1}{2}} \bar{\Theta}^R \cdot (h \vee 1)^{-\bar{\beta}^R}, \quad h \geq 0.$$

Also, for each of the vector-valued observed processes $\mathbf{W} \in (\mathbf{X}, \mathbf{Y}, \mathbf{Z})$, we have

$$\sup_{P \in \mathcal{P}_n} [\mathbb{E}_P(\|\tilde{\mathbf{G}}_n^{\mathbf{W}}(u, \mathcal{H}_0^{\mathbf{W}})\|_2^{\bar{q}})]^{1/\bar{q}} \leq D_n^{\frac{1}{2}} \bar{\Theta}, \quad \sup_{P \in \mathcal{P}_n} \theta_{P,u,0,n}^{\mathbf{W}}(h, \bar{q}, 2) \leq D_n^{\frac{1}{2}} \bar{\Theta} \cdot (h \vee 1)^{-\bar{\beta}}, \quad h \geq 0.$$

Lastly, for each of the vector-valued error processes $\mathbf{e} \in (\boldsymbol{\varepsilon}, \boldsymbol{\xi})$, we have

$$\sup_{P \in \mathcal{P}_n} \left\| \|\tilde{\mathbf{G}}_{P,n}^{\mathbf{e}}(u, \mathcal{H}_0^{\mathbf{e}})\|_2 \right\|_{L^\infty(P)} \leq D_n^{\frac{1}{2}} \bar{\Theta}^\infty, \quad \sup_{P \in \mathcal{P}_n} \theta_{P,u,0,n}^{\mathbf{e},\infty}(h, 2) \leq D_n^{\frac{1}{2}} \bar{\Theta}^\infty \cdot (h \vee 1)^{-\bar{\beta}^\infty}, \quad h \geq 0.$$

Next, we discuss an additional regularity condition required by the sieve estimator that is analogous to Lemma 3.1 in Ding and Zhou [DZ21]. Recall the set of basis functions $\{\varphi_{\ell_2}(z)\}$ from Section 4.3. For each $n \in \mathbb{N}$, $P \in \mathcal{P}_n$, $t \in \mathcal{T}_n$, $i \in [d_X]$, $a \in A_i$, $j \in [d_Y]$, $b \in B_j$ let

$$\begin{aligned} \mathbf{w}_{t,n}^{\varphi(Z)} &= (\varphi_{\ell_2}(Z_{t,n,k,c}))_{k \in [d_Z], c \in C_k, 1 \leq \ell_2 \leq \bar{d}_n}, \\ \mathbf{w}_{P,t,n,i,a}^{\varphi(Z),\varepsilon} &= (\varphi_{\ell_2}(Z_{t,n,k,c}) \varepsilon_{P,t,n,i,a})_{k \in [d_Z], c \in C_k, 1 \leq \ell_2 \leq \bar{d}_n}, \\ \mathbf{w}_{P,t,n,j,b}^{\varphi(Z),\xi} &= (\varphi_{\ell_2}(Z_{t,n,k,c}) \xi_{P,t,n,j,b})_{k \in [d_Z], c \in C_k, 1 \leq \ell_2 \leq \bar{d}_n}. \end{aligned}$$

As in Section 3.2 in Ding and Zhou [DZ21], the $\mathbb{R}^{d_Z \bar{d}_n}$ -valued processes $\mathbf{w}_{t,n}^{\varphi(Z)}$, $\mathbf{w}_{P,t,n,i,a}^{\varphi(Z),\varepsilon}$, and $\mathbf{w}_{P,t,n,j,b}^{\varphi(Z),\xi}$ all have causal representations

$$\begin{aligned} \mathbf{w}_{t,n}^{\varphi(Z)} &= \tilde{\mathbf{G}}_n^{\varphi(Z)}(t/n, \mathcal{H}_t^{\varphi(Z)}), \\ \mathbf{w}_{P,t,n,i,a}^{\varphi(Z),\varepsilon} &= \tilde{\mathbf{G}}_{P,n,i,a}^{\varphi(Z),\varepsilon}(t/n, \mathcal{H}_{t,a}^{\varphi(Z),\varepsilon}), \\ \mathbf{w}_{P,t,n,j,b}^{\varphi(Z),\xi} &= \tilde{\mathbf{G}}_{P,n,j,b}^{\varphi(Z),\xi}(t/n, \mathcal{H}_{t,b}^{\varphi(Z),\xi}), \end{aligned}$$

where

$$\begin{aligned} \mathcal{H}_t^{\varphi(Z)} &= (\eta_t^{\varphi(Z)}, \eta_{t-1}^{\varphi(Z)}, \dots), \\ \mathcal{H}_{t,a}^{\varphi(Z),\varepsilon} &= (\eta_{t,a}^{\varphi(Z),\varepsilon}, \eta_{t-1,a}^{\varphi(Z),\varepsilon}, \dots), \\ \mathcal{H}_{t,b}^{\varphi(Z),\xi} &= (\eta_{t,b}^{\varphi(Z),\xi}, \eta_{t-1,b}^{\varphi(Z),\xi}, \dots), \end{aligned}$$

with $\eta_t^{\varphi(Z)} = \eta_{t+c_{\max}}^Z$, $\eta_{t,a}^{\varphi(Z),\varepsilon} = (\eta_{t+c_{\max}}^Z, \eta_{t+a}^X)^\top$, and $\eta_{t,b}^{\varphi(Z),\xi} = (\eta_{t+c_{\max}}^Z, \eta_{t+b}^Y)^\top$. Define the functional dependence measures of the vector-valued processes $\mathbf{w}_{t,n}^{\varphi(Z)}$, $\mathbf{w}_{P,t,n,i,a}^{\varphi(Z),\varepsilon}$, $\mathbf{w}_{P,t,n,j,b}^{\varphi(Z),\xi}$ by

$$\begin{aligned} \theta_{P,u,t,n}^{\varphi(Z)}(h, q, 2) &= [\mathbb{E}_P(\|\tilde{\mathbf{G}}_n^{\varphi(Z)}(u, \mathcal{H}_t^{\varphi(Z)}) - \tilde{\mathbf{G}}_n^{\varphi(Z)}(u, \tilde{\mathcal{H}}_{t,h}^{\varphi(Z)})\|_2^q)]^{1/q}, \\ \theta_{P,u,t,n,i,a}^{\varphi(Z),\varepsilon}(h, q, 2) &= [\mathbb{E}_P(\|\tilde{\mathbf{G}}_{P,n,i,a}^{\varphi(Z),\varepsilon}(u, \mathcal{H}_{t,a}^{\varphi(Z),\varepsilon}) - \tilde{\mathbf{G}}_{P,n,i,a}^{\varphi(Z),\varepsilon}(u, \tilde{\mathcal{H}}_{t,a,h}^{\varphi(Z),\varepsilon})\|_2^q)]^{1/q}, \\ \theta_{P,u,t,n,j,b}^{\varphi(Z),\xi}(h, q, 2) &= [\mathbb{E}_P(\|\tilde{\mathbf{G}}_{P,n,j,b}^{\varphi(Z),\xi}(u, \mathcal{H}_{t,b}^{\varphi(Z),\xi}) - \tilde{\mathbf{G}}_{P,n,j,b}^{\varphi(Z),\xi}(u, \tilde{\mathcal{H}}_{t,b,h}^{\varphi(Z),\xi})\|_2^q)]^{1/q}. \end{aligned}$$

Recall $\bar{\Theta}$, $\bar{\Theta}^\infty$, $\bar{\beta}$, $\bar{\beta}^\infty$, and \bar{q} from Assumption 4.5. Using the same arguments from Lemma 3.1 from Ding and Zhou [DZ21], for all $n \in \mathbb{N}$, $u \in \mathcal{U}_n$, the vector-valued processes $\mathbf{w}_{P,t,n,i,a}^{\varphi(Z),\varepsilon}$, $\mathbf{w}_{P,t,n,j,b}^{\varphi(Z),\xi}$ satisfy

$$\sup_{P \in \mathcal{P}_n} [\mathbb{E}_P(\|\tilde{\mathbf{G}}_{P,n,i,a}^{\varphi(Z),\varepsilon}(u, \mathcal{H}_{t,a}^{\varphi(Z),\varepsilon})\|_2^{\bar{q}})]^{1/\bar{q}} \leq D_n^{\frac{1}{2}} \tilde{\Theta}, \quad \sup_{P \in \mathcal{P}_n} \theta_{P,u,t,n,i,a}^{\varphi(Z),\varepsilon}(h, \tilde{q}, 2) \leq D_n^{\frac{1}{2}} \tilde{\Theta} \cdot (h \vee 1)^{-\bar{\beta}},$$

$$\sup_{P \in \mathcal{P}_n} [\mathbb{E}_P(\|\tilde{\mathbf{G}}_{P,n,j,b}^{\varphi(Z),\xi}(u, \mathcal{H}_{t,b}^{\varphi(Z),\xi})\|_2^{\bar{q}})]^{1/\bar{q}} \leq D_n^{\frac{1}{2}} \tilde{\Theta}, \quad \sup_{P \in \mathcal{P}_n} \theta_{P,u,t,n,j,b}^{\varphi(Z),\xi}(h, \tilde{q}, 2) \leq D_n^{\frac{1}{2}} \tilde{\Theta} \cdot (h \vee 1)^{-\bar{\beta}},$$

for $h \geq 0$, where $\tilde{q} = \bar{q} > 4$ with $\bar{\beta} = \min(\bar{\beta}, \bar{\beta}^\infty) > 2$ and $\tilde{\Theta} = 2K_1(\max(\bar{\Theta}^\infty, \bar{\Theta}))^2 > 0$ where the constant factor $K_1 > 0$ is due to the basis functions. Similarly, for all $n \in \mathbb{N}$, $u \in \mathcal{U}_n$, the vector-valued process $\mathbf{w}_{t,n}^{\varphi(Z)}$ satisfies

$$\sup_{P \in \mathcal{P}_n} [\mathbb{E}_P(\|\tilde{\mathbf{G}}_n^{\varphi(Z)}(u, \mathcal{H}_t^{\varphi(Z)})\|_2^{\bar{q}})]^{1/\bar{q}} \leq D_n^{\frac{1}{2}} \bar{\Theta} K_2, \quad \sup_{P \in \mathcal{P}_n} \theta_{P,u,t,n}^{\varphi(Z)}(h, \bar{q}, 2) \leq D_n^{\frac{1}{2}} \bar{\Theta} K_2 \cdot (h \vee 1)^{-\bar{\beta}},$$

for $h \geq 0$, where the constant factor $K_2 > 0$ is due to the basis functions.

For Theorem 3.1, we only require that the total variation of the causal mechanism of the process of error products can be bounded distribution-uniformly. However, the sieve estimator requires the stronger assumption that the causal mechanisms of the observed processes and error processes are stochastic Lipschitz functions of rescaled time. We impose the following regularity conditions to control the nonstationarity uniformly over a sequence of collections of distributions $(\mathcal{P}_n)_{n \in \mathbb{N}}$.

Assumption 4.6 (Distribution-uniform stochastic Lipschitz condition for nonstationarity). *For each $n \in \mathbb{N}$, $(W, l, d) \in \mathbb{W}$, $(e, l, d) \in \mathbb{E}$, and $t \in \mathbb{Z}$, we assume that $\tilde{G}_{n,l,d}^W(\cdot, \mathcal{H}_{t,d}^W)$ and $\tilde{G}_{P,n,l,d}^e(\cdot, \mathcal{H}_{t,d}^e)$ are stochastic Lipschitz functions of rescaled time $u \in \mathcal{U}_n$. Recall $\bar{\Theta} > 0$, $\bar{q} > 4$ from Assumption 4.5. Assume that there exists a constant $\bar{L} > 0$ such that for all $n \in \mathbb{N}$, $u, v \in \mathcal{U}_n$, $(W, l, d) \in \mathbb{W}$, $(e, l, d) \in \mathbb{E}$, it holds that*

$$\sup_{P \in \mathcal{P}_n} [\mathbb{E}_P(|\tilde{G}_{n,l,d}^W(u, \mathcal{H}_{0,d}^W) - \tilde{G}_{n,l,d}^W(v, \mathcal{H}_{0,d}^W)|^{\bar{q}})]^{1/\bar{q}} \leq \bar{L} \bar{\Theta} |u - v|,$$

$$\sup_{P \in \mathcal{P}_n} [\mathbb{E}_P(|\tilde{G}_{P,n,l,d}^e(u, \mathcal{H}_{0,d}^e) - \tilde{G}_{P,n,l,d}^e(v, \mathcal{H}_{0,d}^e)|^{\bar{q}})]^{1/\bar{q}} \leq \bar{L} \bar{\Theta} |u - v|.$$

In view of Assumption 4.6, there exist $\tilde{L}^R = \bar{L} > 0$, $\tilde{q}^R = \bar{q} > 4$, $\tilde{\Theta}^R = 2(\max(\bar{\Theta}^\infty, \bar{\Theta}))^2$ such that for all $n \in \mathbb{N}$, $u, v \in \mathcal{U}_n$, $m = (i, j, a, b) \in \mathcal{D}_n$ we have

$$\sup_{P \in \mathcal{P}_n} [\mathbb{E}_P(|\tilde{G}_{P,n,m}^R(u, \mathcal{H}_{0,m}^R) - \tilde{G}_{P,n,m}^R(v, \mathcal{H}_{0,m}^R)|^{\tilde{q}^R})]^{1/\tilde{q}^R} \leq \tilde{L}^R \tilde{\Theta}^R |u - v|.$$

This follows from adding and subtracting cross-terms, the triangle inequality, the distributive property, Hölder's inequality, and applying the moment bounds and stochastic Lipschitz conditions for the individual error processes from Assumptions 4.5 and 4.6. It is easy to verify that Assumption 3.6 is satisfied under this stronger condition on the nonstationarity.

Also, using the same arguments as Lemma 3.1 from Ding and Zhou [DZ21], the individual dimensions of the vector-valued processes $\mathbf{w}_{P,t,n,i,a}^{\varphi(Z),\varepsilon}$ and $\mathbf{w}_{P,t,n,j,b}^{\varphi(Z),\xi}$ can be shown to satisfy this stochastic Lipschitz condition for moment $\bar{q}/2 > 2$ with Lipschitz constant $2K_1 \bar{L}(\max(\bar{\Theta}^\infty, \bar{\Theta}))^2 > 0$, where the constant factor $K_1 > 0$ is due to the basis functions. Similarly, the individual dimensions of the vector-valued process $\mathbf{w}_{t,n}^{\varphi(Z)}$ can be shown to satisfy this stochastic Lipschitz condition for moment $\bar{q} > 4$ with Lipschitz constant $K_2 \bar{L} \bar{\Theta} > 0$, where the constant factor $K_2 > 0$ is due to the basis functions.

4.6 Assumptions on local long-run covariances

To ensure fast convergence rates by the sieve estimator, we require the following assumptions on the local long-run covariance matrices. Note that these assumptions are not made in Section 3.

Definition 4.2 (Local long-run covariance matrices of error products). *For each $n \in \mathbb{N}$, $P \in \mathcal{P}_n$, $u \in \mathcal{U}_n$, define the local long-run covariance matrix $\tilde{\Sigma}_{P,n}^R(u) \in \mathbb{R}^{D_n \times D_n}$ for the \mathbb{R}^{D_n} -valued stationary process $(\tilde{\mathbf{G}}_{P,n}^R(u, \mathcal{H}_t^R))_{t \in \mathbb{Z}}$ by*

$$\tilde{\Sigma}_{P,n}^R(u) = \sum_{h \in \mathbb{Z}} \text{Cov}_P(\tilde{\mathbf{G}}_{P,n}^R(u, \mathcal{H}_0^R), \tilde{\mathbf{G}}_{P,n}^R(u, \mathcal{H}_h^R)).$$

By Lemma B.5, the local long-run covariance matrices of $\mathbf{w}_{t,n}^{\varphi(Z)}$, $\mathbf{w}_{P,t,n,i,a}^{\varphi(Z),\varepsilon}$, $\mathbf{w}_{P,t,n,j,b}^{\varphi(Z),\xi}$ are well-defined in view of the discussion following Assumption 4.5. Now, we will define the local long-run and integrated long-run covariance matrices of these processes as in Section 3.2 of Ding and Zhou [DZ21].

Definition 4.3 (Local long-run and integrated long-run covariance matrices). *For each $n \in \mathbb{N}$, $P \in \mathcal{P}_n$, $u \in \mathcal{U}_n$, $i \in [d_X]$, $a \in A_i$, $j \in [d_Y]$, $b \in B_j$, define the local long-run covariance matrices $\tilde{\Sigma}_{P,n}^{\mathbf{w}^{\varphi(Z)}}(u)$, $\tilde{\Sigma}_{P,n,i,a}^{\mathbf{w}^{\varphi(Z),\varepsilon}}(u)$, $\tilde{\Sigma}_{P,n,j,b}^{\mathbf{w}^{\varphi(Z),\xi}}(u) \in \mathbb{R}^{\mathbf{d}_Z \tilde{d}_n \times \mathbf{d}_Z \tilde{d}_n}$ for the $\mathbb{R}^{\mathbf{d}_Z \tilde{d}_n}$ -valued stationary processes $(\tilde{\mathbf{G}}_n^{\mathbf{w}^{\varphi(Z)}}(u, \mathcal{H}_t^R))_{t \in \mathbb{Z}}$, $(\tilde{\mathbf{G}}_{P,n,i,a}^{\mathbf{w}^{\varphi(Z),\varepsilon}}(u, \mathcal{H}_t^R))_{t \in \mathbb{Z}}$, $(\tilde{\mathbf{G}}_{P,n,j,b}^{\mathbf{w}^{\varphi(Z),\xi}}(u, \mathcal{H}_t^R))_{t \in \mathbb{Z}}$, respectively, by*

$$\begin{aligned}\tilde{\Sigma}_{P,n}^{\mathbf{w}^{\varphi(Z)}}(u) &= \sum_{h \in \mathbb{Z}} \text{Cov}_P(\tilde{\mathbf{G}}_n^{\mathbf{w}^{\varphi(Z)}}(u, \mathcal{H}_0^{\mathbf{w}^{\varphi(Z)}}), \tilde{\mathbf{G}}_n^{\mathbf{w}^{\varphi(Z)}}(u, \mathcal{H}_h^{\mathbf{w}^{\varphi(Z)}})), \\ \tilde{\Sigma}_{P,n,i,a}^{\mathbf{w}^{\varphi(Z),\varepsilon}}(u) &= \sum_{h \in \mathbb{Z}} \text{Cov}_P(\tilde{\mathbf{G}}_{P,n,i,a}^{\mathbf{w}^{\varphi(Z),\varepsilon}}(u, \mathcal{H}_{0,a}^{\mathbf{w}^{\varphi(Z),\varepsilon}}), \tilde{\mathbf{G}}_{P,n,i,a}^{\mathbf{w}^{\varphi(Z),\varepsilon}}(u, \mathcal{H}_{h,a}^{\mathbf{w}^{\varphi(Z),\varepsilon}})), \\ \tilde{\Sigma}_{P,n,j,b}^{\mathbf{w}^{\varphi(Z),\xi}}(u) &= \sum_{h \in \mathbb{Z}} \text{Cov}_P(\tilde{\mathbf{G}}_{P,n,j,b}^{\mathbf{w}^{\varphi(Z),\xi}}(u, \mathcal{H}_{0,b}^{\mathbf{w}^{\varphi(Z),\xi}}), \tilde{\mathbf{G}}_{P,n,j,b}^{\mathbf{w}^{\varphi(Z),\xi}}(u, \mathcal{H}_{h,b}^{\mathbf{w}^{\varphi(Z),\xi}})).\end{aligned}$$

Next, for each $n \in \mathbb{N}$, $P \in \mathcal{P}_n$, $u \in \mathcal{U}_n$, $i \in [d_X]$, $a \in A_i$, $j \in [d_Y]$, $b \in B_j$, define the corresponding integrated long-run covariance matrices $\tilde{\Sigma}_{P,n}^{\mathbf{w}^{\varphi(Z)}}$, $\tilde{\Sigma}_{P,n,i,a}^{\mathbf{w}^{\varphi(Z),\varepsilon}}$, $\tilde{\Sigma}_{P,n,j,b}^{\mathbf{w}^{\varphi(Z),\xi}} \in \mathbb{R}^{\mathbf{d}_Z \tilde{c}_n \tilde{d}_n \times \mathbf{d}_Z \tilde{c}_n \tilde{d}_n}$ by

$$\begin{aligned}\tilde{\Sigma}_{P,n}^{\mathbf{w}^{\varphi(Z)}} &= \int_{\mathcal{U}_n} \tilde{\Sigma}_{P,n}^{\mathbf{w}^{\varphi(Z)}}(u) \otimes (\phi(u) \phi^\top(u)) du, \\ \tilde{\Sigma}_{P,n,i,a}^{\mathbf{w}^{\varphi(Z),\varepsilon}} &= \int_{\mathcal{U}_n} \tilde{\Sigma}_{P,n,i,a}^{\mathbf{w}^{\varphi(Z),\varepsilon}}(u) \otimes (\phi(u) \phi^\top(u)) du, \\ \tilde{\Sigma}_{P,n,j,b}^{\mathbf{w}^{\varphi(Z),\xi}} &= \int_{\mathcal{U}_n} \tilde{\Sigma}_{P,n,j,b}^{\mathbf{w}^{\varphi(Z),\xi}}(u) \otimes (\phi(u) \phi^\top(u)) du,\end{aligned}$$

where $\phi(u) = (\phi_1(u), \dots, \phi_{\tilde{c}_n}(u))^\top$.

We require the following regularity assumption due to the sieve estimator, which is analogous to Assumption 3.2 from Ding and Zhou [DZ21]. Specifically, for some sequence of collections of distributions $(\mathcal{P}_n)_{n \in \mathbb{N}}$, we impose a distribution-uniform lower bound on the eigenvalues of the integrated long-run covariance matrices.

Assumption 4.7 (Eigenvalue condition for integrated long-run covariance matrices). *Recall $\tilde{\Sigma}_{P,n}^{\mathbf{w}^{\varphi(Z)}}$, $\tilde{\Sigma}_{P,n,i,a}^{\mathbf{w}^{\varphi(Z),\varepsilon}}$, $\tilde{\Sigma}_{P,n,j,b}^{\mathbf{w}^{\varphi(Z),\xi}}$ from Definition 4.3. Assume that there exists a universal constant $\kappa > 0$ such that for all $n \in \mathbb{N}$, $u \in \mathcal{U}_n$, $i \in [d_X]$, $a \in A_i$, $j \in [d_Y]$, $b \in B_j$, we have*

$$\inf_{P \in \mathcal{P}_n} \min(\lambda_{\min}(\tilde{\Sigma}_{P,n}^{\mathbf{w}^{\varphi(Z)}}), \lambda_{\min}(\tilde{\Sigma}_{P,n,i,a}^{\mathbf{w}^{\varphi(Z),\varepsilon}}), \lambda_{\min}(\tilde{\Sigma}_{P,n,j,b}^{\mathbf{w}^{\varphi(Z),\xi}})) \geq \kappa,$$

where $\lambda_{\min}(\cdot)$ is the smallest eigenvalue of the given matrix.

Again, we emphasize that the locally stationary framework in this section fits into the more general triangular array framework from Section 3. Hence, we can use the same cumulative covariance estimator $\hat{Q}_{t,n}^R$ from Section 3.5 for the cumulative covariance matrices $Q_{P,t,n}^R = \sum_{s=\mathbb{T}_n^-}^t \Sigma_{P,s,n}^R$, where $\Sigma_{P,s,n}^R = \tilde{\Sigma}_{P,n}^R(s/n)$ denotes the local long-run covariance matrix at time $s \in \mathcal{T}_n$.

4.7 Theoretical result for Sieve-dGCM

The main result of this section is that the Sieve-dGCM test—implemented by running Algorithm 1 with the predictions from the sieve estimator—will have uniformly asymptotic Type-I error control under the previously stated assumptions.

Theorem 4.1. *Suppose that Assumptions 4.1, 4.2, 4.3, 4.4, 4.5, 4.6, 4.7 all hold for the sequence of collections of distributions $(\mathcal{P}_{0,n}^*)_{n \in \mathbb{N}}$, where $\mathcal{P}_{0,n}^* \subset \mathcal{P}_{0,n}^{\text{CI}}$ for each $n \in \mathbb{N}$. Further, suppose that we use*

the sieve time-varying regression estimator from Section 4.3 with the basis functions $\{\phi_{\ell_1}(u)\}, \{\varphi_{\ell_2}(z)\}$ chosen to be mapped Legendre polynomials, where the numbers of basis functions are chosen to satisfy $\tilde{c}_n = O(\log(T_n))$, $\tilde{d}_n = O(\log(T_n))$. Then the assumptions of Theorem 3.1 hold for $(\mathcal{P}_{0,n}^*)_{n \in \mathbb{N}}$, and the sieve estimators will achieve the convergence rates required by Theorem 3.1.

Throughout this section, we have used Legendre polynomials as the basis functions. In the next section, we investigate the finite sample performance of the Sieve-dGCM test using Legendre basis functions. We emphasize that the Legendre polynomials in our theoretical analysis and simulations can easily be substituted with trigonometric polynomials, wavelets, or other Jacobi polynomials.

5 Numerical Simulations

In this section, we evaluate the Sieve-dGCM test, which consists of running Algorithm 1 based on the predictions from the sieve time-varying nonlinear regression estimator. We use Legendre polynomials as the basis functions as in Section 4. The numbers of basis functions for time and the covariate values were chosen using the subsampling cross-validation procedure proposed in Section 5.1. The lag-window parameter for covariance estimation was selected via the minimum volatility method from Section 5.1. We use $s = 5000$ Monte Carlo simulations to approximate the desired quantile of the test statistic.

We compare the dGCM test with the generalized covariance measure (GCM) test [SP20] using a generalized additive model, and the residual prediction test (RPT) [SB18; HPM18] using the Nyström method and a random forest model. We use the implementations from the `CondIndTests` R package [HPM18]. We also examine how the dGCM test performs in the hypothetical scenario in which the time-varying regression functions are estimated perfectly. We refer to this test as Oracle-dGCM, which consists of running Algorithm 1 using the predictions from the true time-varying regression functions. We first generate 100 realizations of the processes at sample sizes $n \in \{250, 500, 750, 1000\}$, and then we calculate the empirical rejection rates for each test using the significance level $\alpha = 0.05$. For the sake of notational simplicity, in this section we suppress the dependence on the distribution $P \in \mathcal{P}_n$.

5.1 Parameter selection

To begin, we introduce a novel cross-validation approach based on subsampling which can be used for selecting the parameters of “global” estimators of time-varying regression functions. The approach we present here is designed for the case where the global estimator is fit once on all the data. When using sequential estimation as in Remark 4.1, standard approaches for time series cross-validation can be used; see Section 5.10 of Hyndman [Hyn18].

Our approach complements the cross-validation procedure suggested in Section 5.1 of Ding and Zhou [DZ21], which is only for parameter selection in the autoregressive forecasting setting. Also, we note that Dahlhaus and Richter [DR19; DR23] theoretically investigated cross-validation for locally stationary processes in the context of selecting bandwidths for kernel smoothing estimators (i.e. a “local” estimation approach). In contrast, our proposed cross-validation approach is for “global” estimators, such as the sieve estimator from Section 4.

The main idea of our cross-validation scheme is to create several folds constructed by sampling the original time series at a lower sampling frequency. Specifically, for some buffer $\gamma \in \mathbb{N}_0$ and index $k = 1, \dots, 2(\gamma + 1)$, the k -th fold will consist of the subsampled time series

$$\mathcal{T}_n^{(k)} = \{\mathbb{T}_n^- + k - 1 + 2j(\gamma + 1) : j = 0, 1, \dots, \lfloor \frac{\mathbb{T}_n^+ - \mathbb{T}_n^- - k + 1}{2(\gamma + 1)} \rfloor\}.$$

For instance, when the buffer $\gamma = 0$, we have $\mathcal{T}_n^{(1)} = \{\mathbb{T}_n^-, \mathbb{T}_n^- + 2, \dots\}$ and $\mathcal{T}_n^{(2)} = \{\mathbb{T}_n^- + 1, \mathbb{T}_n^- + 3, \dots\}$. Similarly, when the buffer $\gamma = 1$, we have $\mathcal{T}_n^{(1)} = \{\mathbb{T}_n^-, \mathbb{T}_n^- + 4, \dots\}$, $\mathcal{T}_n^{(2)} = \{\mathbb{T}_n^- + 1, \mathbb{T}_n^- + 5, \dots\}$, $\mathcal{T}_n^{(3)} = \{\mathbb{T}_n^- + 2, \mathbb{T}_n^- + 6, \dots\}$, and $\mathcal{T}_n^{(4)} = \{\mathbb{T}_n^- + 3, \mathbb{T}_n^- + 7, \dots\}$. The reason we refer to γ as a buffer will be made clear below.

We describe our cross-validation scheme in the context of a basic grid search procedure for pedagogical reasons. For each parameter combination, do the following. For each index $k = 1, \dots, \gamma + 1$, use the k -th fold $\mathcal{T}_n^{(k)}$ to estimate the entire time-varying regression function (i.e. on a suitably fine grid of rescaled times and covariate values) using the “global” estimator. Afterwards, calculate the residuals

based on the observations in the $(k + \gamma + 1)$ -th fold $\mathcal{T}_n^{(k+\gamma+1)}$. By construction, there are γ time points in between the observations in $\mathcal{T}_n^{(k)}$ and $\mathcal{T}_n^{(k+\gamma+1)}$. Next, reverse the roles of the folds. That is, for each index $k = 1, \dots, \gamma + 1$, estimate the entire time-varying regression function (i.e. on a suitably fine grid) using the $(k + \gamma + 1)$ -th fold $\mathcal{T}_n^{(k+\gamma+1)}$, and then calculate the corresponding residuals based on the observations in the k -th fold $\mathcal{T}_n^{(k)}$. Finally, for each $k = 1, \dots, 2(\gamma + 1)$, calculate the mean squared error $\text{MSE}^{(k)}$ based on the residuals in fold $\mathcal{T}_n^{(k)}$. Select the parameter combination which yields the lowest average mean squared error

$$\overline{\text{MSE}} = \frac{1}{2(\gamma + 1)} \sum_{k=1}^{2(\gamma+1)} \text{MSE}^{(k)}.$$

In practice, γ should be chosen large enough to account for the temporal dependence, but small enough so that there is enough data to estimate the time-varying regression functions. In our simulations with Sieve-dGCM, we use the buffer $\gamma = 1$ and the grid $\{2, 4, 6, 8, 10\} \times \{2, 4, 6, 8, 10\}$ for the numbers of sieve basis functions for time and the covariate values. Note that we allow for the regressions of X on \mathbf{Z} and Y on \mathbf{Z} to have different numbers of basis functions. In future work, we will study the statistical properties of this cross-validation procedure as the buffer $\gamma = \gamma_n$ grows with the sample size n using infill asymptotics. For now, this cross-validation approach serves as a practical technique for parameter selection for generic “global” estimators of time-varying regression functions, such as the sieve estimator.

Next, we discuss how to select the lag-window size parameter L_n for the covariance estimator with a version of the minimum volatility method suggested by Luo and Wu [LW23]. First, select $H \in \mathbb{N}$ candidate lag-window sizes $l_1 < l_2 < \dots < l_H$. For each index $h = 1, \dots, H$, let

$$\hat{\Sigma}_{t,n,l_h} = \frac{1}{l_h} \left(\sum_{s=t-l_h+1}^t \hat{R}_{s,n} \right)^{\otimes 2}$$

be the lag-window estimate of the local long-run covariance matrix at time t using the candidate lag-window size $l_h \in \mathbb{N}$. Second, calculate the minimum volatility criterion for each $j = 1, \dots, H$,

$$\mathbf{MV}(j) = \max_{t=\mathbb{T}_n^-+l_H, \dots, \mathbb{T}_n^+} \text{se}[(\hat{\Sigma}_{t,n,l_h})_{h=1 \vee (j-\Delta)}^{H \wedge (j+\Delta)}],$$

where $\Delta \in \mathbb{N}$ is chosen heuristically to balance robustness and adaptivity, and

$$\text{se}[(\hat{\Sigma}_{t,n,l_h})_{h=h_1}^{h_2}] = \text{tr} \left[\frac{1}{h_2 - h_1 + 1} \sum_{h=h_1}^{h_2} \left(\hat{\Sigma}_{t,n,l_h} - \frac{1}{h_2 - h_1 + 1} \sum_{l=h_1}^{h_2} \hat{\Sigma}_{t,n,l_h} \right)^2 \right]^{1/2},$$

with $h_1 = 1 \vee (j - \Delta)$ and $h_2 = H \wedge (j + \Delta)$. Third, select the lag-window size L_n^* that corresponds to the index j^* which yields the smallest minimum volatility criterion

$$j^* = \arg \min_{j=1, \dots, H} \mathbf{MV}(j).$$

We use the following setup in our simulations. We consider $H = \lfloor n^{3/4} \rfloor$ candidate lag-windows with sizes $l_1 = 1, l_2 = 2, \dots, l_H = \lfloor n^{3/4} \rfloor$. We use $\Delta = 12$ so that 25 consecutive lag-window sizes are typically used in the calculation of the minimum volatility criterion $\mathbf{MV}(j)$ for each $j = 1, \dots, H$.

5.2 Time-varying nonlinear regression with correlated shocks

We investigate the setting with $d_X = 1, d_Y = 1, d_Z = 1$ and no time-offsets, so that $A = \{0\}, B = \{0\}, C = \{0\}$, and $\mathcal{T}_n = [n]$. We test for the null hypothesis

$$X_{t,n} \perp\!\!\!\perp Y_{t,n} \mid Z_{t,n} \text{ for all times } t \in \mathcal{T}_n,$$

versus the alternative hypothesis

$$X_{t,n} \not\perp\!\!\!\perp Y_{t,n} \mid Z_{t,n} \text{ for all times } t \in \mathcal{T}_n,$$

because we assume that we can restrict the collection of alternative distributions to be those for processes with time-invariant conditional dependencies. We couple the processes X and Y by using correlated shocks for the error processes. Let the covariate process be a tvAR(1) process (i.e. a time-varying AR(1) process) defined by

$$Z_{t,n} = \theta^Z(t/n)Z_{t-1,n} + \eta_t^Z,$$

where the parameter curve $\theta^Z : [0, 1] \rightarrow \mathbb{R}$ is given by $\theta^Z(u) = 0.35 + 0.2 \cos(2\pi u)$, and the shocks $(\eta_t^Z)_{t \in \mathcal{T}_n}$ are sampled iid from a standard normal distribution. Let

$$X_{t,n} = f_K(Z_{t,n}, t/n) + \varepsilon_{t,n}, \quad Y_{t,n} = g_K(Z_{t,n}, t/n) + \xi_{t,n},$$

where the functions $f_K, g_K : \mathbb{R} \times [0, 1] \rightarrow \mathbb{R}$ are defined by

$$f_K(z, u) = (0.5 + 0.25 \cos(2\pi u)) \exp(-z^2) \sin(Kz), \quad (15)$$

$$g_K(z, u) = (0.3 + 0.15 \sin(\pi u)) \exp(-z^2) \cos(Kz), \quad (16)$$

with regression complexity parameter $K \in \{1, 2, 3, 4\}$.

Define the error processes as

$$\begin{aligned} \varepsilon_{t,n} &= \sigma^\varepsilon(Z_{t,n}, t/n) \varepsilon'_{t,n}, \quad \varepsilon'_{t,n} = \theta^\varepsilon(t/n) \varepsilon'_{t-1,n} + \eta_t^\varepsilon, \\ \xi_{t,n} &= \sigma^\xi(Z_{t,n}, t/n) \xi'_{t,n}, \quad \xi'_{t,n} = \theta^\xi(t/n) \xi'_{t-1,n} + \eta_t^\xi, \end{aligned}$$

where the parameter curves $\theta^\varepsilon, \theta^\xi : [0, 1] \rightarrow \mathbb{R}$ are given by

$$\theta^\varepsilon(u) = 0.4 + 0.2 \sin(\pi u), \quad \theta^\xi(u) = 0.5 + 0.25 \sin(2\pi u),$$

and where the functions $\sigma^\varepsilon, \sigma^\xi : \mathbb{R} \times [0, 1] \rightarrow \mathbb{R}$ are given by

$$\begin{aligned} \sigma^\varepsilon(z, u) &= 0.2 + (0.5 + 0.25 \sin(2\pi u)) \left(\frac{\exp(-5z)}{1 + \exp(-5z)} \right), \\ \sigma^\xi(z, u) &= 0.5 + (0.4 + 0.2 \cos(2\pi u)) \exp(-z^2) \sin(z). \end{aligned}$$

The shocks $(\eta_t^\varepsilon, \eta_t^\xi)_{t \in \mathcal{T}_n}$ are sampled iid from a centered bivariate normal distribution with unit variances and correlation $\rho \in \{0, 0.3, 0.6, 0.9\}$

$$\begin{bmatrix} \eta_t^\varepsilon \\ \eta_t^\xi \end{bmatrix} \sim \mathcal{N} \left(\begin{bmatrix} 0 \\ 0 \end{bmatrix}, \begin{bmatrix} 1 & \rho \\ \rho & 1 \end{bmatrix} \right), \quad t \in \mathcal{T}_n.$$

The null hypothesis is true when ρ is zero, and the alternative hypothesis is true when it is nonzero.

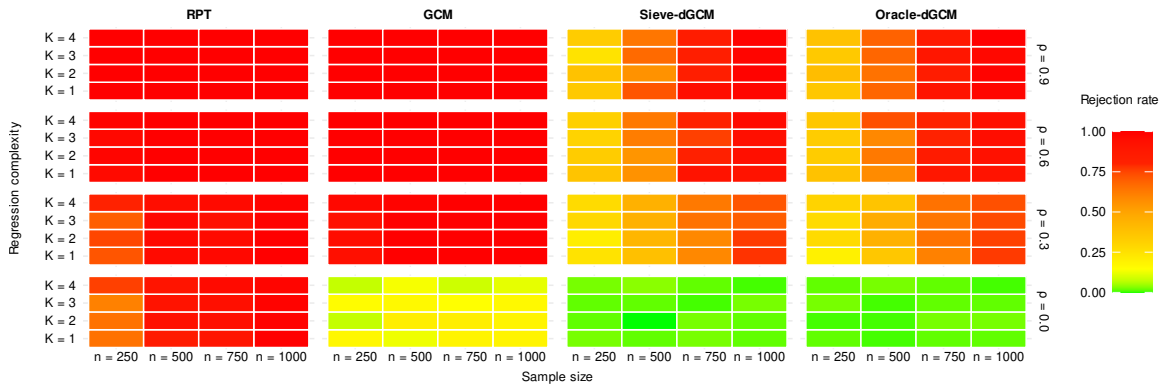


Figure 2: The Sieve-dGCM test holds the level even with fairly small sample sizes, and gains power as we increase the correlation and sample size. The other tests fail to hold the level.

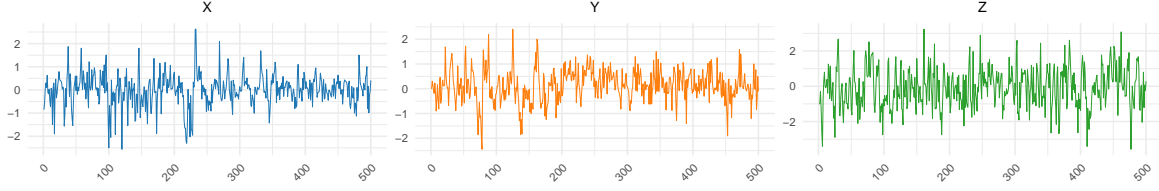


Figure 3: One realization of length $n = 500$ from the null distribution with $\rho = 0$ and $K = 1$.

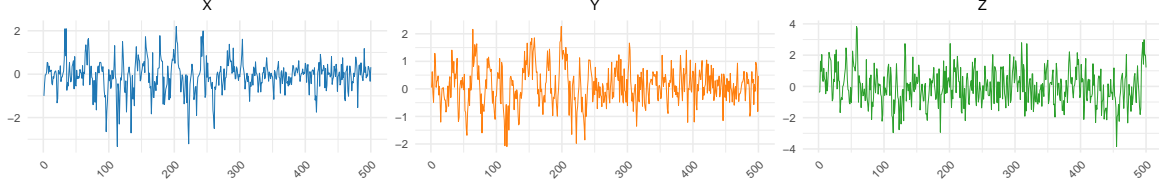


Figure 4: One realization of length $n = 500$ from the alternative distribution with $\rho = 0.9$ and $K = 1$.

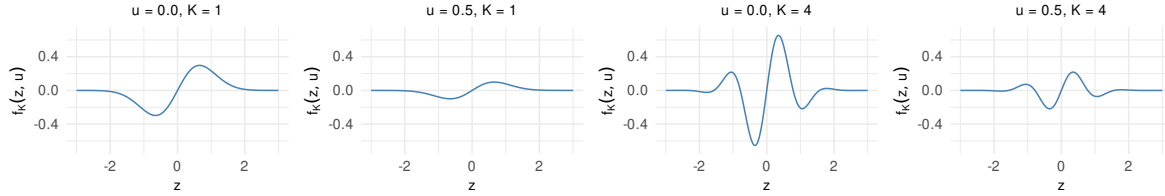


Figure 5: The time-varying regression function $f_K(z, u)$ from (15) at different times and complexities.

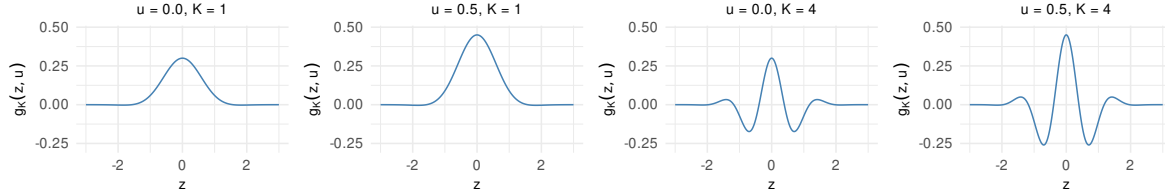


Figure 6: The time-varying regression function $g_K(z, u)$ from (16) at different times and complexities.

5.3 Illustration of no-free-lunch in conditional independence testing

To demonstrate the hardness of conditional independence testing, we consider a variant of a setup from Shah and Peters [SP20]. We work in the setting with $d_X = 1$, $d_Y = 1$, $d_Z = 1$ and no time-offsets, so that $A = \{0\}$, $B = \{0\}$, $C = \{0\}$, and $\mathcal{T}_n = [n]$. We test for the null hypothesis

$$X_{t,n} \perp\!\!\!\perp Y_{t,n} \mid Z_{t,n} \text{ for all times } t \in \mathcal{T}_n,$$

versus the alternative

$$X_{t,n} \not\perp\!\!\!\perp Y_{t,n} \mid Z_{t,n} \text{ for all times } t \in \mathcal{T}_n,$$

since we assume we can restrict the collection of alternative distributions to be those for processes with time-invariant conditional dependencies. This setup is challenging because X and Y have the same time-varying regression functions, so similar prediction errors are likely to occur at the same covariate values and rescaled times. When the regression complexity is high relative to the sample size, this induces non-negligible correlation between the residuals even when the error processes are uncorrelated under the null.

Let the covariate process be defined by

$$Z_{t,n} = \theta^Z(t/n)Z_{t-1,n} + \eta_t^Z,$$

where the parameter curve $\theta^Z : [0, 1] \rightarrow \mathbb{R}$ is defined by $\theta^Z(u) = 0.5 + 0.25 \cos(\pi u)$ and the shocks $(\eta_t^Z)_{t \in \mathcal{T}_n}$ are sampled iid from a standard normal distribution. The processes X and Y here are coupled due to X having an additive effect on Y at each time. Let

$$X_{t,n} = f_K(Z_{t,n}, t/n) + \varepsilon_{t,n}, \quad Y_{t,n} = f_K(Z_{t,n}, t/n) + \beta X_{t,n} + \xi_{t,n},$$

with effect size $\beta \in \{0, 0.3, 0.6, 0.9\}$, where the function $f_K : \mathbb{R} \times [0, 1] \rightarrow \mathbb{R}$ is defined by

$$f_K(z, u) = (0.4 + 0.2 \sin(2\pi u)) \exp(-z^2) \sin(Kz),$$

with regression complexity parameter $K \in \{1, 2, 3, 4\}$. Let the error processes be given by

$$\varepsilon_{t,n} = 0.3\varepsilon'_{t,n}, \quad \varepsilon'_{t,n} = \theta^e(t/n)\varepsilon'_{t-1,n} + \eta_t^\varepsilon,$$

$$\xi_{t,n} = 0.3\xi'_{t,n}, \quad \xi'_{t,n} = \theta^e(t/n)\xi'_{t-1,n} + \eta_t^\xi,$$

where the parameter curve $\theta^e : [0, 1] \rightarrow \mathbb{R}$ is given by $\theta^e(u) = 0.45 + 0.3 \sin(2\pi u)$. The shocks $(\eta_t^\varepsilon)_{t \in \mathcal{T}_n}, (\eta_t^\xi)_{t \in \mathcal{T}_n}$ are each sampled iid from a standard normal distribution. The null hypothesis is true when the effect size β is zero, and the alternative hypothesis is true when it is nonzero.

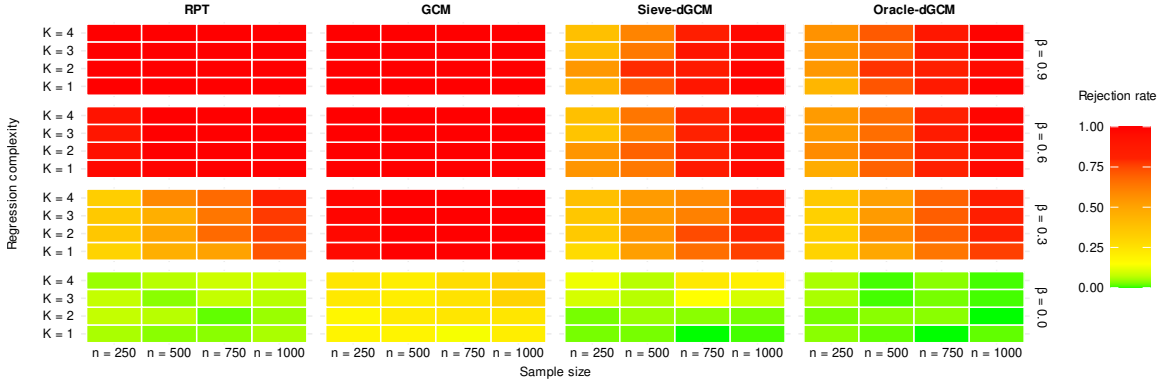


Figure 7: In this setup with identical time-varying regression functions, the Sieve-dGCM test fails to hold the level when the time-varying regression functions are too complex to reliably estimate at the given sample size. All of the tests can detect conditional dependencies.

This simulation setup provides an empirical demonstration of the no-free-lunch results from [SP20; BP24]. By letting the regression complexity K_n grow with the sample size n at a fairly rapid rate so that $K_{250} = 1$, $K_{500} = 2$, $K_{750} = 3$, $K_{1000} = 4$, and so on, we see the Sieve-dGCM test lose control of the Type-I error. A similar phenomena can also be observed with the original GCM test in Section 5 of Shah and Peters [SP20].

Consider a sequence of null distributions from this simulation setup parametrized by the sequence of regression complexity parameters $K_n = \lfloor Cn^r \rfloor$, $n \in \mathbb{N}$, for some $r \geq 1/5$, $C > 0$. In this case, the number of basis functions for the covariate values must also grow polynomially in n , since the n -th Legendre polynomial is a degree n polynomial. As a result, the convergence rate of the sieve estimator will be slower than the rate required by Theorem 3.1. See Theorem 3.2 and Appendix C of Ding and Zhou [DZ21] for the details about how the convergence rate of the sieve estimator is affected by the growth rate of the number of basis functions.

Consequently, the uniform asymptotic Type-I error control guarantee we provide for the dGCM test in Theorem 3.1 is not applicable. On the other hand, this guarantee is applicable if we fix $K_n = K$ to some positive constant or have K_n grow slowly in the sample size (e.g. logarithmically). This illustration helps us understand when GCM-type tests can fail to control Type-I error, and highlights the transparency of uniform level guarantees.

In light of the no-free-lunch results, we understand that it is impossible to ensure the correct significance level for every null distribution, no matter how large the sample size. That is, there will always be some null distribution in which the Type-I error rate exceeds the prespecified significance level. As Shah [Sha25] puts it, “with great power comes great Type-I error.”

6 Real Data Application: Links in International Markets

We investigate how stock markets in the United States, United Kingdom, Hong Kong, and Japan are linked. Our dataset¹ consists of daily log returns based on the adjusted closing prices of the S&P 500, FTSE 100, Hang Seng, and Nikkei 225 from January 2022 to April 2025. To deal with holidays observed by each exchange, we linearly interpolate based on the daily log returns from before and after the holiday. This yields $n = 860$ observations. Denote the time series of daily log returns by **S&P**, **FTSE**, **HangSeng**, **Nikkei**, and the daily log return on day t by $\mathbf{S\&P}(t)$, $\mathbf{FTSE}(t)$, $\mathbf{HangSeng}(t)$, $\mathbf{Nikkei}(t)$.

A great deal of work has been devoted to this topic; see [ES89; KK91; BY03; GZ13] and the references therein. For instance, Gao and Zhao [GZ13] find that the market in the United States acts as an intermediary in the transmission of information between Asia-Pacific markets and European markets. The authors suggest that this could reflect varying degrees of economic integration. We emphasize that Gao and Zhao [GZ13] analyze weekly log returns from 2002 to 2010, whereas we use daily log returns from the post-COVID-19 era. Thus, our study captures a fundamentally different market environment at a finer temporal resolution. For more discussion about time resolution and nonstationarity in finance, see Dong et al. [Don+23].

To begin, we test for (unconditional) independence between the daily log returns of each stock market index. For each pair of stock indices $X, Y \in \{\mathbf{S\&P}, \mathbf{FTSE}, \mathbf{HangSeng}, \mathbf{Nikkei}\}$, where $X \neq Y$, we test the null hypothesis

$$X_{t,n} \perp\!\!\!\perp Y_{t,n} \text{ for all times } t \in \mathcal{T}_n,$$

versus the alternative hypothesis

$$X_{t,n} \not\perp\!\!\!\perp Y_{t,n} \text{ for all times } t \in \mathcal{T}_n,$$

because we assume that we can restrict the collection of alternative distributions to be those for processes with time-invariant dependencies. We use the independence test for nonstationary nonlinear time series from Section C.10, which consists of running Algorithm 1 using the sieve estimator from Section 4 to estimate the unconditional means at each time, rather than the conditional means at each time. Otherwise, the setup is the same as that of the Sieve-dGCM test used in Section 5. We reject each null hypothesis of independence at the significance level $\alpha = 0.05$.

Next, we test for conditional independence between each pair of stock market indices given one of the others. For each triplet $X, Y, Z \in \{\mathbf{S\&P}, \mathbf{FTSE}, \mathbf{HangSeng}, \mathbf{Nikkei}\}$, where $X \neq Y$, $X \neq Z$, and $Y \neq Z$, we test the null hypothesis

$$X_{t,n} \perp\!\!\!\perp Y_{t,n} \mid Z_{t,n} \text{ for all times } t \in \mathcal{T}_n,$$

versus the alternative hypothesis

$$X_{t,n} \not\perp\!\!\!\perp Y_{t,n} \mid Z_{t,n} \text{ for all times } t \in \mathcal{T}_n,$$

because we assume we can restrict the collection of alternative distributions to be those in which the processes have time-invariant conditional dependence relationships. We use the Sieve-dGCM test with significance level $\alpha = 0.05$ using the same setup as Section 5. We focus our discussion on the main takeaways, and summarize our results as a conditional independence graph. For the exact results of each test, see Section C.12.

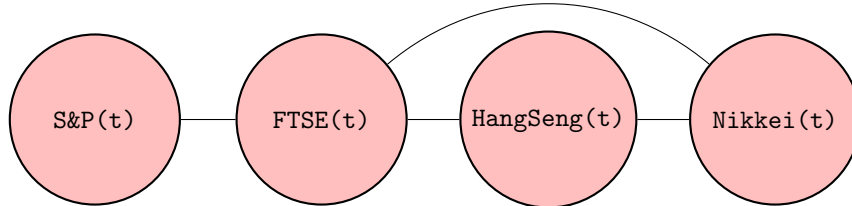


Figure 8: Conditional independence graph for the stock market indices.

¹The data can be downloaded for free from Yahoo Finance.

On each day, the related stock exchanges close in the following order, from earliest to latest: Japan Exchange Group (Tokyo Stock Exchange), Hong Kong Stock Exchange, London Stock Exchange, and New York Stock Exchange. Given this domain knowledge of the temporal ordering of stock exchange closing times, it makes sense that $\text{S\&P}(t)$ is independent of $\text{HangSeng}(t)$ and $\text{Nikkei}(t)$ given $\text{FTSE}(t)$, and also that $\text{S\&P}(t)$ is not independent of $\text{FTSE}(t)$ given either $\text{HangSeng}(t)$ or $\text{Nikkei}(t)$. The simplest explanation for this is that the market in the United Kingdom can absorb the closing price information from Asia-Pacific markets on the same day, but not vice versa. Interestingly, we also find that $\text{FTSE}(t)$ is not independent of $\text{Nikkei}(t)$ given $\text{HangSeng}(t)$. As suggested by Gao and Zhao [GZ13], this could reflect varying degrees of economic integration.

It would be of interest to supplement this analysis with additional macroeconomic time series and to use a causal discovery algorithm for nonstationary nonlinear time series; see Section C.9 for more discussion. For instance, Jiang and Shimizu [JS23] use a causal discovery algorithm to identify time-lagged causal relationships among foreign exchange, stock, and bond markets. However, these analyses are beyond the scope of this paper.

7 Conclusion

The dGCM test shows promise for detecting conditional dependencies among nonstationary processes while controlling the Type-I error in finite samples. Specifically, we find that the Sieve-dGCM test can hold the level if the sample size is large enough to reliably estimate the time-varying regression functions. Exhaustive comparisons of the dGCM test with other conditional independence tests for time series will be reported in a separate manuscript, as many of these tests do not have off-the-shelf implementations available and several are computationally demanding (e.g. kernel-based tests).

Specifically, we plan to investigate how the dGCM test performs in causal discovery algorithms for time series and variable selection procedures for forecasting. Also, it is of practical importance to understand the tradeoffs between statistical performance and computational efficiency when using different time-varying regression estimators with the dGCM test. We aim to explore many settings with varying dimensionality, nonlinearity, nonstationarity, and temporal correlation.

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A Proofs of Theoretical Results for dGCM and Sieve-dGCM

We will denote the three bias terms by

$$\begin{aligned}\hat{\mathbf{w}}_{P,t,n}^{\mathbf{f},\mathbf{g}} &= (\hat{w}_{P,t,n,m}^{\mathbf{f},\mathbf{g}})_{m \in \mathcal{D}_n} = (\hat{w}_{P,t,n,i,a}^{\mathbf{f}} \hat{w}_{P,t,n,j,b}^{\mathbf{g}})_{m \in \mathcal{D}_n}, \\ \hat{\mathbf{w}}_{P,t,n}^{\mathbf{g},\boldsymbol{\varepsilon}} &= (\hat{w}_{P,t,n,m}^{\mathbf{g},\boldsymbol{\varepsilon}})_{m \in \mathcal{D}_n} = (\hat{w}_{P,t,n,j,b}^{\mathbf{g}} \varepsilon_{P,t,n,i,a})_{m \in \mathcal{D}_n}, \\ \hat{\mathbf{w}}_{P,t,n}^{\mathbf{f},\boldsymbol{\xi}} &= (\hat{w}_{P,t,n,m}^{\mathbf{f},\boldsymbol{\xi}})_{m \in \mathcal{D}_n} = (\hat{w}_{P,t,n,i,a}^{\mathbf{f}} \xi_{P,t,n,j,b})_{m \in \mathcal{D}_n},\end{aligned}$$

where $m = (i, j, a, b) \in \mathcal{D}_n$. Also, denote

$$\begin{aligned}\hat{\mathbf{w}}_{P,n}^{\mathbf{f},\mathbf{g}} &= (\hat{\mathbf{w}}_{P,t,n}^{\mathbf{f},\mathbf{g}})_{t \in \mathcal{T}_{n,L}}, \\ \hat{\mathbf{w}}_{P,n}^{\mathbf{g},\boldsymbol{\varepsilon}} &= (\hat{\mathbf{w}}_{P,t,n}^{\mathbf{g},\boldsymbol{\varepsilon}})_{t \in \mathcal{T}_{n,L}}, \\ \hat{\mathbf{w}}_{P,n}^{\mathbf{f},\boldsymbol{\xi}} &= (\hat{\mathbf{w}}_{P,t,n}^{\mathbf{f},\boldsymbol{\xi}})_{t \in \mathcal{T}_{n,L}}.\end{aligned}$$

Note that when we write $o_{\mathcal{P}}(\cdot)$ and $O_{\mathcal{P}}(\cdot)$, we will always be doing so with reference to the collection of distributions $\mathcal{P}_{0,n}^*$ defined in the statement of the theorem.

A.1 Proof of Theorem 3.1

Step 1 (Bias Terms): We decompose the products of residuals into the products of errors and the three bias terms, and then apply the triangle inequality and subadditivity, which yields

$$\begin{aligned}& \sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{P}_P(S_{n,p}(\hat{\mathbf{R}}_n) > \hat{q}_{1-\alpha+\nu_n} + \tau_n) \\ & \leq \sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{P}_P(S_{n,p}(\mathbf{R}_{P,n}) > \hat{q}_{1-\alpha+\nu_n} + \frac{\tau_n}{2}) \\ & \quad + \sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{P}_P(S_{n,p}(\hat{\mathbf{w}}_{P,n}^{\mathbf{f},\mathbf{g}}) > \frac{\tau_n}{6}) \\ & \quad + \sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{P}_P(S_{n,p}(\hat{\mathbf{w}}_{P,n}^{\mathbf{g},\boldsymbol{\varepsilon}}) > \frac{\tau_n}{6}) \\ & \quad + \sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{P}_P(S_{n,p}(\hat{\mathbf{w}}_{P,n}^{\mathbf{f},\boldsymbol{\xi}}) > \frac{\tau_n}{6}).\end{aligned}$$

We will handle each of the three bias terms separately.

Step 1.1: Observe that for any $\delta > 0$, we have

$$\begin{aligned}& \sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{P}_P(\tau_n^{-1} S_{n,p}(\hat{\mathbf{w}}_{P,n}^{\mathbf{f},\mathbf{g}}) > \delta) \\ & \stackrel{(1)}{\leq} \delta^{-1} \tau_n^{-1} T_{n,L}^{-\frac{1}{2}} \sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{E}_P \left(\max_{s \in \mathcal{T}_{n,L}} \left\| \sum_{t \leq s} \hat{\mathbf{w}}_{P,t,n}^{\mathbf{f},\mathbf{g}} \right\|_2 \right) \\ & \stackrel{(2)}{\leq} \delta^{-1} \tau_n^{-1} T_{n,L}^{-\frac{1}{2}} D_n \sup_{P \in \mathcal{P}_{0,n}^*} \max_{(i,j,a,b) \in \mathcal{D}_n} \mathbb{E}_P \left(\sum_{t \in \mathcal{T}_{n,L}} |\hat{w}_{P,t,n,i,a}^{\mathbf{f}}| |\hat{w}_{P,t,n,j,b}^{\mathbf{g}}| \right) \\ & \stackrel{(3)}{\leq} \delta^{-1} \tau_n^{-1} T_{n,L}^{\frac{1}{2}} D_n \sup_{P \in \mathcal{P}_{0,n}^*} \max_{(i,j,a,b) \in \mathcal{D}_n} \mathbb{E}_P \left(|\hat{w}_{P,t,n,i,a}^{\mathbf{f}}|^2 \right)^{\frac{1}{2}} \mathbb{E}_P \left(|\hat{w}_{P,t,n,j,b}^{\mathbf{g}}|^2 \right)^{\frac{1}{2}} \\ & \stackrel{(4)}{=} o(1),\end{aligned}$$

where the previous lines follow by (1) Markov's inequality and ℓ_p -norm inequalities, (2) the triangle inequality, ℓ_p -norm inequalities, linearity of expectation, (3) linearity of expectation and the Cauchy-Schwarz inequality, (4) the convergence rate requirements for the time-varying regression estimators.

Step 1.2: Observe that for any $\delta > 0$, we have

$$\begin{aligned}
& \sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{P}_P(\tau_n^{-1} S_{n,p}(\hat{\mathbf{w}}_{P,n}^{g,\varepsilon}) > \delta) \\
& \stackrel{(1)}{\leq} \sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{P}_P \left(\tau_n^{-2} \max_{s \in \mathcal{T}_{n,L}} \left\| \frac{1}{\sqrt{T_{n,L}}} \sum_{t \leq s} \hat{\mathbf{w}}_{P,t,n}^{g,\varepsilon} \right\|_2^2 \geq \delta^2 \right) \\
& \stackrel{(2)}{\leq} \delta^{-2} \tau_n^{-2} T_{n,L}^{-1} \sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{E}_P \left(\max_{s \in \mathcal{T}_{n,L}} \left\| \sum_{t \leq s} \hat{\mathbf{w}}_{P,t,n}^{g,\varepsilon} \right\|_2^2 \right) \\
& \stackrel{(3)}{\leq} \delta^{-2} \tau_n^{-2} T_{n,L}^{-1} (\bar{K} T_{n,L}^{\frac{1}{2}} D_n^{\frac{1}{2}} \sup_{P \in \mathcal{P}_{0,n}^*} \max_{t \in \mathcal{T}_{n,L}} \max_{j \in [d_Y]} \max_{b \in B_j} \mathbb{E}_P(|\hat{w}_{P,t,n,j,b}^g|^2)^{\frac{1}{2}})^2 \\
& \stackrel{(4)}{\leq} \delta^{-2} \tau_n^{-2} \bar{K}^2 D_n \sup_{P \in \mathcal{P}_{0,n}^*} \max_{t \in \mathcal{T}_{n,L}} \max_{j \in [d_Y]} \max_{b \in B_j} \mathbb{E}_P(|\hat{w}_{P,t,n,j,b}^g|^2) \\
& \stackrel{(5)}{=} o(1),
\end{aligned}$$

where the previous lines follow by (1) the assumption about the form of the test statistic and squaring, (2) Markov's inequality and linearity of expectation, (3) for some constant $\bar{K} > 0$ by the arguments below, (4) simplifying the expression, and (5) the convergence rate requirements for the time-varying regression estimator.

The following arguments are to show (3). These arguments are based on the constructions used in the proof of Theorem 3.2 in Mies and Steland [MS23], which build on the proof techniques from Theorem 1 in Liu et al. [LXW13]. For each $t \in \mathcal{T}_{n,L}$ and $h \in \mathbb{N}_0$, let

$$\mathcal{F}_{t,h}^{\hat{\mathbf{w}}^{g,\varepsilon}} = \sigma(\eta_t^\varepsilon, \eta_{t-1}^\varepsilon, \dots, \eta_{t-h}^\varepsilon, \mathcal{H}_t^{\hat{g}}),$$

where the input η_t^ε is from (11) and the input sequence $\mathcal{H}_t^{\hat{g}}$ is defined following Assumption 3.3. For each $n \in \mathbb{N}$, $P \in \mathcal{P}_{0,n}^*$, $t \in \mathcal{T}_{n,L}$, and $h \in \mathbb{N}_0$ let

$$\begin{aligned}
\hat{\mathbf{S}}_{P,t,n,h}^{g,\varepsilon} &= \sum_{k \leq t} \hat{\mathbf{w}}_{P,k,n,h}^{g,\varepsilon}, \\
\hat{\mathbf{w}}_{P,t,n,h}^{g,\varepsilon} &= \mathbb{E}_P(\hat{\mathbf{w}}_{P,t,n}^{g,\varepsilon} | \mathcal{F}_{t,h}^{\hat{\mathbf{w}}^{g,\varepsilon}}), \\
\hat{\mathbf{w}}_{P,t,n,-1}^{g,\varepsilon} &= \mathbb{E}_P(\hat{\mathbf{w}}_{P,t,n}^{g,\varepsilon} | \mathcal{H}_t^{\hat{g}}) = \mathbf{0},
\end{aligned}$$

almost surely, because for each $n \in \mathbb{N}$, $P \in \mathcal{P}_{0,n}^*$, $(i, j, a, b) \in \mathcal{D}_n$, $t \in \mathcal{T}_{n,L}$ we have

$$\mathbb{E}_P(\hat{w}_{P,t,n,j,b}^g \varepsilon_{P,t,n,i,a} | \mathcal{H}_t^{\hat{g}}) = \hat{w}_{P,t,n,j,b}^g \mathbb{E}_P(\varepsilon_{P,t,n,i,a} | \mathcal{H}_t^{\hat{g}}) = 0, \quad (17)$$

almost surely, by Assumptions 3.3 and 3.4. For each $n \in \mathbb{N}$, $P \in \mathcal{P}_{0,n}^*$, $t \in \mathcal{T}_{n,L}$, and $h \in \mathbb{N}_0$ we have

$$\mathbb{E}_P(\|\hat{\mathbf{w}}_{P,t,n,h}^{g,\varepsilon}\|_2^2) < \infty, \quad (18)$$

by linearity of expectation, the contraction property of conditional expectation, and Assumption 3.3. For each $n \in \mathbb{N}$, $P \in \mathcal{P}_{0,n}^*$, $t \in \mathcal{T}_{n,L}$, and $h \in \mathbb{N}_0$, by the tower property we have

$$\mathbb{E}_P(\hat{\mathbf{w}}_{P,t,n,h+1}^{g,\varepsilon} | \mathcal{F}_{t,h}^{\hat{\mathbf{w}}^{g,\varepsilon}}) = \hat{\mathbf{w}}_{P,t,n,h}^{g,\varepsilon},$$

almost surely. Hence, for each $n \in \mathbb{N}$, $P \in \mathcal{P}_{0,n}^*$, and $t \in \mathcal{T}_{n,L}$, $(\hat{\mathbf{w}}_{P,t,n,h}^{g,\varepsilon})_{h=0}^\infty$ is a martingale with respect to the filtration $(\mathcal{F}_{t,h}^{\hat{\mathbf{w}}^{g,\varepsilon}})_{h=0}^\infty$. The martingale convergence theorem (see e.g. Theorem 1.5 of [Pis16]) ensures that for each $n \in \mathbb{N}$, $P \in \mathcal{P}_{0,n}^*$, $t \in \mathcal{T}_{n,L}$ there exists some random vector $\tilde{\mathbf{w}}_{P,t,n}^{g,\varepsilon}$ such that $\mathbb{E}_P\|\tilde{\mathbf{w}}_{P,t,n}^{g,\varepsilon} - \hat{\mathbf{w}}_{P,t,n,h}^{g,\varepsilon}\|_2^2 \rightarrow 0$ as $h \rightarrow \infty$. The measurability of $\mathbf{G}_{P,t,n}^{\hat{\mathbf{w}}^{g,\varepsilon}}$ with respect to the projection σ -algebra, in view of Assumptions 3.1, 3.2, 3.3, 3.4, ensures that $\tilde{\mathbf{w}}_{P,t,n}^{g,\varepsilon} = \hat{\mathbf{w}}_{P,t,n}^{g,\varepsilon}$. Thus, for each $t \in \mathcal{T}_{n,L}$ we have

$$\hat{\mathbf{S}}_{P,t,n}^{g,\varepsilon} = \sum_{k \leq t} \hat{\mathbf{w}}_{P,k,n}^{g,\varepsilon} = \sum_{h=0}^\infty (\hat{\mathbf{S}}_{P,t,n,h}^{g,\varepsilon} - \hat{\mathbf{S}}_{P,t,n,h-1}^{g,\varepsilon}), \quad (19)$$

by telescoping. For each $n \in \mathbb{N}$, $P \in \mathcal{P}_{0,n}^*$, and $h \in \mathbb{N}_0$,

$$(\hat{\mathbf{w}}_{P, \mathbb{T}_n^+ - k, n, h}^{g, \varepsilon} - \hat{\mathbf{w}}_{P, \mathbb{T}_n^+ - k, n, h-1}^{g, \varepsilon})_{k=0}^{\mathbb{T}_n^+ - \mathbb{T}_n^- - L_n},$$

are martingale differences with respect to the filtration $(\mathcal{G}_{\mathbb{T}_n^+, k, h}^{\hat{\mathbf{w}}^{g, \varepsilon}})_{k=0}^{\mathbb{T}_n^+ - \mathbb{T}_n^- - L_n}$, where

$$\mathcal{G}_{\mathbb{T}_n^+, k, h}^{\hat{\mathbf{w}}^{g, \varepsilon}} = \sigma(\mathcal{H}_{\mathbb{T}_n^+ - k}^{\hat{\mathbf{w}}^g}, \eta_{\mathbb{T}_n^+ - k - h}^\varepsilon, \eta_{\mathbb{T}_n^+ - k - h + 1}^\varepsilon \dots),$$

because for any $n \in \mathbb{N}$, $P \in \mathcal{P}_{0,n}^*$, $h \in \mathbb{N}_0$ and $k = 0, 1, \dots$, we have

$$\begin{aligned} & \mathbb{E}_P(\hat{\mathbf{w}}_{P, \mathbb{T}_n^+ - k, n, h}^{g, \varepsilon} - \hat{\mathbf{w}}_{P, \mathbb{T}_n^+ - k, n, h-1}^{g, \varepsilon} | \mathcal{G}_{\mathbb{T}_n^+, k-1, h}^{\hat{\mathbf{w}}^{g, \varepsilon}}) \\ &= \mathbb{E}_P(\mathbb{E}_P(\hat{\mathbf{w}}_{P, \mathbb{T}_n^+ - k, n}^{g, \varepsilon} | \mathcal{F}_{\mathbb{T}_n^+ - k, h}^{\hat{\mathbf{w}}^{g, \varepsilon}}) | \mathcal{G}_{\mathbb{T}_n^+, k-1, h}^{\hat{\mathbf{w}}^{g, \varepsilon}}) \\ &= \mathbb{E}_P(\mathbb{E}_P(\hat{\mathbf{w}}_{P, \mathbb{T}_n^+ - k, n}^{g, \varepsilon} | \mathcal{F}_{\mathbb{T}_n^+ - k, h-1}^{\hat{\mathbf{w}}^{g, \varepsilon}}) | \mathcal{G}_{\mathbb{T}_n^+, k-1, h}^{\hat{\mathbf{w}}^{g, \varepsilon}}) \\ &= \mathbf{0}, \end{aligned}$$

almost surely, because for each $n \in \mathbb{N}$, $P \in \mathcal{P}_{0,n}^*$, $(i, j, a, b) \in \mathcal{D}_n$, $h \in \mathbb{N}_0$ and $k = 0, 1, \dots$, we have

$$\begin{aligned} & \mathbb{E}_P(\mathbb{E}_P(\hat{w}_{P, \mathbb{T}_n^+ - k, n, j, b}^g \varepsilon_{P, \mathbb{T}_n^+ - k, n, i, a} | \mathcal{H}_{\mathbb{T}_n^+ - k}^{\hat{\mathbf{w}}^g}, \eta_{\mathbb{T}_n^+ - k - h}^\varepsilon, \eta_{\mathbb{T}_n^+ - k - h + 1}^\varepsilon, \dots, \eta_{\mathbb{T}_n^+ - k}^\varepsilon) \\ & | \mathcal{H}_{\mathbb{T}_n^+ - k + 1}^{\hat{\mathbf{w}}^g}, \eta_{\mathbb{T}_n^+ - k - h + 1}^\varepsilon, \eta_{\mathbb{T}_n^+ - k - h + 2}^\varepsilon, \dots) \\ & - \mathbb{E}_P(\mathbb{E}_P(\hat{w}_{P, \mathbb{T}_n^+ - k, n, j, b}^g \varepsilon_{P, \mathbb{T}_n^+ - k, n, i, a} | \mathcal{H}_{\mathbb{T}_n^+ - k}^{\hat{\mathbf{w}}^g}, \eta_{\mathbb{T}_n^+ - k - h + 1}^\varepsilon, \eta_{\mathbb{T}_n^+ - k - h + 2}^\varepsilon, \dots, \eta_{\mathbb{T}_n^+ - k}^\varepsilon) \\ & | \mathcal{H}_{\mathbb{T}_n^+ - k + 1}^{\hat{\mathbf{w}}^g}, \eta_{\mathbb{T}_n^+ - k - h + 1}^\varepsilon, \eta_{\mathbb{T}_n^+ - k - h + 2}^\varepsilon, \dots) \\ & \stackrel{(1)}{=} \hat{w}_{P, \mathbb{T}_n^+ - k, n, j, b}^g \mathbb{E}_P(\mathbb{E}_P(\varepsilon_{P, \mathbb{T}_n^+ - k, n, i, a} | \mathcal{H}_{\mathbb{T}_n^+ - k}^{\hat{\mathbf{w}}^g}, \eta_{\mathbb{T}_n^+ - k - h}^\varepsilon, \eta_{\mathbb{T}_n^+ - k - h + 1}^\varepsilon, \dots, \eta_{\mathbb{T}_n^+ - k}^\varepsilon) \\ & | \mathcal{H}_{\mathbb{T}_n^+ - k + 1}^{\hat{\mathbf{w}}^g}, \eta_{\mathbb{T}_n^+ - k - h + 1}^\varepsilon, \eta_{\mathbb{T}_n^+ - k - h + 2}^\varepsilon, \dots, \eta_{\mathbb{T}_n^+ - k}^\varepsilon) \\ & - \hat{w}_{P, \mathbb{T}_n^+ - k, n, j, b}^g \mathbb{E}_P(\mathbb{E}_P(\varepsilon_{P, \mathbb{T}_n^+ - k, n, i, a} | \mathcal{H}_{\mathbb{T}_n^+ - k}^{\hat{\mathbf{w}}^g}, \eta_{\mathbb{T}_n^+ - k - h + 1}^\varepsilon, \eta_{\mathbb{T}_n^+ - k - h + 2}^\varepsilon, \dots, \eta_{\mathbb{T}_n^+ - k}^\varepsilon) \\ & | \mathcal{H}_{\mathbb{T}_n^+ - k + 1}^{\hat{\mathbf{w}}^g}, \eta_{\mathbb{T}_n^+ - k - h + 1}^\varepsilon, \eta_{\mathbb{T}_n^+ - k - h + 2}^\varepsilon, \dots, \eta_{\mathbb{T}_n^+ - k}^\varepsilon) \\ & \stackrel{(2)}{=} \hat{w}_{P, \mathbb{T}_n^+ - k, n, j, b}^g \mathbb{E}_P(\varepsilon_{P, \mathbb{T}_n^+ - k, n, i, a} | \mathcal{H}_{\mathbb{T}_n^+ - k}^{\hat{\mathbf{w}}^g}, \eta_{\mathbb{T}_n^+ - k - h + 1}^\varepsilon, \eta_{\mathbb{T}_n^+ - k - h + 2}^\varepsilon, \dots, \eta_{\mathbb{T}_n^+ - k}^\varepsilon) \\ & - \hat{w}_{P, \mathbb{T}_n^+ - k, n, j, b}^g \mathbb{E}_P(\varepsilon_{P, \mathbb{T}_n^+ - k, n, i, a} | \mathcal{H}_{\mathbb{T}_n^+ - k}^{\hat{\mathbf{w}}^g}, \eta_{\mathbb{T}_n^+ - k - h + 1}^\varepsilon, \eta_{\mathbb{T}_n^+ - k - h + 2}^\varepsilon, \dots, \eta_{\mathbb{T}_n^+ - k}^\varepsilon) \\ & \stackrel{(3)}{=} 0, \end{aligned}$$

almost surely, by (1) Assumption 3.3, (2) the tower property and measurability, and (3) subtraction. Also, $\mathbb{E}_P(\|\hat{\mathbf{w}}_{P, \mathbb{T}_n^+ - k, n, h}^{g, \varepsilon} - \hat{\mathbf{w}}_{P, \mathbb{T}_n^+ - k, n, h-1}^{g, \varepsilon}\|_2^2) < \infty$ by the triangle inequality, squaring, linearity of expectation, the Cauchy-Schwarz inequality, and the same arguments as (18) (i.e. linearity of expectation, the contraction property of conditional expectation, and Assumption 3.3).

Next, observe that for each $n \in \mathbb{N}$ and $h \in \mathbb{N}_0$, we have

$$\begin{aligned} & \sup_{P \in \mathcal{P}_{0,n}^*} (\mathbb{E}_P \max_{t \in \mathcal{T}_{n,L}} \|\hat{\mathbf{S}}_{P,t,n,h}^{g, \varepsilon} - \hat{\mathbf{S}}_{P,t,n,h-1}^{g, \varepsilon}\|_2^2)^{\frac{1}{2}} \\ & \stackrel{(1)}{\leq} \sup_{P \in \mathcal{P}_{0,n}^*} (\mathbb{E}_P \|\hat{\mathbf{S}}_{P, \mathbb{T}_n^+, n, h}^{g, \varepsilon} - \hat{\mathbf{S}}_{P, \mathbb{T}_n^+, n, h-1}^{g, \varepsilon}\|_2^2)^{\frac{1}{2}} \\ & + \sup_{P \in \mathcal{P}_{0,n}^*} (\mathbb{E}_P \max_{t \in \mathcal{T}_{n,L}} \|(\hat{\mathbf{S}}_{P, \mathbb{T}_n^+, n, h}^{g, \varepsilon} - \hat{\mathbf{S}}_{P, \mathbb{T}_n^+, n, h-1}^{g, \varepsilon}) - (\hat{\mathbf{S}}_{P,t,n,h}^{g, \varepsilon} - \hat{\mathbf{S}}_{P,t,n,h-1}^{g, \varepsilon})\|_2^2)^{\frac{1}{2}} \\ & \stackrel{(2)}{\leq} \sup_{P \in \mathcal{P}_{0,n}^*} (\mathbb{E}_P \|\hat{\mathbf{S}}_{P, \mathbb{T}_n^+, n, h}^{g, \varepsilon} - \hat{\mathbf{S}}_{P, \mathbb{T}_n^+, n, h-1}^{g, \varepsilon}\|_2^2)^{\frac{1}{2}} \\ & + \sup_{P \in \mathcal{P}_{0,n}^*} \left(\mathbb{E}_P \max_{\ell=0, \dots, \mathbb{T}_n^+ - \mathbb{T}_n^- - L_n} \left\| \sum_{k=0}^{\ell} (\hat{\mathbf{w}}_{P, \mathbb{T}_n^+ - k, n, h}^{g, \varepsilon} - \hat{\mathbf{w}}_{P, \mathbb{T}_n^+ - k, n, h-1}^{g, \varepsilon}) \right\|_2^2 \right)^{\frac{1}{2}} \end{aligned}$$

by (1) adding and subtracting $\hat{\mathbf{S}}_{P, \mathbb{T}_n^+, n, h}^{g, \varepsilon} - \hat{\mathbf{S}}_{P, \mathbb{T}_n^+, n, h-1}^{g, \varepsilon}$ and the triangle inequality, (2) including the “last” term in this reversed partial sum and rewriting as the corresponding martingale. Continuing on from (2), we have

$$\begin{aligned}
& \sup_{P \in \mathcal{P}_{0,n}^*} (\mathbb{E}_P \|\hat{\mathbf{S}}_{P, \mathbb{T}_n^+, n, h}^{g, \varepsilon} - \hat{\mathbf{S}}_{P, \mathbb{T}_n^+, n, h-1}^{g, \varepsilon}\|_2^2)^{\frac{1}{2}} \\
& + \sup_{P \in \mathcal{P}_{0,n}^*} \left(\mathbb{E}_P \max_{\ell=0, \dots, \mathbb{T}_n^+ - \mathbb{T}_n^- - L_n} \left\| \sum_{k=0}^{\ell} (\hat{\mathbf{w}}_{P, \mathbb{T}_n^+ - k, n, h}^{g, \varepsilon} - \hat{\mathbf{w}}_{P, \mathbb{T}_n^+ - k, n, h-1}^{g, \varepsilon}) \right\|_2^2 \right)^{\frac{1}{2}} \\
& \stackrel{(3)}{\leq} 3 \sup_{P \in \mathcal{P}_{0,n}^*} (\mathbb{E}_P \|\hat{\mathbf{S}}_{P, \mathbb{T}_n^+, n, h}^{g, \varepsilon} - \hat{\mathbf{S}}_{P, \mathbb{T}_n^+, n, h-1}^{g, \varepsilon}\|_2^2)^{\frac{1}{2}} \\
& \stackrel{(4)}{\leq} K \sup_{P \in \mathcal{P}_{0,n}^*} \left(\sum_{t \in \mathcal{T}_{n,L}} \mathbb{E}_P \|\hat{\mathbf{w}}_{P, t, n, h}^{g, \varepsilon} - \hat{\mathbf{w}}_{P, t, n, h-1}^{g, \varepsilon}\|_2^2 \right)^{\frac{1}{2}},
\end{aligned}$$

by (3) Doob’s maximal inequality (see e.g. Theorem 1.9 of [Pis16]), and (4) upper bounding by max of partial sums and applying Lemma B.6 with the finite constant $K/3 > 0$. Hence, we have the inequality

$$\begin{aligned}
& \sup_{P \in \mathcal{P}_{0,n}^*} (\mathbb{E}_P \max_{t \in \mathcal{T}_{n,L}} \|\hat{\mathbf{S}}_{P, t, n, h}^{g, \varepsilon} - \hat{\mathbf{S}}_{P, t, n, h-1}^{g, \varepsilon}\|_2^2)^{\frac{1}{2}} \\
& \leq K \sup_{P \in \mathcal{P}_{0,n}^*} \left(\sum_{t \in \mathcal{T}_{n,L}} \mathbb{E}_P \|\hat{\mathbf{w}}_{P, t, n, h}^{g, \varepsilon} - \hat{\mathbf{w}}_{P, t, n, h-1}^{g, \varepsilon}\|_2^2 \right)^{\frac{1}{2}}.
\end{aligned} \tag{20}$$

Observe that for $h = 1, 2, \dots$, we have

$$\begin{aligned}
& \mathbb{E}_P \|\hat{\mathbf{w}}_{P, t, n, h}^{g, \varepsilon} - \hat{\mathbf{w}}_{P, t, n, h-1}^{g, \varepsilon}\|_2^2 \\
& \stackrel{(1)}{=} \sum_{m=(i,j,a,b) \in \mathcal{D}_n} \mathbb{E}_P (|\mathbb{E}_P(\hat{w}_{P, t, n, j, b}^g \varepsilon_{P, t, n, i, a} | \eta_t^\varepsilon, \dots, \eta_{t-h}^\varepsilon, \mathcal{H}_t^{\hat{g}}) \\
& - \mathbb{E}_P(\hat{w}_{P, t, n, j, b}^g \varepsilon_{P, t, n, i, a} | \eta_t^\varepsilon, \dots, \eta_{t-h+1}^\varepsilon, \mathcal{H}_t^{\hat{g}})|^2) \\
& \stackrel{(2)}{=} \sum_{m=(i,j,a,b) \in \mathcal{D}_n} \mathbb{E}_P (|\mathbb{E}_P(\hat{w}_{P, t, n, j, b}^g (\mathbb{E}_P(\varepsilon_{P, t, n, i, a} | \eta_t^\varepsilon, \dots, \eta_{t-h}^\varepsilon, \mathcal{H}_t^{\hat{g}}) \\
& - \mathbb{E}_P(\varepsilon_{P, t, n, i, a} | \eta_t^\varepsilon, \dots, \eta_{t-h+1}^\varepsilon, \mathcal{H}_t^{\hat{g}}))|^2) \\
& \stackrel{(3)}{=} \sum_{m=(i,j,a,b) \in \mathcal{D}_n} \mathbb{E}_P (|\hat{w}_{P, t, n, j, b}^g \mathbb{E}_P[(\mathbb{E}_P(\varepsilon_{P, t, n, i, a} | \eta_{t,a}^\varepsilon, \dots, \eta_{t-h,a}^\varepsilon, \mathcal{H}_t^{\hat{g}}) \\
& - \mathbb{E}_P(\varepsilon_{P, t, n, i, a} | \eta_{t,a}^\varepsilon, \dots, \eta_{t-h+1,a}^\varepsilon, \mathcal{H}_t^{\hat{g}})) | \eta_{t,a}^\varepsilon, \dots, \eta_{t-h,a}^\varepsilon, \mathcal{H}_t^{\hat{g}}]|^2) \\
& \stackrel{(4)}{=} \sum_{m=(i,j,a,b) \in \mathcal{D}_n} \mathbb{E}_P (|\hat{w}_{P, t, n, j, b}^g \mathbb{E}_P[(\mathbb{E}_P(G_{P, t, n, i, a}^\varepsilon(\mathcal{H}_{t,a}^\varepsilon) | \eta_{t,a}^\varepsilon, \dots, \eta_{t-h,a}^\varepsilon, \mathcal{H}_t^{\hat{g}}) \\
& - \mathbb{E}_P(G_{P, t, n, i, a}^\varepsilon(\mathcal{H}_{t,a}^\varepsilon) | \eta_{t,a}^\varepsilon, \dots, \eta_{t-h+1,a}^\varepsilon, \mathcal{H}_t^{\hat{g}})) | \eta_{t,a}^\varepsilon, \dots, \eta_{t-h,a}^\varepsilon, \mathcal{H}_t^{\hat{g}}]|^2) \\
& \stackrel{(5)}{=} \sum_{m=(i,j,a,b) \in \mathcal{D}_n} \mathbb{E}_P (|\hat{w}_{P, t, n, j, b}^g \mathbb{E}_P[(\mathbb{E}_P(G_{P, t, n, i, a}^\varepsilon(\mathcal{H}_{t,a}^\varepsilon) | \eta_{t,a}^\varepsilon, \dots, \eta_{t-h,a}^\varepsilon, \mathcal{H}_t^{\hat{g}}) \\
& - \mathbb{E}_P(G_{P, t, n, i, a}^\varepsilon(\tilde{\mathcal{H}}_{t,a,h}^\varepsilon) | \eta_{t,a}^\varepsilon, \dots, \eta_{t-h,a}^\varepsilon, \mathcal{H}_t^{\hat{g}})) | \eta_{t,a}^\varepsilon, \dots, \eta_{t-h,a}^\varepsilon, \mathcal{H}_t^{\hat{g}}]|^2),
\end{aligned}$$

by (1) rewriting the expression, (2) the causal representation from Assumption 3.3, (3) measurability of the conditional expectations and the linearity property of conditional expectation, (4) the causal representation from Assumption 3.4, and (5) replacing $\eta_{t-h,a}^\varepsilon$ with the iid copy $\tilde{\eta}_{t-h,a}^\varepsilon$. Continuing on

from line (5), we have

$$\begin{aligned}
& \sum_{m=(i,j,a,b) \in \mathcal{D}_n} \mathbb{E}_P(|\hat{w}_{P,t,n,j,b}^g| \mathbb{E}_P[(\mathbb{E}_P(G_{P,t,n,i,a}^\varepsilon(\mathcal{H}_{t,a}^\varepsilon)|\eta_{t,a}^\varepsilon, \dots, \eta_{t-h,a}^\varepsilon, \mathcal{H}_t^{\hat{g}}) \\
& - \mathbb{E}_P(G_{P,t,n,i,a}^\varepsilon(\tilde{\mathcal{H}}_{t,a,h}^\varepsilon)|\eta_{t,a}^\varepsilon, \dots, \eta_{t-h,a}^\varepsilon, \mathcal{H}_t^{\hat{g}}))|\eta_{t,a}^\varepsilon, \dots, \eta_{t-h,a}^\varepsilon, \mathcal{H}_t^{\hat{g}}]|^2) \\
& \stackrel{(6)}{=} \sum_{m=(i,j,a,b) \in \mathcal{D}_n} \mathbb{E}_P(|\hat{w}_{P,t,n,j,b}^g| \mathbb{E}_P[G_{P,t,n,i,a}^\varepsilon(\mathcal{H}_{t,a}^\varepsilon) - G_{P,t,n,i,a}^\varepsilon(\tilde{\mathcal{H}}_{t,a,h}^\varepsilon)|\eta_{t,a}^\varepsilon, \dots, \eta_{t-h,a}^\varepsilon, \mathcal{H}_t^{\hat{g}}]|^2) \\
& \stackrel{(7)}{\leq} D_n \max_{(i,j,a,b) \in \mathcal{D}_n} \mathbb{E}_P(|\hat{w}_{P,t,n,j,b}^g|^2)(\theta_{P,t,n,i,a}^{\varepsilon,\infty}(h))^2 \\
& \stackrel{(8)}{\leq} D_n \max_{(i,j,a,b) \in \mathcal{D}_n} \mathbb{E}_P(|\hat{w}_{P,t,n,j,b}^g|^2)(\bar{\Theta}^\infty(h \vee 1)^{-\bar{\beta}^\infty})^2,
\end{aligned}$$

by (6) measurability and linearity of the conditional expectations, and (7) Hölder's inequality, contraction property of conditional expectation, rewriting as the functional dependence measure from Definition 3.1, and upper bounding by the sum by D_n times the maximum over the dimension/time-offset combinations in \mathcal{D}_n , and (8) the upper bound on the L^∞ functional dependence measure from Assumption 3.5. Similarly, for $h = 0$, we have

$$\begin{aligned}
& \mathbb{E}_P \|\hat{w}_{P,t,n,0}^{g,\varepsilon} - \hat{w}_{P,t,n,-1}^{g,\varepsilon}\|_2^2 \\
& \stackrel{(1)}{=} \mathbb{E}_P \|\hat{w}_{P,t,n,0}^{g,\varepsilon}\|_2^2 \\
& \stackrel{(2)}{=} \sum_{m=(i,j,a,b) \in \mathcal{D}_n} \mathbb{E}_P(|\mathbb{E}_P(\hat{w}_{P,t,n,j,b}^g \varepsilon_{P,t,n,i,a} | \eta_t^\varepsilon, \mathcal{H}_t^{\hat{g}})|^2) \\
& \stackrel{(3)}{=} \sum_{m=(i,j,a,b) \in \mathcal{D}_n} \mathbb{E}_P(|\hat{w}_{P,t,n,j,b}^g| \mathbb{E}_P(\varepsilon_{P,t,n,i,a} | \eta_t^\varepsilon, \mathcal{H}_t^{\hat{g}})|^2) \\
& \stackrel{(4)}{\leq} D_n \max_{(i,j,a,b) \in \mathcal{D}_n} \mathbb{E}_P(|\hat{w}_{P,t,n,j,b}^g|^2)(\bar{\Theta}^\infty)^2,
\end{aligned}$$

because (1) $\hat{w}_{P,t,n,-1}^{g,\varepsilon} = 0$, (2) rewriting the expression, (3) Assumption 3.3, and (4) Hölder's inequality, contraction property of conditional expectation, applying the upper bound on the L^∞ norm from Assumption 3.5, and upper bounding by the sum by D_n times the maximum over the dimension/time-offset combinations in \mathcal{D}_n . Hence, for all $h \in \mathbb{N}_0$ we have

$$\mathbb{E}_P \|\hat{w}_{P,t,n,h}^{g,\varepsilon} - \hat{w}_{P,t,n,h-1}^{g,\varepsilon}\|_2^2 \leq D_n \max_{(i,j,a,b) \in \mathcal{D}_n} \mathbb{E}_P(|\hat{w}_{P,t,n,j,b}^g|^2)(\bar{\Theta}^\infty(h \vee 1)^{-\bar{\beta}^\infty})^2. \quad (21)$$

Summing over $h \in \mathbb{N}_0$, we have

$$\begin{aligned}
& \sup_{P \in \mathcal{P}_{0,n}^*} (\mathbb{E}_P \max_{t \in \mathcal{T}_{n,L}} \|\hat{S}_{P,t,n}^{g,\varepsilon}\|_2^2)^{\frac{1}{2}} \\
& \stackrel{(1)}{\leq} \sum_{h=0}^{\infty} \sup_{P \in \mathcal{P}_{0,n}^*} (\mathbb{E}_P \max_{t \in \mathcal{T}_{n,L}} \|\hat{S}_{P,t,n,h}^{g,\varepsilon} - \hat{S}_{P,t,n,h-1}^{g,\varepsilon}\|_2^2)^{\frac{1}{2}} \\
& \stackrel{(2)}{\leq} \sum_{h=0}^{\infty} K \sup_{P \in \mathcal{P}_{0,n}^*} \left(\sum_{t \in \mathcal{T}_{n,L}} \mathbb{E}_P \|\hat{w}_{P,t,n,h}^{g,\varepsilon} - \hat{w}_{P,t,n,h-1}^{g,\varepsilon}\|_2^2 \right)^{\frac{1}{2}} \\
& \stackrel{(3)}{\leq} \sum_{h=0}^{\infty} K \sup_{P \in \mathcal{P}_{0,n}^*} \left(\sum_{t \in \mathcal{T}_{n,L}} D_n \max_{j \in [d_Y]} \max_{b \in B_j} \mathbb{E}_P(|\hat{w}_{P,t,n,j,b}^g|^2)(\bar{\Theta}^\infty(h \vee 1)^{-\bar{\beta}^\infty})^2 \right)^{\frac{1}{2}},
\end{aligned}$$

by (1) the telescoping argument from (19) and the triangle inequality, (2) applying the inequality (20),

and (3) applying the inequality (21). Continuing on from line (3), we have

$$\begin{aligned}
& \sum_{h=0}^{\infty} K \sup_{P \in \mathcal{P}_{0,n}^*} \left(\sum_{t \in \mathcal{T}_{n,L}} D_n \max_{j \in [d_Y]} \max_{b \in B_j} \mathbb{E}_P(|\hat{w}_{P,t,n,j,b}^g|^2) (\bar{\Theta}^\infty (h \vee 1)^{-\bar{\beta}^\infty})^2 \right)^{\frac{1}{2}} \\
& \stackrel{(4)}{\leq} \sum_{h=0}^{\infty} K \sup_{P \in \mathcal{P}_{0,n}^*} (T_{n,L} D_n \max_{t \in \mathcal{T}_{n,L}} \max_{j \in [d_Y]} \max_{b \in B_j} \mathbb{E}_P(|\hat{w}_{P,t,n,j,b}^g|^2) (\bar{\Theta}^\infty (h \vee 1)^{-\bar{\beta}^\infty})^2)^{\frac{1}{2}} \\
& \stackrel{(5)}{\leq} \bar{\Theta}^\infty K T_{n,L}^{\frac{1}{2}} D_n^{\frac{1}{2}} \sup_{P \in \mathcal{P}_{0,n}^*} \max_{t \in \mathcal{T}_{n,L}} \max_{j \in [d_Y]} \max_{b \in B_j} \mathbb{E}_P(|\hat{w}_{P,t,n,j,b}^g|^2)^{\frac{1}{2}} \sum_{h=0}^{\infty} (h \vee 1)^{-\bar{\beta}^\infty} \\
& \stackrel{(6)}{\leq} \bar{\Theta}^\infty \bar{K}^\infty K T_{n,L}^{\frac{1}{2}} D_n^{\frac{1}{2}} \sup_{P \in \mathcal{P}_{0,n}^*} \max_{t \in \mathcal{T}_{n,L}} \max_{j \in [d_Y]} \max_{b \in B_j} \mathbb{E}_P(|\hat{w}_{P,t,n,j,b}^g|^2)^{\frac{1}{2}} \\
& \stackrel{(7)}{\leq} \bar{K} T_{n,L}^{\frac{1}{2}} D_n^{\frac{1}{2}} \sup_{P \in \mathcal{P}_{0,n}^*} \max_{t \in \mathcal{T}_{n,L}} \max_{j \in [d_Y]} \max_{b \in B_j} \mathbb{E}_P(|\hat{w}_{P,t,n,j,b}^g|^2)^{\frac{1}{2}},
\end{aligned}$$

by (4) upper bounding each term by the maximum over time t , (5) simplifying the expression, (6) writing $\bar{K}^\infty = \sum_{h=0}^{\infty} (h \vee 1)^{-\bar{\beta}^\infty} < \infty$ since $\bar{\beta}^\infty > 1$ by upper Assumption 3.5, and (7) grouping together the positive constants into the positive constant \bar{K} .

Step 1.3: The same arguments as Step 1.2 (i.e. exchanging g, ε with f, ξ) can be used to show that for $n \in \mathbb{N}$ and $\delta > 0$ we have

$$\sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{P}_P(\tau_n^{-1} S_{n,p}(\hat{\mathbf{w}}_{P,n}^{f,\xi}) > \delta) = o(1).$$

Step 2 (Strong Gaussian Approximation): Next, we turn to the products of errors $(\mathbf{R}_{P,t,n})_{t \in \mathcal{T}_{n,L}}$. Denote the Gaussian random vectors associated with the strong Gaussian approximation of the product of errors by $\mathbf{R}_{t,n}^\dagger \sim \mathcal{N}(0, \Sigma_{P,t,n}^{\mathbf{R}})$ for $t \in \mathcal{T}_{n,L}$. Observe that

$$\begin{aligned}
& \sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{P}_P(S_{n,p}(\mathbf{R}_{P,n}) > \hat{q}_{1-\alpha+\nu_n} + \frac{\tau_n}{2}) \\
& \stackrel{(1)}{\leq} \sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{P}_P(S_{n,p}(\mathbf{R}_n^\dagger) > \hat{q}_{1-\alpha+\nu_n} + \frac{\tau_n}{4}) \\
& + \sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{P}_P \left(\max_{s \in \mathcal{T}_{n,L}} \left\| \frac{1}{\sqrt{T_{n,L}}} \sum_{t \leq s} (\mathbf{R}_{P,t,n} - \mathbf{R}_{t,n}^\dagger) \right\|_2 > \frac{\tau_n}{4} \right) \\
& \stackrel{(2)}{\leq} \sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{P}_P(S_{n,p}(\mathbf{R}_n^\dagger) > \hat{q}_{1-\alpha+\nu_n} + \frac{\tau_n}{4}) \\
& + 4\tau_n^{-1} \sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{E}_P \left(\max_{s \in \mathcal{T}_{n,L}} \left\| \frac{1}{\sqrt{T_{n,L}}} \sum_{t \leq s} (\mathbf{R}_{P,t,n} - \mathbf{R}_{t,n}^\dagger) \right\|_2 \right) \\
& \stackrel{(3)}{\leq} \sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{P}_P(S_{n,p}(\mathbf{R}_n^\dagger) > \hat{q}_{1-\alpha+\nu_n} + \frac{\tau_n}{4}) \\
& + 4\tau_n^{-1} K D_n^{\frac{1}{2}} \bar{\Theta}^R (\bar{\Gamma}_n^R)^{\frac{1}{2} \frac{\bar{\beta}^R - 2}{\bar{\beta}^R - 1}} \sqrt{\log(T_{n,L})} \left(\frac{D_n}{T_{n,L}} \right)^{\xi(\bar{q}^R, \bar{\beta}^R)},
\end{aligned}$$

where (1) follows from the triangle inequality, subadditivity, and the assumption about the form of the test statistic, (2) follows by Markov's inequality, and (3) follows by the distribution-uniform strong Gaussian approximation for high-dimensional nonstationary processes from Lemma B.1. By

subadditivity and monotonicity, we have

$$\begin{aligned}
& \sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{P}_P(S_{n,p}(\mathbf{R}_n^\dagger) > \hat{q}_{1-\alpha+\nu_n} + \frac{\tau_n}{4}) \\
& \leq \sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{P}_P(S_{n,p}(\mathbf{R}_n^\dagger) > q_{1-\alpha}) \\
& + \sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{P}_P(q_{1-\alpha} > \hat{q}_{1-\alpha+\nu_n} + \frac{\tau_n}{4}) \\
& = \alpha + \sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{P}_P(q_{1-\alpha} > \hat{q}_{1-\alpha+\nu_n} + \frac{\tau_n}{4}).
\end{aligned}$$

Step 3 (Covariance Approximation): Now, we focus on upper bounding

$$\sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{P}_P(q_{1-\alpha} > \hat{q}_{1-\alpha+\nu_n} + \frac{\tau_n}{4}).$$

Step 3.1: Let us reflect on the implications of Proposition 4.2 of Mies and Steland [MS23], which is the distribution-pointwise version of Lemma B.4. Proposition 4.2 states that for each $n \in \mathbb{N}$ and $P \in \mathcal{P}_{0,n}^*$, for some cumulative covariance $\bar{Q}_{P,n}^{\mathbf{R}}$, there exist *independent* Gaussian random vectors $\bar{\mathbf{R}}_{t,n} \sim \mathcal{N}(0, \bar{\Sigma}_{P,t,n}^{\mathbf{R}})$ for $t \in \mathcal{T}_{n,L}$ with $\bar{\Sigma}_{P,t,n}^{\mathbf{R}} = \bar{Q}_{P,t,n}^{\mathbf{R}} - \bar{Q}_{P,t-1,n}^{\mathbf{R}}$ that are coupled with the Gaussian random vectors from the strong Gaussian approximation of the product of errors $\mathbf{R}_{t,n}^\dagger \sim \mathcal{N}(0, \Sigma_{P,t,n}^{\mathbf{R}})$ for $t \in \mathcal{T}_{n,L}$, such that

$$\mathbb{E}_P \max_{k \in \mathcal{T}_{n,L}} \left\| \sum_{t \leq k} \mathbf{R}_{t,n}^\dagger - \sum_{t \leq k} \bar{\mathbf{R}}_{t,n} \right\|_2^2 \leq K \log(T_{n,L}) [\sqrt{T_{n,L} \bar{\delta}_{P,n} \rho_{P,n}} + \rho_{P,n}] = \bar{\Delta}_{P,n},$$

where

$$\bar{\delta}_{P,n} = \max_{k \in \mathcal{T}_{n,L}} \left\| \sum_{t \leq k} \Sigma_{P,t,n}^{\mathbf{R}} - \sum_{t \leq k} \bar{\Sigma}_{P,t,n}^{\mathbf{R}} \right\|_{\text{tr}}$$

and

$$\rho_{P,n} = \max_{t \in \mathcal{T}_{n,L}} \|\Sigma_{P,t,n}^{\mathbf{R}}\|_{\text{tr}}.$$

Let $\bar{\mathbf{R}}_n = (\bar{\mathbf{R}}_{t,n})_{t \in \mathcal{T}_{n,L}}$ and denote the $(1 - \alpha)$ quantile of $S_{n,p}(\bar{\mathbf{R}}_n)$ by $\bar{q}_{1-\alpha}$. For each $n \in \mathbb{N}$ and $P \in \mathcal{P}_{0,n}^*$, we have

$$\begin{aligned}
& \mathbb{P}_P(S_{n,p}(\mathbf{R}_n^\dagger) > \bar{q}_{1-\alpha+\nu_n} + \frac{\tau_n}{4}) \\
& \stackrel{(1)}{\leq} \mathbb{P}_P(S_{n,p}(\bar{\mathbf{R}}_n) > \bar{q}_{1-\alpha+\nu_n}) \\
& + \mathbb{P}_P \left(\max_{s \in \mathcal{T}_{n,L}} \left\| \frac{1}{\sqrt{T_{n,L}}} \sum_{t \leq s} (\mathbf{R}_{t,n}^\dagger - \bar{\mathbf{R}}_{t,n}) \right\|_2 > \frac{\tau_n}{4} \right) \\
& \stackrel{(2)}{=} \mathbb{P}_P(S_{n,p}(\bar{\mathbf{R}}_n) > \bar{q}_{1-\alpha+\nu_n}) \\
& + \mathbb{P}_P \left(\max_{s \in \mathcal{T}_{n,L}} \left\| \frac{1}{\sqrt{T_{n,L}}} \sum_{t \leq s} (\mathbf{R}_{t,n}^\dagger - \bar{\mathbf{R}}_{t,n}) \right\|_2^2 > \frac{\tau_n^2}{16} \right) \\
& \stackrel{(3)}{\leq} \mathbb{P}_P(S_{n,p}(\bar{\mathbf{R}}_n) > \bar{q}_{1-\alpha+\nu_n}) \\
& + 16\tau_n^{-2} T_{n,L}^{-1} \mathbb{E}_P \left(\max_{s \in \mathcal{T}_{n,L}} \left\| \sum_{t \leq s} (\mathbf{R}_{t,n}^\dagger - \bar{\mathbf{R}}_{t,n}) \right\|_2^2 \right) \\
& \stackrel{(4)}{\leq} (\alpha - \nu_n) + 16\tau_n^{-2} \bar{\Delta}_{P,n} T_{n,L}^{-1} \stackrel{(5)}{=} \alpha + \left[16\tau_n^{-2} \bar{\Delta}_{P,n} T_{n,L}^{-1} - \nu_n \right],
\end{aligned}$$

where the previous lines follow by (1) the triangle inequality, subadditivity, the assumption about the form of the test statistic, (2) squaring, (3) Markov's inequality, (4) Proposition 4.2 from Mies and Steland [MS23], and (5) rearranging terms. We see that if

$$\left[16\tau_n^{-2}\bar{\Delta}_{P,n}T_{n,L}^{-1} - \nu_n\right] < 0,$$

then

$$\mathbb{P}_P(S_{n,p}(\mathbf{R}_n^\dagger) > \bar{q}_{1-\alpha+\nu_n} + \frac{\tau_n}{4}) < \alpha,$$

which implies that $\bar{q}_{1-\alpha+\nu_n} + \frac{\tau_n}{4}$ is greater than $q_{1-\alpha}^\dagger$, the $(1-\alpha)$ quantile of $S_{n,p}(\mathbf{R}_n^\dagger)$. Hence, if

$$q_{1-\alpha}^\dagger \geq \bar{q}_{1-\alpha+\nu_n} + \frac{\tau_n}{4},$$

then

$$\left[16\tau_n^{-2}\bar{\Delta}_{P,n}T_{n,L}^{-1} - \nu_n\right] \geq 0,$$

or equivalently

$$\bar{\Delta}_{P,n} \geq \frac{1}{16}T_{n,L}\nu_n\tau_n^2.$$

Step 3.2: Now, we apply this idea with the cumulative covariance of the residual products \hat{Q}_n^R . By the implication stated at the end of Step 3.1 and monotonicity, we have

$$\begin{aligned} & \sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{P}_P(q_{1-\alpha} > \hat{q}_{1-\alpha+\nu_n} + \frac{\tau_n}{4}) \\ & \leq \sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{P}_P(\hat{\Delta}_{P,n} \geq \frac{1}{16}T_{n,L}\nu_n\tau_n^2), \end{aligned}$$

where we have replaced $\bar{\Delta}_{P,n}$, $\bar{\delta}_{P,n}$ with $\hat{\Delta}_{P,n}$, $\hat{\delta}_{P,n}$ which are defined by

$$\hat{\Delta}_{P,n} = K \log(T_{n,L}) \left[\sqrt{T_{n,L}\hat{\delta}_{P,n}\rho_{P,n} + \rho_{P,n}} \right],$$

$$\hat{\delta}_{P,n} = \max_{k \in \mathcal{T}_{n,L}} \left\| \sum_{t \leq k} \Sigma_{P,t,n}^R - \hat{Q}_{k,n}^R \right\|_{\text{tr}},$$

and $\rho_{P,n}$ is defined in the same way as

$$\rho_{P,n} = \max_{t \in \mathcal{T}_{n,L}} \|\Sigma_{P,t,n}^R\|_{\text{tr}}.$$

Thus, if we can find φ_n such that $\hat{\Delta}_{P,n} = O_{\mathcal{P}}(\varphi_n)$ and if we select the offsets so that $\nu_n\tau_n^2 \gg T_{n,L}^{-1}\varphi_n$, or equivalently $\nu_n \gg \tau_n^{-2}T_{n,L}^{-1}\varphi_n$, then we will have

$$\sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{P}_P(\hat{\Delta}_{P,n} \geq \frac{1}{16}T_{n,L}\nu_n\tau_n^2) = o(1).$$

By Lemma B.5 and Assumption 3.5, we have

$$\sup_{P \in \mathcal{P}_{0,n}^*} \rho_{P,n} \leq K_\rho D_n (\bar{\Theta}^R)^2,$$

for some constant $K_\rho > 0$, so we obtain $\hat{\Delta}_{P,n} = O_{\mathcal{P}}(\varphi_n)$ with

$$\varphi_n = \log(T_{n,L})D_n \left[T_{n,L}^{\frac{1}{2}}D_n^{-\frac{1}{2}}\hat{\delta}_{P,n}^{\frac{1}{2}} + 1 \right].$$

Plugging φ_n into the offset condition $\nu_n \gg \tau_n^{-2}T_{n,L}^{-1}\varphi_n$ that we wish to satisfy, if we have

$$\nu_n \gg \log(T_{n,L})D_n(\tau_n^{-2}(T_{n,L}^{-\frac{1}{2}}D_n^{-\frac{1}{2}}\hat{\delta}_{P,n}^{\frac{1}{2}} + T_{n,L}^{-1})),$$

then

$$\sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{P}_P(\hat{\Delta}_{P,n} \geq \frac{1}{16} T_{n,L} \nu_n \tau_n^2) = o(1).$$

Step 3.3: It remains to analyze $\hat{\delta}_{P,n}$. By the triangle inequality, we have

$$\begin{aligned} \hat{\delta}_{P,n} &= \max_{k \in \mathcal{T}_{n,L}} \left\| \sum_{t \leq k} \Sigma_{P,t,n}^R - \hat{Q}_{k,n}^R \right\|_{\text{tr}} \\ &\leq \max_{k \in \mathcal{T}_{n,L}} \left\| \sum_{t \leq k} \Sigma_{P,t,n}^R - Q_{P,k,n}^R \right\|_{\text{tr}} \\ &\quad + \max_{k \in \mathcal{T}_{n,L}} \|\hat{Q}_{k,n}^R - Q_{P,k,n}^R\|_{\text{tr}}. \end{aligned}$$

By Lemma B.3, Assumption 3.5, and Assumption 3.6, the covariance estimation error can be bounded as

$$\begin{aligned} &\sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{E}_P \left(\max_{k \in \mathcal{T}_{n,L}} \left\| \sum_{t \leq k} \Sigma_{P,t,n}^R - Q_{P,k,n}^R \right\|_{\text{tr}} \right) \\ &\leq K(\bar{\Theta}^R)^2 D_n (\bar{\Gamma}_n^R L_n^{\frac{1}{2}} + T_{n,L}^{\frac{1}{2}} D_n^{\frac{1}{2}} L_n^{\frac{1}{2}} + T_{n,L} L_n^{-1} + T_{n,L} L_n^{2-\bar{\beta}^R}) \\ &= O(r_{n,1}^\delta), \end{aligned}$$

where

$$r_{n,1}^\delta = D_n (\bar{\Gamma}_n^R L_n^{\frac{1}{2}} + T_{n,L}^{\frac{1}{2}} D_n^{\frac{1}{2}} L_n^{\frac{1}{2}} + T_{n,L} L_n^{-1} + T_{n,L} L_n^{2-\bar{\beta}^R}).$$

Next, we must handle the prediction errors due using the residual products instead of the error products. For any $\epsilon > 0$, we have

$$\begin{aligned} &\sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{E}_P \left(\max_{k \in \mathcal{T}_{n,L}} \|\hat{Q}_{k,n}^R - Q_{P,k,n}^R\|_{\text{tr}} \wedge \epsilon \right) \\ &\stackrel{(1)}{=} \sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{E}_P \left(\max_{k \in \mathcal{T}_{n,L}} \left\| \frac{1}{L_n} \sum_{r=L_n+\mathbb{T}_n^- - 1}^k \left[\left(\sum_{s=r-L_n+1}^r \hat{R}_{s,n} \right)^{\otimes 2} - \left(\sum_{s=r-L_n+1}^r R_{P,s,n} \right)^{\otimes 2} \right] \right\|_{\text{tr}} \wedge \epsilon \right) \\ &\stackrel{(2)}{\leq} \frac{1}{L_n} \sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{E}_P \left(\left[\sum_{r \in \mathcal{T}_{n,L}} \left\| \left(\sum_{s=r-L_n+1}^r \hat{R}_{s,n} \right)^{\otimes 2} - \left(\sum_{s=r-L_n+1}^r R_{P,s,n} \right)^{\otimes 2} \right\|_{\text{tr}} \right] \wedge \epsilon \right) \\ &\stackrel{(3)}{\leq} \frac{2}{L_n} \sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{E}_P \left(\left[\sum_{r \in \mathcal{T}_{n,L}} \left(\left\| \sum_{s=r-L_n+1}^r (\hat{R}_{s,n} - R_{P,s,n}) \right\|_2 \left\| \sum_{s=r-L_n+1}^r R_{P,s,n} \right\|_2 \right. \right. \right. \\ &\quad \left. \left. \left. + \left\| \sum_{s=r-L_n+1}^r (\hat{R}_{s,n} - R_{P,s,n}) \right\|_2^2 \right) \right] \wedge \epsilon \right), \end{aligned}$$

where (1) is from the definitions of $Q_{P,k,n}^R$, $\hat{Q}_{k,n}^R$, (2) is from the triangle inequality, and (3) is from the following outer product inequality for vectors $\hat{v}, v \in \mathbb{R}^d$

$$\begin{aligned} &\|\hat{v}\hat{v}^\top - vv^\top\|_{\text{tr}} \\ &\stackrel{(1)}{=} \|(\hat{v} - v)v^\top + v(\hat{v} - v)^\top + (\hat{v} - v)(\hat{v} - v)^\top\|_{\text{tr}} \\ &\stackrel{(2)}{\leq} 2\|(\hat{v} - v)v^\top\|_{\text{tr}} + \|(\hat{v} - v)(\hat{v} - v)^\top\|_{\text{tr}} \\ &\stackrel{(3)}{=} 2\|\hat{v} - v\|_2 \|v\|_2 + \|\hat{v} - v\|_2^2, \end{aligned}$$

where (1) follows from adding and subtracting terms, (2) follows from the triangle inequality, and (3) follows by the properties of outer products and the definition of the trace norm. For each $r \in \mathcal{T}_{n,L}$, we have the following decomposition into the three bias terms from Step 1 by the triangle inequality

$$\begin{aligned} & \left\| \sum_{s=r-L_n+1}^r \left(\hat{\mathbf{R}}_{s,n} - \mathbf{R}_{P,s,n} \right) \right\|_2 \\ & \leq \left\| \sum_{s=r-L_n+1}^r \hat{\mathbf{w}}_{P,s,n}^{f,g}(\mathbf{Z}_{s,n}) \right\|_2 + \left\| \sum_{s=r-L_n+1}^r \hat{\mathbf{w}}_{P,s,n}^{g,\varepsilon}(\mathbf{Z}_{s,n}) \right\|_2 + \left\| \sum_{s=r-L_n+1}^r \hat{\mathbf{w}}_{P,s,n}^{f,\xi}(\mathbf{Z}_{s,n}) \right\|_2. \end{aligned}$$

Observe that for any $\delta > 0$ and any $r \in \mathcal{T}_{n,L}$, we have

$$\begin{aligned} & \sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{P}_P \left(\left\| \sum_{s=r-L_n+1}^r \hat{\mathbf{w}}_{P,s,n}^{f,g}(\mathbf{Z}_{s,n}) \right\|_2 > \delta L_n^{\frac{1}{2}} \tau_n^7 D_n^{-2} \right) \\ & \leq \delta^{-1} L_n^{-\frac{1}{2}} \tau_n^{-7} D_n^2 \sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{E}_P \left(\left\| \sum_{s=r-L_n+1}^r \hat{\mathbf{w}}_{P,s,n}^{f,g}(\mathbf{Z}_{s,n}) \right\|_2^2 \right) \\ & \leq \delta^{-1} L_n^{\frac{1}{2}} \tau_n^{-7} D_n^3 \sup_{P \in \mathcal{P}_{0,n}^*} \max_{(i,j,a,b) \in \mathcal{D}_n} \mathbb{E}_P(|\hat{w}_{P,t,n,i,a}^f|^2)^{\frac{1}{2}} \mathbb{E}_P(|\hat{w}_{P,t,n,j,b}^g|^2)^{\frac{1}{2}} \\ & = o(1), \end{aligned}$$

using the same arguments as Step 1.1 replacing $T_{n,L}$ with L_n , and noting that $D_n = O(T_n^{\frac{1}{6}})$ which corresponds to a lag-window size of $L_n = O(T_n^{\frac{1}{3-\delta'}})$ for any $\delta' > 0$. Next, for any $\delta > 0$ and any $r \in \mathcal{T}_{n,L}$, we have

$$\begin{aligned} & \sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{P}_P \left(\left\| \sum_{s=r-L_n+1}^r \hat{\mathbf{w}}_{P,s,n}^{g,\varepsilon}(\mathbf{Z}_{s,n}) \right\|_2 > \delta L_n^{\frac{1}{2}} D_n^{-2} \tau_n^7 \right) \\ & = \sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{P}_P \left(\left\| \sum_{s=r-L_n+1}^r \hat{\mathbf{w}}_{P,s,n}^{g,\varepsilon}(\mathbf{Z}_{s,n}) \right\|_2^2 > \delta^2 L_n D_n^{-4} \tau_n^{14} \right) \\ & \leq \delta^{-2} L_n^{-1} D_n^4 \tau_n^{-14} \sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{E}_P \left(\left\| \sum_{s=r-L_n+1}^r \hat{\mathbf{w}}_{P,s,n}^{g,\varepsilon}(\mathbf{Z}_{s,n}) \right\|_2^2 \right) \\ & \leq \delta^{-2} D_n^5 \tau_n^{-14} \bar{K}^2 \sup_{P \in \mathcal{P}_{0,n}^*} \max_{t \in \mathcal{T}_{n,L}} \max_{j \in [d_Y]} \max_{b \in B_j} \mathbb{E}_P(|\hat{w}_{P,t,n,j,b}^g|^2) \\ & = o(1), \end{aligned}$$

for some $\bar{K} > 0$ using the same arguments as Step 1.2 replacing $T_{n,L}$ with L_n . The same arguments as Step 1.2 (i.e. exchanging g, ε with f, ξ) can be used to show that

$$\left\| \sum_{s=r-L_n+1}^r \hat{\mathbf{w}}_{P,s,n}^{f,\xi}(\mathbf{Z}_{s,n}) \right\|_2 = o_P(L_n^{\frac{1}{2}} D_n^{-2} \tau_n^7).$$

Hence, for any $r \in \mathcal{T}_{n,L}$ we have

$$\left\| \sum_{s=r-L_n+1}^r \left(\hat{\mathbf{R}}_{s,n} - \mathbf{R}_{P,s,n} \right) \right\|_2 = o_P(L_n^{\frac{1}{2}} D_n^{-2} \tau_n^7).$$

By Lemma B.2, we have for all $r \in \mathcal{T}_{n,L}$ that

$$\begin{aligned}
& \sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{P}_P \left(\left\| \sum_{s=r-L_n+1}^r \mathbf{R}_{P,s,n} \right\|_2 > L_n^{\frac{1}{2}} D_n^{\frac{1}{2}} \epsilon \right) \\
&= \sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{P}_P \left(\left\| \sum_{s=r-L_n+1}^r \mathbf{R}_{P,s,n} \right\|_2^2 > L_n D_n \epsilon^2 \right) \\
&\leq L_n^{-1} D_n^{-1} \epsilon^{-2} \sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{E}_P \left(\left\| \sum_{s=r-L_n+1}^r \mathbf{R}_{P,s,n} \right\|_2^2 \right) \\
&\leq L_n^{-1} D_n^{-1} \epsilon^{-2} (2 L_n^{\frac{1}{2}} D_n^{\frac{1}{2}} \bar{\Theta}^R K \sum_{h=1}^{\infty} h^{-\bar{\beta}^R})^2 \\
&= O(1),
\end{aligned}$$

where $\sum_{h=1}^{\infty} h^{-\bar{\beta}^R} < \infty$ since $\bar{\beta}^R > 1$ by Assumption 3.5. Therefore, by Markov's inequality, bounded convergence (Lemma B.7), and noting that the previous statements hold for all times in $\mathcal{T}_{n,L}$, we have

$$\max_{k \in \mathcal{T}_{n,L}} \|\hat{Q}_{k,n}^{\mathbf{R}} - Q_{P,k,n}^{\mathbf{R}}\|_{\text{tr}} = O_{\mathcal{P}}(r_{n,2}^{\delta}),$$

where

$$r_{n,2}^{\delta} = T_{n,L} \tau_n^7 D_n^{-\frac{3}{2}} + T_{n,L} D_n^{-4} \tau_n^{14}.$$

Putting it all together, by the triangle inequality, Markov's inequality, and the covariance approximation results above, we have that

$$\hat{\delta}_{P,n} = O_{\mathcal{P}}(r_{n,1}^{\delta} + r_{n,2}^{\delta}),$$

where

$$\begin{aligned}
r_{n,1}^{\delta} &= D_n (\bar{\Gamma}_n^R L_n^{\frac{1}{2}} + T_{n,L}^{\frac{1}{2}} D_n^{\frac{1}{2}} L_n^{\frac{1}{2}} + T_{n,L} L_n^{-1} + T_{n,L} L_n^{2-\bar{\beta}^R}), \\
r_{n,2}^{\delta} &= T_{n,L} \tau_n^7 D_n^{-\frac{3}{2}} + T_{n,L} D_n^{-4} \tau_n^{14}.
\end{aligned}$$

Next, recall the offset condition

$$\nu_n \gg \log(T_{n,L}) D_n (\tau_n^{-2} (T_{n,L}^{-\frac{1}{2}} D_n^{-\frac{1}{2}} \hat{\delta}_{P,n}^{\frac{1}{2}} + T_{n,L}^{-1})).$$

Observe that

$$\begin{aligned}
& T_{n,L}^{-\frac{1}{2}} D_n^{-\frac{1}{2}} (r_{n,1}^{\delta})^{\frac{1}{2}} + T_{n,L}^{-1} \\
&\leq T_{n,L}^{-\frac{1}{2}} D_n^{-\frac{1}{2}} (D_n^{\frac{1}{2}} ((\bar{\Gamma}_n^R)^{\frac{1}{2}} L_n^{\frac{1}{4}} + T_{n,L}^{\frac{1}{4}} D_n^{\frac{1}{4}} L_n^{\frac{1}{4}} + T_{n,L}^{\frac{1}{2}} L_n^{-\frac{1}{2}} + T_{n,L}^{\frac{1}{2}} L_n^{1-\frac{\bar{\beta}^R}{2}})) + T_{n,L}^{-1} \\
&= T_{n,L}^{-\frac{1}{2}} (\bar{\Gamma}_n^R)^{\frac{1}{2}} L_n^{\frac{1}{4}} + T_{n,L}^{-\frac{1}{4}} D_n^{\frac{1}{4}} L_n^{\frac{1}{4}} + L_n^{-\frac{1}{2}} + L_n^{1-\frac{\bar{\beta}^R}{2}} + T_{n,L}^{-1} \\
&= \varphi_{n,1},
\end{aligned}$$

which comes from the covariance estimation error. Also, we have

$$\begin{aligned}
& T_{n,L}^{-\frac{1}{2}} D_n^{-\frac{1}{2}} (r_{n,2}^{\delta})^{\frac{1}{2}} \\
&\leq T_{n,L}^{-\frac{1}{2}} D_n^{-\frac{1}{2}} (T_{n,L}^{\frac{1}{2}} \tau_n^{\frac{7}{2}} D_n^{-\frac{3}{4}} + T_{n,L}^{\frac{1}{2}} \tau_n^7 D_n^{-2}) \\
&= \tau_n^{\frac{7}{2}} D_n^{-\frac{5}{4}} + \tau_n^7 D_n^{-\frac{5}{2}} \\
&= \varphi_{n,2},
\end{aligned}$$

which comes from the prediction errors since we use the residual products instead of the error products. The assumption on the offset condition (13) implies that

$$\nu_n \gg \log(T_{n,L}) D_n (\tau_n^{-2} (\varphi_{n,1} + \varphi_{n,2})),$$

and therefore

$$\sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{P}_P(\hat{\Delta}_{P,n} \geq \frac{1}{16} T_{n,L} \nu_n \tau_n^2) = o(1).$$

Combining the results from Step 1, Step 2, and Step 3, we obtain the final result

$$\limsup_{n \rightarrow \infty} \sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{P}_P(S_{n,p}(\hat{\mathbf{R}}_n) > \hat{q}_{1-\alpha+\nu_n} + \tau_n) \leq \alpha.$$

□

A.2 Proof of Theorem 4.1

It suffices to establish the following two points. First, that the assumptions of Theorem 4.1 imply those of Theorem 3.1. Second, that the sieve time-varying regression estimators, under the setup of Theorem 4.1, satisfy the convergence rate requirements of Theorem 3.1.

By using the following notation, we see that Assumption 4.1 implies Assumption 3.1 and Assumption 4.4 implies Assumption 3.4. Note that the causal representations for the observed processes and error processes from Assumptions 4.1 and 4.4 are defined for all rescaled times \mathcal{U}_n , and therefore for all $\{t/n\}_{t \in \mathcal{T}_n} \subset \mathcal{U}_n$ in particular. For a generic high-dimensional locally stationary observed process $W \in \{X, Y, Z\}$ and any time t , sample size n , dimension l , and time-offset d , we write

$$G_{t,n}^W(\cdot) = \tilde{G}_n^W(t/n, \cdot), \quad G_{t,n,l}^W(\cdot) = \tilde{G}_{n,l}^W(t/n, \cdot), \quad G_{t,n,l,d}^W(\cdot) = \tilde{G}_{n,l,d}^W(t/n, \cdot),$$

to respectively denote the causal representations of all dimensions of the process W , dimension l of the process W , and dimension l of the process W with time-offset d . For a generic high-dimensional locally stationary error process $e \in \{\varepsilon, \xi\}$ and any distribution P , time t , sample size n , dimension l , and time-offset d , we denote

$$G_{P,t,n}^e(\cdot) = \tilde{G}_{P,n}^e(t/n, \cdot), \quad G_{P,t,n,l}^e(\cdot) = \tilde{G}_{P,n,l}^e(t/n, \cdot), \quad G_{P,t,n,l,d}^e(\cdot) = \tilde{G}_{P,n,l,d}^e(t/n, \cdot),$$

to respectively denote the causal representations of all dimensions of the error process e , dimension l of the error process e , and dimension l of the error process e with time-offset d . The causal representations of the error products are defined similarly. Using this notation, we see that Assumption 4.5 implies Assumption 3.5 and Assumption 4.6 implies Assumption 3.6. Specifically, Assumption 3.6 is satisfied with $\bar{\Gamma}_n^R = D_n^{\frac{1}{2}}$ by using linearity of expectation and directly applying the stochastic Lipschitz condition for the product of errors from the discussion below Assumption 4.6 to each term in the sum.

It remains to show that Assumptions 3.2 and 3.3 are implied. To see this, let us consider the following notation. For any distribution P , time t , sample size n , dimensions i, j , and time-offsets a, b , we write

$$f_{P,t,n,i,a}(\cdot) = f_{P,n,i,a}(t/n, \cdot), \quad \hat{f}_{t,n,i,a}(\cdot) = \hat{f}_{t,n,i,a}(t/n, \cdot), \\ g_{P,t,n,j,b}(\cdot) = g_{P,n,j,b}(t/n, \cdot), \quad \hat{g}_{t,n,j,b}(\cdot) = \hat{g}_{t,n,j,b}(t/n, \cdot),$$

to denote the time-varying regression functions and the corresponding sieve estimators from Section 4.3 using the notation of Section 2.3. For all times $t \in \mathcal{T}_n$, the algorithms used to construct the sieve estimators from Section 4.3 for rescaled time $t/n \in \mathcal{U}_n$ are Borel measurable functions of the datasets $\mathfrak{D}_{t,n,i,a}^{\hat{f}}$ and $\mathfrak{D}_{t,n,j,b}^{\hat{g}}$. The measurability of the causal mechanisms of the observed processes from Assumption 4.1 ensures that these sieve estimators have the causal representations $G_{t,n,i,a}^{\mathcal{A}^{\hat{f}}}(\mathcal{H}_{t,a}^{\mathfrak{D}^{\hat{f}}})$ and $G_{t,n,j,b}^{\mathcal{A}^{\hat{g}}}(\mathcal{H}_{t,b}^{\mathfrak{D}^{\hat{g}}})$ from Assumption 3.2.

Further, note that the sieve estimators are Borel measurable functions from \mathbb{R}^{dz} to \mathbb{R} . The measurability of the causal mechanisms of the covariate processes from Assumption 4.1 ensures that the sieve estimator's predictions have the causal representations $G_{t,n,i,a}^{\hat{f}}(\mathcal{H}_{t,a}^{\hat{f}})$ and $G_{t,n,j,b}^{\hat{g}}(\mathcal{H}_{t,b}^{\hat{g}})$ from Assumption 3.3. Similarly, note that the Borel measurability of the conditional expectations $f_{P,t,n,i,a}$ and $g_{P,t,n,j,b}$ ensures that the sieve estimator's prediction errors are Borel measurable functions from \mathbb{R}^{dz} to \mathbb{R} . Again, by the measurability of the causal mechanisms of the covariate processes from Assumption 4.1, the prediction errors are ensured to have the causal representations $G_{P,t,n,i,a}^{\hat{w}^{\hat{f}}}(\mathcal{H}_{t,a}^{\hat{f}})$

and $G_{P,t,n,j,b}^{\hat{w}^g}(\mathcal{H}_{t,b}^{\hat{g}})$ from Assumption 3.3. In view of the boundedness of the sieve estimator's predictions by construction, the regularity conditions for the time-varying partial response functions from Assumption 4.2, and the additive form of the time-varying regression functions from Assumption 4.2, there exists some $q \geq 2$ such that for all $n \in \mathbb{N}$, $t \in \mathcal{T}_n$, and $(i, j, a, b) \in \mathcal{D}_n$, the prediction errors satisfy

$$\sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{E}_P(|\hat{w}_{P,t,n,i,a}^f|^q) < \infty, \quad \sup_{P \in \mathcal{P}_{0,n}^*} \mathbb{E}_P(|\hat{w}_{P,t,n,j,b}^g|^q) < \infty.$$

Hence, the sieve estimator's predictions and prediction errors meet all the conditions required by Assumption 3.3.

The distribution-uniform assumptions of Theorem 4.1 imply that the distribution-pointwise assumptions of Theorem 3.2 in Ding and Zhou [DZ21] hold for each distribution in the collection. Specifically, for each $n \in \mathbb{N}$ and $P \in \mathcal{P}_{0,n}^*$, Assumption 4.2 implies the additive form of the time-varying regression functions in [DZ21], Assumptions 4.1, 4.4, 4.6 imply Assumption 2.1 in [DZ21], Assumption 4.5 implies Assumption 2.2 in [DZ21], Assumption 4.3 implies Assumption 3.1 in [DZ21], and Assumption 4.7 implies Assumption 3.2 in [DZ21]. Next, we consider the additional regularity condition required by Theorem 3.2 in Ding and Zhou [DZ21] involving the rate of decay in temporal dependence and the rate of growth of the largest sup-norm of the basis functions for time.

Recall the numbers of observations $T_{t,n,i,a}^{\hat{f}}, T_{t,n,j,b}^{\hat{g}}$ in the datasets $\mathfrak{D}_{t,n,i,a}^{\hat{f}}, \mathfrak{D}_{t,n,j,b}^{\hat{g}}$ used to construct the sieve estimators $\hat{f}_{t,n,i,a}(t/n, \cdot), \hat{g}_{t,n,j,b}(t/n, \cdot)$ of the time-varying regression functions at rescaled time $t/n \in \mathcal{U}_n$. Also, recall the numbers of basis functions \tilde{c}_n, \tilde{d}_n from Section 4.3. As previously noted in Section 4.3, we simplified the notation for the numbers of basis functions $\{\phi_{\ell_1}(u)\}, \{\varphi_{\ell_2}(z)\}$ for the estimators $\hat{f}_{t,n,i,a,k,c}(t/n, \cdot)$ and $\hat{g}_{t,n,j,b,k,c}(t/n, \cdot)$ of the time-varying partial response functions at rescaled time $t/n \in \mathcal{U}_n$ from $\tilde{c}_{t,n,i,a,k,c}^{\hat{f}}, \tilde{d}_{t,n,i,a,k,c}^{\hat{f}}$ and $\tilde{c}_{t,n,j,b,k,c}^{\hat{g}}, \tilde{d}_{t,n,j,b,k,c}^{\hat{g}}$ to \tilde{c}_n, \tilde{d}_n . We will now require the full notation for the numbers of basis functions.

For the convergence rate guarantees from Theorem 3.2 in Ding and Zhou [DZ21] to be applicable in our setting, we must have

$$\begin{aligned} \tilde{c}_{t,n,i,a,k,c}^{\hat{f}} \tilde{d}_{t,n,i,a,k,c}^{\hat{f}} & \left(\frac{1}{\sqrt{T_{t,n,i,a}^{\hat{f}}}} + \frac{(T_{t,n,i,a}^{\hat{f}})^{\frac{2}{\min(\bar{\beta}, \bar{\beta}^\infty)+1}}}{T_{t,n,i,a}^{\hat{f}}} \right) \sup_{\ell_1 \in [\tilde{c}_{t,n,i,a,k,c}^{\hat{f}}]} \sup_{u \in \mathcal{U}_n} |\phi_{\ell_1}(u)|^2 = o(1), \\ \tilde{c}_{t,n,j,b,k,c}^{\hat{g}} \tilde{d}_{t,n,j,b,k,c}^{\hat{g}} & \left(\frac{1}{\sqrt{T_{t,n,j,b}^{\hat{g}}}} + \frac{(T_{t,n,j,b}^{\hat{g}})^{\frac{2}{\min(\bar{\beta}, \bar{\beta}^\infty)+1}}}{T_{t,n,j,b}^{\hat{g}}} \right) \sup_{\ell_1 \in [\tilde{c}_{t,n,j,b,k,c}^{\hat{g}}]} \sup_{u \in \mathcal{U}_n} |\phi_{\ell_1}(u)|^2 = o(1), \end{aligned}$$

for each time $t \in \mathcal{T}_n$ and combination of dimensions $i \in [d_X], j \in [d_Y], k \in [d_Z]$ and time-offsets $a \in A_i, b \in B_j, c \in C_k$. This condition is satisfied for the following reasons. First, we have

$$\sup_{\ell_1 \in [\tilde{c}_{t,n,i,a,k,c}^{\hat{f}}]} \sup_{u \in \mathcal{U}_n} |\phi_{\ell_1}(u)|^2 \lesssim (\tilde{c}_{t,n,i,a,k,c}^{\hat{f}})^2, \quad \sup_{\ell_1 \in [\tilde{c}_{t,n,j,b,k,c}^{\hat{g}}]} \sup_{u \in \mathcal{U}_n} |\phi_{\ell_1}(u)|^2 \lesssim (\tilde{c}_{t,n,j,b,k,c}^{\hat{g}})^2,$$

because the basis functions are chosen to be mapped Legendre polynomials; see Appendix C in Ding and Zhou [DZ21] and Section 3 in Belloni et al. [Bel+15]. For more information about sieve estimators and other basis functions, see [New97; Hua98; Che07; DZ20; DZ25]. Second, because we have chosen the numbers of basis functions to be $O(\log(T_n))$ in the setup of Theorem 4.1. Third, because the constants $\bar{\beta}, \bar{\beta}^\infty$ from Assumption 4.5 are both greater than 2. Fourth, because $T_{t,n,i,a}^{\hat{f}} = o(T_n)$ and $T_{t,n,j,b}^{\hat{g}} = o(T_n)$ regardless of whether the sieve estimators are fit once based on all the data or sequentially as in Remark 4.1. To be clear, this is due to the infill asymptotic framework of locally stationary processes, so that more and more observations are available for each local structure as n grows.

Therefore, the main inequality in the proof of Theorem 3.2 in [DZ21] holds for each $P \in \mathcal{P}_{0,n}^*$ and $n \in \mathbb{N}$ because all of the theorem's assumptions are satisfied under the stronger assumptions of Theorem 4.1. Moreover, for each $n \in \mathbb{N}$, the supremum over $P \in \mathcal{P}_{0,n}^*$ of the final upper bound for the main inequality in the proof of Theorem 3.2 in [DZ21] is finite under the distribution-uniform assumptions of Theorem 4.1. Thus, by basic properties of the supremum, the main inequality in the proof of Theorem 3.2 in [DZ21] holds with a supremum over $P \in \mathcal{P}_{0,n}^*$ for each $n \in \mathbb{N}$. In view of the

notational changes described in Remark 3.2 in [DZ21], the same steps in the proof of Theorem 3.2 in [DZ21] imply that this distribution-uniform inequality also holds in the general regression setting with time-offsets. We do not repeat the proof of Theorem 3.2 in [DZ21] here, as the only changes are in the notation. Putting it all together, the prediction errors of the sieve estimators with the setup of Theorem 4.1 satisfy

$$\sup_{P \in \mathcal{P}_{0,n}^*} \max_{i \in [d_X], a \in A_i} \max_{t \in \mathcal{T}_n} \mathbb{E}_P \left(\left| \hat{w}_{P,t,n,i,a}^f \right|^2 \right)^{\frac{1}{2}} = o(T_n^{-\frac{1}{2+\delta}} \log^3(T_n)),$$

$$\sup_{P \in \mathcal{P}_{0,n}^*} \max_{j \in [d_Y], b \in B_j} \max_{t \in \mathcal{T}_n} \mathbb{E}_P \left(\left| \hat{w}_{P,t,n,j,b}^g \right|^2 \right)^{\frac{1}{2}} = o(T_n^{-\frac{1}{2+\delta}} \log^3(T_n)),$$

for any $\delta > 0$. Since $D_n = O(T_n^{\frac{1}{6}})$ and $\tau_n = o(\log^{-(1+\delta')}(T_n))$ for some $\delta' > 0$, the convergence rates required by Theorem 3.1 are achieved by the sieve estimators with the setup of Theorem 4.1.

□

B Distribution-Uniform Theory

In this section, we state distribution-uniform versions of the results from Mies and Steland [MS23]. All of the results in this section can be applied to general triangular array frameworks for high-dimensional nonstationary nonlinear processes, such as locally stationary processes.

B.1 Distribution-uniform strong Gaussian approximation

To begin, let us introduce the setting rigorously. Let Ω be a sample space, \mathcal{B} the Borel sigma-algebra, and (Ω, \mathcal{B}) a measurable space. Fix $n \in \mathbb{N}$ and define $\mathcal{T}_n = [n]$ throughout this section. Let (Ω, \mathcal{B}) be equipped with a family of probability measures $(\mathbb{P}_P)_{P \in \mathcal{P}_n}$ so that the distribution of the high-dimensional stochastic system

$$(G_{t,n}(\mathcal{H}_s))_{t \in \mathcal{T}_n, s \in \mathbb{Z}},$$

or, in the locally stationary setting

$$(\tilde{G}_n(u, \mathcal{H}_s))_{u \in [0,1], s \in \mathbb{Z}},$$

under \mathbb{P}_P is $P \in \mathcal{P}_n$. Here $\mathcal{H}_t = (\eta_t, \eta_{t-1}, \dots)$, where $(\eta_t)_{t \in \mathbb{Z}}$ is a sequence of iid random vectors with dimension $d^\eta = d_n^\eta$, and $G_{t,n} : (\mathbb{R}^{d^\eta})^\infty \rightarrow \mathbb{R}^{d_n}$ is a measurable function—where we endow $(\mathbb{R}^{d^\eta})^\infty$ with the σ -algebra generated by all finite projections. For each $t \in \mathcal{T}_n$, $G_{t,n}(\mathcal{H}_s)$ is a well-defined high-dimensional random vector for every $s \in \mathbb{Z}$, and $(G_{t,n}(\mathcal{H}_s))_{s \in \mathbb{Z}}$ is a high-dimensional stationary ergodic process.

For each $n \in \mathbb{N}$, write the \mathbb{R}^{d_n} -valued process of interest as $(W_{t,n})_{t \in \mathcal{T}_n}$. We assume that for each $n \in \mathbb{N}$ and $t \in \mathcal{T}_n$, the random vector $W_{t,n}$ has a causal representation; that is, it can be represented as a measurable function of these iid random vectors

$$W_{t,n} = G_{t,n}(\mathcal{H}_t).$$

Similarly, for the causal representations in the locally stationary setting, the measurable function $\tilde{G}_n(u, \cdot) : (\mathbb{R}^{d^\eta})^\infty \rightarrow \mathbb{R}^{d_n}$ is defined for each rescaled time $u \in [0, 1]$, and we assume that

$$W_{t,n} = \tilde{G}_n(t/n, \mathcal{H}_t).$$

We can use the results in this section for locally stationary processes by writing

$$G_{t,n}(\mathcal{H}_t) = \tilde{G}_n(t/n, \mathcal{H}_t).$$

The family of probability measures $(\mathbb{P}_P)_{P \in \mathcal{P}_n}$ is defined with respect to the same measurable space (Ω, \mathcal{B}) , but need not have the same dominating measure. Denote a family of probability spaces by $(\Omega, \mathcal{B}, \mathbb{P}_P)_{P \in \mathcal{P}_n}$. When we say that the process $(W_{t,n})_{t \in \mathcal{T}_n}$ is defined on the collection of probability spaces $(\Omega, \mathcal{B}, \mathbb{P}_P)_{P \in \mathcal{P}_n}$ for some $n \in \mathbb{N}$, we mean that $(W_{t,n})_{t \in \mathcal{T}_n}$ is defined on the probability space $(\Omega, \mathcal{B}, \mathbb{P}_P)$ for each $P \in \mathcal{P}_n$.

Note that the causal representations in this paper use sequences of iid random vectors, whereas the causal representations in Mies and Steland [MS23] use sequences of iid $\text{Unif}[0, 1]$ random variables. The same arguments used in Mies and Steland [MS23] can be applied when using our formulation of the causal representations with iid random vectors. The only reason we write the causal representations in this way is for the sake of clarity.

In fact, standard results in probability theory imply that the causal representations based on measurable functions of sequences of iid $\text{Unif}[0, 1]$ random variables as in Mies and Steland [MS23] are already sufficiently general. For example, see Kallenberg [Kal21] Lemma 4.21, Lemma 4.22, and the surrounding discussion. More specifically, the causal representations with sequences of iid $\text{Unif}[0, 1]$ random variables can express the causal representations with sequences of random vectors by including compositions with additional measurable functions for (1) replicating each of the iid $\text{Unif}[0, 1]$ random variables, and (2) inverse sampling via products of conditional distributions; see Section 2.5 of Rubinstein and Kroese [RK16] on random vector generation.

Next, we define our measure of temporal dependence for the process. For the following definition, let $(\tilde{\eta}_t)_{t \in \mathbb{Z}}$ be an iid copy of $(\eta_t)_{t \in \mathbb{Z}}$ and denote

$$\tilde{\mathcal{H}}_{t,j} = (\eta_t, \dots, \eta_{j+1}, \tilde{\eta}_{t-j}, \eta_{t-j-1}, \dots),$$

to be \mathcal{H}_t with the j -th input in the past η_{t-j} replaced with the iid copy $\tilde{\eta}_{t-j}$.

Definition B.1 (Functional dependence measure). *For $n \in \mathbb{N}$, $P \in \mathcal{P}_n$, $t \in \mathcal{T}_n$, define the functional dependence measure of $W_{t,n} = G_{t,n}(\mathcal{H}_t)$ as*

$$\theta_{P,t,n}(j, q, r) = (\mathbb{E}_P \|G_{t,n}(\mathcal{H}_t) - G_{t,n}(\tilde{\mathcal{H}}_{t,j})\|_r^q)^{\frac{1}{q}},$$

with $h \in \mathbb{N}_0$, $q \geq 1$, $r \geq 1$.

We state the following distribution-uniform assumptions about the temporal dependence and nonstationarity of the process for some collections of distributions \mathcal{P}_n for some $n \in \mathbb{N}$. Note that we write the time of the input sequence as 0 when it does not matter due to stationarity.

Assumption B.1 (Distribution-uniform decay of temporal dependence). *We assume that there exist $\beta > 0$, $q \geq 2$ and a constant $\Theta_n > 0$, such that for all times $t \in \mathcal{T}_n$ it holds*

$$\sup_{P \in \mathcal{P}_n} \theta_{P,t,n}(j, q, r) \leq \Theta_n \cdot (j \vee 1)^{-\beta},$$

for $j \geq 0$, and that

$$\sup_{P \in \mathcal{P}_n} (\mathbb{E}_P \|G_{t,n}(\mathcal{H}_0)\|_2^q)^{1/q} \leq \Theta_n.$$

Assumption B.2 (Distribution-uniform total variation condition for nonstationarity). *Recall Θ_n from Assumption B.1. Assume that there exists some $\Gamma_n \geq 1$ such that*

$$\sup_{P \in \mathcal{P}_n} \left(\sum_{t=2}^n (\mathbb{E}_P \|G_{t,n}(\mathcal{H}_0) - G_{t-1,n}(\mathcal{H}_0)\|_2^2)^{\frac{1}{2}} \right) \leq \Gamma_n \cdot \Theta_n.$$

Note that the assumptions regarding the temporal dependence and nonstationarity of the process of error products, as stated in Section 3.4, ensure that both Assumptions B.1 and B.2 hold for each $n \in \mathbb{N}$. Furthermore, since the assumptions in Section 4.5 are strictly stronger than those in Section 3.4, the results from this section can be applied to the process of error products in both Sections 3 and 4.

Define the two rates

$$\chi(q, \beta) = \begin{cases} \frac{q-2}{6q-4}, & \beta \geq \frac{3}{2}, \\ \frac{(\beta-1)(q-2)}{q(4\beta-3)-2}, & \beta \in (1, \frac{3}{2}), \end{cases}$$

and

$$\xi(q, \beta) = \begin{cases} \frac{q-2}{6q-4}, & \beta \geq 3, \\ \frac{(\beta-2)(q-2)}{(4\beta-6)q-4}, & \frac{3+\frac{2}{q}}{1+\frac{2}{q}} < \beta < 3, \\ \frac{1}{2} - \frac{1}{\beta}, & 2 < \beta \leq \frac{3+\frac{2}{q}}{1+\frac{2}{q}}, \end{cases}$$

which will appear in the results in this section. In general, the Gaussian approximation allows the dimensions to grow as $d_n = O(n^{\frac{1-\delta}{1+2\xi(q,\beta)}})$ for some $\delta > 0$. In the limiting case when $\beta \geq 3$ and $q \rightarrow \infty$, this corresponds to $d_n = O(n^{\frac{1}{4}-\delta'})$ for some $\delta' > 0$.

Let us briefly recall some notation used in the main text. Recall $\bar{\beta}^R > 3$, $\bar{q}^R > 4$ from Assumption 3.5, as well as the number of times T_n and the number of dimension/time-offset combinations D_n from Section 2.1. For the dGCM test, we allow $D_n = O(T_n^{r(\bar{q}^R, \bar{\beta}^R)})$ where

$$r(\bar{q}^R, \bar{\beta}^R) = \min \left(\frac{1-\delta}{1 + \frac{1}{2\xi(\bar{q}^R, \bar{\beta}^R)}}, \frac{1}{6} \right), \quad (22)$$

for some $\delta > 0$. The limiting factor that leads to this requirement is not from the strong Gaussian approximation, but rather to ensure that the convergence rate requirements can be achieved by the time-varying nonparametric regression estimators.

The following result is a distribution-uniform version of the strong Gaussian approximation from Theorem 3.1 in Mies and Steland [MS23].

Lemma B.1. *For some sample size $n \in \mathbb{N}$ and collection of distributions \mathcal{P}_n for the stochastic system $(G_{t,n}(\mathcal{H}_s))_{t \in \mathcal{T}_n, s \in \mathbb{Z}}$, suppose that Assumption B.1 is satisfied for \mathcal{P}_n with some $q > 2$, $\beta > 1$ and constant $\Theta_n > 0$. Let the \mathbb{R}^{d_n} -valued process $(W_{t,n})_{t \in \mathcal{T}_n}$ be defined on the collection of probability spaces $(\Omega, \mathcal{B}, \mathbb{P}_P)_{P \in \mathcal{P}_n}$ so that $W_{t,n} = G_{t,n}(\mathcal{H}_t)$ with $\mathbb{E}_P(W_{t,n}) = 0$ for each time $t \in \mathcal{T}_n$ and distribution $P \in \mathcal{P}_n$. Also, suppose the dimension $d_n < cn$ for some constant $c > 0$. Then, on a potentially enriched collection of probability spaces $(\Omega', \mathcal{B}', \mathbb{P}'_P)_{P \in \mathcal{P}_n}$, there exist random vectors $(W'_{t,n})_{t \in \mathcal{T}_n}$ with the same distribution as $(W_{t,n})_{t \in \mathcal{T}_n}$ for each $P \in \mathcal{P}_n$, and independent Gaussian random vectors $(V'_{t,n})_{t \in \mathcal{T}_n}$ with $\mathbb{E}_P(V'_{t,n}) = 0$ for each $t \in \mathcal{T}_n$, $P \in \mathcal{P}_n$, such that*

$$\sup_{P \in \mathcal{P}_n} \left(\mathbb{E}_P \max_{k \leq n} \left\| \frac{1}{\sqrt{n}} \sum_{t=1}^k (W'_{t,n} - V'_{t,n}) \right\|_2^2 \right)^{\frac{1}{2}} \leq K \Theta_n \sqrt{\log(n)} \left(\frac{d_n}{n} \right)^{\chi(q,\beta)}$$

for some universal constant K depending only on q , c , and β .

If $\beta > 2$, then the local long-run covariance matrix $\Sigma_{P,t,n} = \sum_{h=-\infty}^{\infty} \text{Cov}_P(G_{t,n}(\mathcal{H}_0), G_{t,n}(\mathcal{H}_h))$ is well-defined for each $t \in \mathcal{T}_n$, $P \in \mathcal{P}_n$ by Lemma B.5. If Assumption B.2 is also satisfied for \mathcal{P}_n , then on $(\Omega', \mathcal{B}', \mathbb{P}'_P)_{P \in \mathcal{P}_n}$ there exist random vectors $(W'_{t,n})_{t \in \mathcal{T}_n}$ which have the same distribution as $(W_{t,n})_{t \in \mathcal{T}_n}$ for each $P \in \mathcal{P}_n$, and independent Gaussian random vectors $(V_{t,n}^*)_{t \in \mathcal{T}_n}$ where $V_{t,n}^* \sim \mathcal{N}(0, \Sigma_{P,t,n})$ for each $t \in \mathcal{T}_n$, $P \in \mathcal{P}_n$, such that

$$\sup_{P \in \mathcal{P}_n} \left(\mathbb{E}_P \max_{k \leq n} \left\| \frac{1}{\sqrt{n}} \sum_{t=1}^k (W'_{t,n} - V_{t,n}^*) \right\|_2^2 \right)^{\frac{1}{2}} \leq K \Theta_n \Gamma_n^{\frac{1}{2} \frac{\beta-2}{\beta-1}} \sqrt{\log(n)} \left(\frac{d_n}{n} \right)^{\xi(q,\beta)}$$

for some universal constant K depending only on q , c , and β .

Proof of Lemma B.1: Assumptions B.1 and B.2 are distribution-uniform versions of conditions (G.1) and (G.2) from Mies and Steland [MS23]. Hence, under the assumptions of the Lemma related to Assumptions B.1 and B.2, the distribution-pointwise inequalities from Theorem 3.1 in Mies and Steland [MS23] hold for each $P \in \mathcal{P}_n$. Since the suprema over all distributions in the collection \mathcal{P}_n of the upper bounds are finite, the distribution-uniform inequalities from the Lemma hold for \mathcal{P}_n by basic properties of the supremum. \square

Recently, Bonnerjee et al. [BKW24] introduced univariate strong Gaussian approximation results with optimal rates and explicit constructions, building on prior work by Karmakar and Wu [KW20]. We emphasize that the distribution-uniform strong Gaussian approximation for high-dimensional non-stationary nonlinear processes from Lemma B.1 does not achieve this optimal rate. However, the convergence rates for the prediction errors from the estimation of the time-varying regression functions dominate the strong Gaussian approximation rates. Therefore, we do not “lose anything” by using Lemma B.1 in our regression-based conditional independence test instead of a distribution-uniform version of the strong Gaussian approximation from Bonnerjee et al. [BKW24], as our main results would not change in any meaningful way.

The following result is a distribution-uniform version of Theorem 3.2 from Mies and Steland [MS23].

Lemma B.2. For some sample size $n \in \mathbb{N}$ and collection of distributions \mathcal{P}_n for the stochastic system $(G_{t,n}(\mathcal{H}_s))_{t \in \mathcal{T}_n, s \in \mathbb{Z}}$, let the \mathbb{R}^{d_n} -valued process $(W_{t,n})_{t \in \mathcal{T}_n}$ be defined on the collection of probability spaces $(\Omega, \mathcal{B}, \mathbb{P}_P)_{P \in \mathcal{P}_n}$ so that $W_{t,n} = G_{t,n}(\mathcal{H}_t)$ with $W_{t,n} \in L_q(P)$ and $\theta_{P,t,n}(j, q, r)$ as in Definition B.1 for each $P \in \mathcal{P}_n$ and some $2 \leq r \leq q < \infty$. There exists a universal constant $K = K(q, r)$ such that

$$\begin{aligned} \sup_{P \in \mathcal{P}_n} \left(\mathbb{E}_P \max_{k \leq n} \left\| \sum_{t=1}^k (W_{t,n} - \mathbb{E}_P(W_{t,n})) \right\|_r^q \right)^{\frac{1}{q}} &\leq \sup_{P \in \mathcal{P}_n} \left(K n^{\frac{1}{2} - \frac{1}{q}} \sum_{j=1}^{\infty} \left(\sum_{t=1}^n \theta_{P,t,n}^q(j, q, r) \right)^{\frac{1}{q}} \right) \\ &\leq \sup_{P \in \mathcal{P}_n} \left(K n^{\frac{1}{2}} \sum_{j=1}^{\infty} \max_{t \leq n} \theta_{P,t,n}(j, q, r) \right). \end{aligned}$$

In the special case $r = 2$, the inequality may be improved to

$$\begin{aligned} \sup_{P \in \mathcal{P}_n} \left(\mathbb{E}_P \max_{k \leq n} \left\| \sum_{t=1}^k (W_{t,n} - \mathbb{E}_P(W_{t,n})) \right\|_2^q \right)^{\frac{1}{q}} \\ \leq \sup_{P \in \mathcal{P}_n} \left(K \sum_{j=1}^{\infty} (j \wedge n)^{\frac{1}{2} - \frac{1}{q}} \left(\sum_{t=1}^n \theta_{P,t,n}^q(j, q, 2) \right)^{\frac{1}{q}} + K \sum_{j=1}^n \left(\sum_{t=1}^n \theta_{P,t,n}^2(j, 2, 2) \right)^{\frac{1}{2}} \right). \end{aligned}$$

Proof of Lemma B.2: Under the assumptions of the Lemma, the distribution-pointwise inequalities from Theorem 3.2 in Mies and Steland [MS23] hold for each $P \in \mathcal{P}_n$. Since the suprema over all distributions in the collection \mathcal{P}_n of the upper bounds are always finite, the distribution-uniform inequalities from the Lemma hold for \mathcal{P}_n by basic properties of the supremum. \square

B.2 Distribution-uniform feasible Gaussian approximation

We introduce distribution-uniform versions of Theorem 4.1 and Proposition 4.2 from Mies and Steland [MS23] so that the distribution-uniform strong Gaussian approximation from Section B.1 can be used for statistical inference. The key is a distribution-uniform cumulative covariance estimator $\hat{Q}_{k,n}$ of the cumulative covariance matrices $Q_{P,k,n} = \sum_{t=1}^k \Sigma_{P,t,n}$ where $\Sigma_{P,t,n} = \sum_{h=-\infty}^{\infty} \text{Cov}_P(G_{t,n}(\mathcal{H}_0), G_{t,n}(\mathcal{H}_h))$ and $W_{t,n} = G_{t,n}(\mathcal{H}_t)$. We will prove these guarantees for the same estimator from Mies and Steland [MS23], namely

$$\hat{Q}_{k,n} = \sum_{r=L_n}^k \frac{1}{L_n} \left(\sum_{s=r-L_n+1}^r W_{s,n} \right)^{\otimes 2}$$

for some window size $L_n \asymp n^\zeta$ for some $\zeta \in (0, \frac{1}{2})$.

The following result is a distribution-uniform version of Theorem 4.1 from Mies and Steland [MS23].

Lemma B.3. For some sample size $n \in \mathbb{N}$ and collection of distributions \mathcal{P}_n for the stochastic system $(G_{t,n}(\mathcal{H}_s))_{t \in \mathcal{T}_n, s \in \mathbb{Z}}$, let the \mathbb{R}^{d_n} -valued process $(W_{t,n})_{t \in \mathcal{T}_n}$ be defined on the collection of probability spaces $(\Omega, \mathcal{B}, \mathbb{P}_P)_{P \in \mathcal{P}_n}$ so that $W_{t,n} = G_{t,n}(\mathcal{H}_t)$ and Assumptions B.1 and B.2 are satisfied for \mathcal{P}_n with $q \geq 4$ and $\beta > 2$. Then

$$\sup_{P \in \mathcal{P}_n} \left(\mathbb{E}_P \max_{k=L_n, \dots, n} \left\| \hat{Q}_{k,n} - \sum_{t=1}^k \Sigma_{P,t,n} \right\|_{\text{tr}} \right) \leq K \Theta_n^2 \left(\Gamma_n \sqrt{L_n} + \sqrt{nd_n L_n} + n L_n^{-1} + n L_n^{2-\beta} \right)$$

for some universal constant K depending only on β and q .

Proof of Lemma B.3: Assumptions B.1 and B.2 are distribution-uniform versions of conditions (G.1) and (G.2) from Mies and Steland [MS23]. Hence, under the assumptions of the Lemma related to Assumptions B.1 and B.2, the distribution-pointwise inequalities from Theorem 4.1 in Mies and Steland [MS23] hold for each $P \in \mathcal{P}_n$. Since the supremum over all distributions in the collection \mathcal{P}_n of the upper bound is always finite, the distribution-uniform inequality from the Lemma holds for \mathcal{P}_n by basic properties of the supremum. \square

The next result is a distribution-uniform version of Proposition 4.2 from Mies and Steland [MS23].

Lemma B.4. For some sample size $n \in \mathbb{N}$, let \mathcal{P}_n be a collection of distributions for the stochastic system $(G_{t,n}(\mathcal{H}_s))_{t \in \mathcal{T}_n, s \in \mathbb{Z}}$. Let $\Sigma_{P,t,n}, \Sigma'_{P,t,n} \in \mathbb{R}^{d_n \times d_n}$ be symmetric, positive definite matrices for each $t \in \mathcal{T}_n$, $P \in \mathcal{P}_n$, and let $(V_{t,n})_{t \in \mathcal{T}_n}$ be independent random vectors defined on the collection of probability spaces $(\Omega, \mathcal{B}, \mathbb{P}_P)_{P \in \mathcal{P}_n}$ so that $V_{t,n} \sim \mathcal{N}(0, \Sigma_{P,t,n})$ for each $t \in \mathcal{T}_n$, $P \in \mathcal{P}_n$. Then, on a potentially enriched collection of probability spaces $(\Omega', \mathcal{B}', \mathbb{P}'_P)_{P \in \mathcal{P}_n}$, there exist independent random vectors $(V'_{t,n})_{t \in \mathcal{T}_n}$ with $V'_{t,n} \sim \mathcal{N}(0, \Sigma'_{P,t,n})$ for each $t \in \mathcal{T}_n$, $P \in \mathcal{P}_n$ such that

$$\sup_{P \in \mathcal{P}_n} \left(\mathbb{E}_P \max_{k \leq n} \left\| \sum_{t=1}^k V_{t,n} - \sum_{t=1}^k V'_{t,n} \right\|_2^2 \right) \leq \sup_{P \in \mathcal{P}_n} \left(K \log(n) [\sqrt{n \delta_{P,n} \rho_{P,n}} + \rho_{P,n}] \right),$$

where

$$\delta_{P,n} = \max_{k \leq n} \left\| \sum_{t=1}^k \Sigma_{P,t,n} - \sum_{t=1}^k \Sigma'_{P,t,n} \right\|_{\text{tr}},$$

$$\rho_{P,n} = \max_{t \leq n} \|\Sigma_{P,t,n}\|_{\text{tr}}.$$

Proof of Lemma B.4: The distribution-pointwise inequalities from Proposition 4.2 in Mies and Steland [MS23] hold for each $P \in \mathcal{P}_n$. Since the supremum over all distributions in the collection \mathcal{P}_n of the upper bound is always finite, the distribution-uniform inequality from the Lemma holds for \mathcal{P}_n by basic properties of the supremum. \square

B.3 Auxiliary Lemmas

The following result is a distribution-uniform version of Proposition 5.4 from Mies and Steland [MS23].

Lemma B.5. For some sample size $n \in \mathbb{N}$ and collection of distributions \mathcal{P}_n for the stochastic system $(G_{t,n}(\mathcal{H}_s))_{t \in \mathcal{T}_n, s \in \mathbb{Z}}$, let Assumption B.1 be satisfied for \mathcal{P}_n with some $q \geq 2$, $\beta > 0$, and constant $\Theta_n > 0$. Denote

$$\gamma_{P,t,n}(h) = \text{Cov}_P[G_{t,n}(\mathcal{H}_0), G_{t,n}(\mathcal{H}_h)] \in \mathbb{R}^{d_n \times d_n}.$$

Then for all $t \in \mathcal{T}_n$, $h \in \mathbb{Z}$, we have

$$\sup_{P \in \mathcal{P}_n} \|\gamma_{P,t,n}(h)\|_{\text{tr}} \leq \Theta_n^2 \sum_{j=h}^{\infty} j^{-\beta},$$

where $\|\cdot\|_{\text{tr}}$ denotes the trace norm. Hence, if $\beta > 2$, then the long-run covariance matrix

$$\gamma_{P,t,n} = \sum_{h=-\infty}^{\infty} \gamma_{P,t,n}(h),$$

is well-defined for all $t \in \mathcal{T}_n$, $P \in \mathcal{P}_n$.

Proof of Lemma B.5: The distribution-pointwise inequality from Proposition 5.4 in Mies and Steland [MS23] holds for each $P \in \mathcal{P}_n$. Since the supremum over all distributions in the collection \mathcal{P}_n of the upper bound is always finite, the distribution-uniform inequality from the Lemma holds for \mathcal{P}_n by basic properties of the supremum. \square

The following result is a distribution-uniform version of the Rosenthal inequality from the first part of Theorem 5.6 from Mies and Steland [MS23].

Lemma B.6. For some sample size $n \in \mathbb{N}$ and collection of distributions \mathcal{P}_n , let $(M_{t,n})_{t \in \mathcal{T}_n}$ be a \mathbb{R}^{d_n} -valued martingale-difference sequence with distribution determined by $P \in \mathcal{P}_n$. For each $2 \leq r \leq q < \infty$, there exists a finite factor $C_{q,r}$ such that for any $n, d_n \in \mathbb{N}$, we have

$$\begin{aligned} \sup_{P \in \mathcal{P}_n} \left(\mathbb{E}_P \max_{k \leq n} \left\| \sum_{t=1}^k M_{t,n} \right\|_r^q \right)^{\frac{1}{q}} &\leq C_{q,r} n^{\frac{1}{2} - \frac{1}{q}} \sup_{P \in \mathcal{P}_n} \left(\sum_{t=1}^n \mathbb{E}_P \|M_{t,n}\|_r^q \right)^{\frac{1}{q}} \\ &\leq C_{q,r} n^{\frac{1}{2}} \sup_{P \in \mathcal{P}_n} \left(\max_{t \leq n} (\mathbb{E}_P \|M_{t,n}\|_r^q)^{\frac{1}{q}} \right). \end{aligned}$$

Proof of Lemma B.6: The distribution-pointwise inequalities from the first part of Theorem 5.6 in Mies and Steland [MS23] hold for each $P \in \mathcal{P}_n$. Since the suprema over all distributions in the collection \mathcal{P}_n of the upper bounds are always finite, the distribution-uniform inequality from the Lemma holds for \mathcal{P}_n by basic properties of the supremum. \square

The following result is similar to the bounded convergence lemma from Lemma 25 in Shah and Peters [SP20].

Lemma B.7. *For some sample size $n \in \mathbb{N}$ and collection of distributions \mathcal{P}_n , let X_n be a generic real-valued random variable with distribution determined by $P \in \mathcal{P}_n$, where the collection of distributions \mathcal{P}_n can change with n . Let $K > 0$, and suppose that $|X_n| \leq K$ for all $n \in \mathbb{N}$ and $X_n = o_{\mathcal{P}}(1)$. Then we have*

$$\sup_{P \in \mathcal{P}_n} \mathbb{E}_P(|X_n|) = o(1).$$

Proof of Lemma B.7: For any given $\epsilon > 0$,

$$|X_n| = |X_n| \mathbb{1}_{\{|X_n| > \epsilon\}} + |X_n| \mathbb{1}_{\{|X_n| \leq \epsilon\}} \leq K \mathbb{1}_{\{|X_n| > \epsilon\}} + \epsilon.$$

By the assumption that $X_n = o_{\mathcal{P}}(1)$, we can find some $N \in \mathbb{N}$ such that $\sup_{P \in \mathcal{P}_n} \mathbb{P}_P(|X_n| > \epsilon) < \epsilon/K$ for $n \geq N$. Hence, for $n \geq N$ we have

$$\sup_{P \in \mathcal{P}_n} \mathbb{E}_P(|X_n|) \leq K \sup_{P \in \mathcal{P}_n} \mathbb{P}_P(|X_n| > \epsilon) + \epsilon < 2\epsilon.$$

Since $\epsilon > 0$ was arbitrary, we obtain the desired result. \square

C Supplementary Appendix

C.1 Directions for future work

We discuss three promising avenues for future work. First, we plan to develop statistical techniques for nonstationary nonlinear time series which utilize our conditional independence test as a core component. In our companion paper [WHR25], we apply our test to the problem of identifying auxiliary indicators for forecasting nonstationary time series. It would also be of interest to develop a variable selection procedure for forecasting and a causal discovery algorithm for nonstationary nonlinear time series. We note that it may be possible to use the theoretical tools for nonlinear locally stationary processes and the functional dependence measure to develop a unified causal inference framework for nonstationary processes by building on prior work [Sag+20; RGR22; Run+23; Run+19b; Run18a]. However, this line of inquiry requires a solution to the problem of post-selection inference [KKK22], as there does not yet exist a general solution analogous to sample splitting for the iid setting.

Second, we plan to explore topics related to time-varying regression estimation. It would be of interest to theoretically investigate the sieve estimator from Section 4 in the high-dimensional setting by introducing a sparsity-inducing penalty for regularization as in Zhang and Simon [ZS23]. It would also be of interest to develop a computationally efficient online estimation procedure for the sieve estimator by taking inspiration from Zhang and Simon [ZS22]. Along the way, we plan to theoretically investigate our subsampling cross-validation procedure from Section 5.1 for selecting the parameters of global estimators of time-varying regression functions. Additionally, it would be of interest to develop guarantees for time-varying nonlinear regression estimators in the context of processes with the total variation-type nonstationarity condition from Assumption 3.6. It may also be possible to investigate the convergence rates for deep neural network regression estimators as in Kurisu et al. [KFK25], but in the context of nonstationary processes with the functional dependence measure.

Third, there are several possible future research directions for conditional independence testing in this setting. While our test is based on the expected conditional covariance functional, our framework can easily be adapted to use any other functional equal to zero under the null of conditional independence. In particular, using higher-order functionals may be of interest in more complicated settings because the expected conditional covariance functional lacks sensitivity to nonlinear relationships and interactions; see Zhang and Janson [ZJ20] for more discussion. Specifically, it would be valuable to develop such tests without compromising on practicality, which is one of the key advantages of our

regression-based approach. In Section C.6, we discuss various conditional independence tests designed specifically for the locally stationary setting. However, those test statistics utilize kernel smoothing, so the resulting tests can be very sensitive to the choice of the bandwidth parameters.

C.2 Alternative test statistics

Consider the test statistic

$$S_{n,p}^*(\hat{\mathbf{R}}_n) = \left\| \frac{1}{\sqrt{T_{n,L}}} \sum_{t \in \mathcal{T}_{n,L}} \hat{\mathbf{R}}_{t,n} \right\|_p,$$

based on the ℓ_p -norm ($p \geq 2$) of the full sum of residual products. For example, we can use the test statistics

$$S_{n,\infty}^*(\hat{\mathbf{R}}_n) = \left\| \frac{1}{\sqrt{T_{n,L}}} \sum_{t \in \mathcal{T}_{n,L}} \hat{\mathbf{R}}_{t,n} \right\|_\infty, \quad S_{n,2}^*(\hat{\mathbf{R}}_n) = \left\| \frac{1}{\sqrt{T_{n,L}}} \sum_{t \in \mathcal{T}_{n,L}} \hat{\mathbf{R}}_{t,n} \right\|_2.$$

Crucially, the full sum test statistic $S_{n,p}^*(\hat{\mathbf{R}}_n)$ will not have power against alternatives in which the time-averages of the time-varying expected conditional covariances are close to zero (e.g. positive during the first half of times, and negative during the second half). On the other hand, the maximum partial sum test statistic $S_{n,p}(\hat{\mathbf{R}}_n)$ from (10) does have power against these alternatives. If the time-varying expected conditional covariances are suspected to consistently maintain the same sign (whether positive or negative), then users might be able to gain some power by using $S_{n,p}^*(\hat{\mathbf{R}}_n)$, although we emphasize that $S_{n,p}(\hat{\mathbf{R}}_n)$ will also have power against these alternatives. However, in settings where we have little prior knowledge about the time-varying expected conditional covariances between the nonstationary processes under alternatives, then the maximum partial sum test statistic $S_{n,p}(\hat{\mathbf{R}}_n)$ should be used because it has power against a wider range of alternatives. For similar reasons, we recommend using $S_{n,p}(\hat{\mathbf{R}}_n)$ when conducting automated multiple conditional independence testing (e.g. for screening out irrelevant time series in a large database of possible forecasting signals).

It is perhaps most intuitive to frame the problem in the following way. Consider the time-varying partially linear model

$$\mathbb{E}_P(Y_{t,n,j,b} | X_{t,n,i,a}, \mathbf{Z}_{t,n}) = \beta_{P,t,n,m} X_{t,n,i,a} + h_{P,t,n,j,b}(\mathbf{Z}_{t,n}),$$

for some function $h_{P,t,n,j,b}(\cdot)$. When the time-varying conditional expectation $\mathbb{E}_P(Y_{t,n,j,b} | X_{t,n,i,a}, \mathbf{Z}_{t,n})$ is assumed to have this time-varying partially linear form, the time-varying coefficient $\beta_{P,t,n,m}$ is equal to the expected conditional covariance of $X_{t,n,i,a}$ and $Y_{t,n,j,b}$ given $\mathbf{Z}_{t,n}$ divided by the expected conditional variance of $X_{t,n,i,a}$ given $\mathbf{Z}_{t,n}$; see Robins et al. [Rob+09] and Hines et al. [Hin+22] for more discussion. If domain knowledge suggests that the time-varying coefficients $(\beta_{P,t,n,m})_{m=(i,j,a,b) \in \mathcal{D}_n}$ consistently maintain the same sign over time $t \in \mathcal{T}_n$, then the full sum test statistic $S_{n,p}^*(\hat{\mathbf{R}}_n)$ can be used to gain some power. Otherwise, if we cannot make this assumption, then use the maximum partial sum test statistic $S_{n,p}(\hat{\mathbf{R}}_n)$ because it has power against a broader range of alternatives.

C.3 Cyclostationary processes

The general triangular array framework from Section 3 also allows for nonstationary processes that exhibit some form of repetition over time, such as periodic stationarity or cyclostationarity [Ben58; PP79; BHL94; Gar94; GNP06; Nap16b]. We emphasize that these types of nonstationary processes are not necessarily locally stationary. The theoretical justification of the dGCM test from Theorem 3.1 requires that we improve our estimates of the time-varying regression functions as n grows. See Remark 2.1 in Chen et al. [CSW22] and the preceding discussion about time-varying regression with periodic stationarity or cyclostationary processes. Also, see Section 2.5.1 of Bonnerjee et al. [BKW24] for a discussion of how strong Gaussian approximations for nonstationary nonlinear processes with causal representations as in Sections 3.1, 3.3, 3.4 can be used with cyclostationary processes. Ideally, we would like to be able to handle even more complex forms of nonstationarity than cyclostationarity. See Gardner et al. [GNP06] and Napolitano [Nap16a] for generalizations of this concept.

C.4 Simulation-and-regression for nonstationary processes

The general triangular array framework from Section 3 can also be used with simulation-and-regression approaches for estimating the time-varying regression functions. Suppose we have access to a black-box simulator which can be used to generate realistic paths of (X, Z) . Naturally, this simulation-based approach assumes that we either know the parameters of the simulator, or how to estimate them using an appropriate technique [CBL20; McF89; GMR93; FK21]. The main idea of this approach is to simulate s paths of (X, Z) , then fit separate regression models, such as XGBoost [CG16], LightGBM [Ke+17], or random forests [Bre01], for each time t by using the observations across the s iid simulated paths. To obtain the residuals for the time-varying regression of X on Z , the fitted regression models for each time t can be used with the observed realization of (X, Z) . The residuals for the time-varying regression of Y on Z can be obtained as in Section 4 or Section C.3.

The asymptotic arguments can be based on letting the number of simulations s grow with n , where n can be linked to the sample size (e.g. sampling frequency and/or duration of time) and number of dimensions. We can also allow n to be linked to the quality of the simulations, so that as n grows we can generate more realistic simulations—perhaps at a higher computational cost—and the simulator can be seen as converging in some sense to the true data generating process. Note that if the simulator can generate paths for (X, Y, Z) , then conditional independence tests which require multiple realizations of a nonstationary process, such as [Man+24; LSP22; Liu+23], can be used. In contrast, our proposed approach only requires a simulator for (X, Z) and a single realization of Y .

The main advantage of this simulation-based approach is that it leverages domain knowledge about (X, Z) to obtain better estimates of the time-varying regression functions without assuming anything about Y . For example, stochastic simulators can be used to generate paths of climatic variables, such as precipitation, surface temperature, surface water vapor, and ozone. Using the approach described above, we can identify conditional dependencies between these simulatable climatic variables (X, Z) and another process Y that is harder to model (e.g. commodity prices, consumer behavior, flu cases).

In this setting, it may also be possible to develop a conditional independence test based on simulation-based conditional density estimation using a model-X approach [Can+18; Liu+22; Niu+24; BCS20; HJ20; BJ22; SMR23; GHL24]. Nevertheless, the dGCM test can still be a very practical choice for this setting, particularly when it is much more feasible to do simulation-and-regression than simulation-based conditional density estimation. For instance, when the simulator’s computational demands make it impossible to generate a large number of paths for a high-dimensional process (X, Z) .

C.5 Simplifications under stationarity

Throughout this paper, we have completely avoided the assumption of stationarity. However, it is worth explaining how things would simplify if we are willing to assume that the processes are stationary. Overall, the takeaway is that the original GCM test from Shah and Peters [SP20] would require minimal modifications.

To begin, suppose we have $n \in \mathbb{N}$ observations of a stationary mixing time series, so that the regression functions are time-invariant. Further, suppose that the errors are iid. The statistical guarantees of many machine learning algorithms and statistical models, such as support vector machines [SA09; SHS09; HS14], random forests [Goe20; DN20], lasso [WLT20], and high-dimensional vector autoregressive models [WT23], have been studied in the context of stationary mixing time series with iid errors. Over the last decade, the literature on statistical learning theory for time series has been able to move beyond the restrictive assumptions of stationarity and mixing [Yu94; WLT20; KV02; ALW13] (or asymptotic stationarity [AD12]) by describing nonstationarity in terms of discrepancy measures [KM14; KM15; KM17; HY19; MK20]. This literature has recently considered new notions of learnability for general non-iid stochastic processes [DT22] and conditions under which learning from general non-iid stochastic processes is possible [Han21].

Zhang and Wu [ZW15] considered the setting in which the regression functions are time-varying and the errors are iid. Since the errors are iid, a multiplier bootstrap testing procedure can be justified by the Gaussian approximation from Chernozhukov et al. [CCK13] which was used by Shah and Peters [SP20]. Hence, the resulting test would be very similar to the original GCM test for the iid setting from Shah and Peters [SP20]. The main difference is that there can be time-lagged conditional dependencies in the stationary time series setting.

Suppose that the observed processes are temporally dependent (e.g. some form of mixing) and stationary so that the regression functions are time-invariant as before, but the errors are also temporally dependent. The guarantees of the lasso and vector autoregressive models are fairly well-studied in this setting. Basu and Michailidis [BM15] investigated high-dimensional vector autoregressive models with serially correlated errors. Gupta [Gup12] and Xie and Xiao [XX18] studied the lasso with errors satisfying various weak dependence conditions. Peng et al. [PZZ23] and Xie et al. [XXY17] studied the lasso with ϕ -mixing and β -mixing errors, respectively. Wu and Wu [WW16] studied the guarantees of the lasso in the setting with temporally dependent errors by using the functional dependence measure of Wu [Wu05].

In the serially correlated error setting, the key difference with the GCM test from Shah and Peters [SP20] is that one must use a suitable Gaussian approximation result to justify a multiplier bootstrap-type testing procedure. See Chang et al. [CCW24] for a comprehensive overview of Gaussian approximations for dependent data. Chernozhukov et al. [CCK19] investigated a block multiplier bootstrap under a β -mixing assumption, and Zhang and Cheng [ZC14] explored a wild multiplier bootstrap under the functional dependence measure of Wu [Wu05]. Also, Zhang and Wu [ZW17] discuss estimators for the long-run covariance matrix so that their Gaussian approximation for high-dimensional time series can be applied in practice. See Wu and Xiao [WX12] and Wu [Wu11] for more discussion about long-run covariance matrix estimation for stationary time series.

One could also have time-invariant regression functions with errors that are nonstationary and temporally dependent. For instance, Xia et al. [XCG24] studies the lasso with locally stationary errors. However, the statistical guarantees of other machine learning algorithms and statistical models have not been studied in this setting. If the process of error products is mean-nonstationary (i.e. time-varying expected conditional covariance) under alternatives, then the same test statistics from Section 2.4 can be used. Otherwise, if domain knowledge suggests that the time-varying expected conditional covariances usually maintain the same sign, then the test statistics from Section C.2 can be used.

To recap, we considered how the assumption of stationarity would vastly simplify the problem. We find that the original GCM test for the iid setting from Shah and Peters [SP20] can be adapted to the stationary time series setting by making the previously mentioned changes. In contrast, we consider the much more complicated setting in which the observed processes can be nonstationary and temporally dependent, the regression functions can vary over time, and the error processes can be nonstationary and temporally dependent. We emphasize that our dGCM test can be used with stationary processes and iid sequences, which are special cases of the general framework from Section 3.

C.6 Additional tests for locally stationary processes

We discuss three conditional independence tests for locally stationary processes that we did not pursue in this paper. Crucially, the test statistics below require local long-run covariance estimation. Most local long-run covariance estimators use kernel smoothing, and therefore require selecting bandwidths. Unfortunately, test statistics that use kernel smoothing can be very sensitive to the choice of the bandwidths, which can be hard to select in practice. Inspired by the success of bandwidth-free approaches in other areas of time series analysis [Lob01; Sha10; RS13; Sha15; ZS24], we designed the dGCM test so that it does not require local long-run covariance estimation and therefore avoids kernel smoothing.

Recall the notation for the locally stationary setting from Section 4. To begin, let us translate the “weak” conditional independence criterion of Daudin [Dau80] into the locally stationary setting as follows. For some $n \in \mathbb{N}$, $u \in \mathcal{U}_n$, $(i, j, a, b) \in \mathcal{D}_n$, $t \in \mathcal{T}_n$, if

$$\tilde{X}_{[un],n,i,a}(u) \perp\!\!\!\perp \tilde{Y}_{[un],n,j,b}(u) \mid \tilde{Z}_{[un],n}(u),$$

then

$$\mathbb{E}_P[\phi(\tilde{X}_{[un],n,i,a}(u), \tilde{Z}_{[un],n}(u)) \varphi(\tilde{Y}_{[un],n,j,b}(u), \tilde{Z}_{[un],n}(u))] = 0,$$

for all functions

$$\phi \in L^2_{\tilde{X}_{[un],n,i,a}(u), \tilde{Z}_{[un],n}(u)}, \quad \varphi \in L^2_{\tilde{Y}_{[un],n,j,b}(u), \tilde{Z}_{[un],n}(u)},$$

such that

$$\begin{aligned} \mathbb{E}_P[\phi(\tilde{X}_{[un],n,i,a}(u), \tilde{Z}_{[un],n}(u)) \mid \tilde{Z}_{[un],n}(u)] &= 0, \\ \mathbb{E}_P[\varphi(\tilde{Y}_{[un],n,j,b}(u), \tilde{Z}_{[un],n}(u)) \mid \tilde{Z}_{[un],n}(u)] &= 0. \end{aligned}$$

Hence, the corresponding local expected conditional covariance

$$\rho_{P,n,m}(u) = \mathbb{E}_P[\text{Cov}_P(\tilde{X}_{[un],n,i,a}(u), \tilde{Y}_{[un],n,j,b}(u) | \tilde{Z}_{[un],n}(u))],$$

is equal to zero for $m = (i, j, a, b) \in \mathcal{D}_n$.

First, consider the global null hypothesis of conditional independence

$$\tilde{X}_{[un],n,i,a}(u) \perp\!\!\!\perp \tilde{Y}_{[un],n,j,b}(u) \mid \tilde{Z}_{[un],n}(u) \text{ for all } u \in \mathcal{U}_n, \text{ for all } (i, j, a, b) \in \mathcal{D}_n. \quad (23)$$

In the univariate setting, \mathcal{D}_n simply consists of one dimension/time-offset tuple as in Section 2.2. Also, note that this hypothesis can be extended to the group of time series setting as discussed in Section 2.2. Note that the null hypothesis (23) implies the null hypothesis (14), so the process of error products from the time-varying nonlinear regressions of $(X_{t,n,i,a})_{t \in \mathcal{T}_n}$ on $(Z_{t,n})_{t \in \mathcal{T}_n}$ and $(Y_{t,n,j,b})_{t \in \mathcal{T}_n}$ on $(Z_{t,n})_{t \in \mathcal{T}_n}$ will still have mean zero as in Section 4. To test for the null hypothesis (23), we could, for example, use the test statistic

$$\sup_{u \in \mathcal{U}_n} \left\| \frac{1}{\sqrt{T_n}} \sum_{t \in \mathbb{T}_n^-} (\hat{\Sigma}_{t,n}^R)^{-1/2} \hat{R}_{t,n} \right\|_p,$$

based on some ℓ_p norm ($p \geq 2$) of the studentized partial sum process, where $\hat{\Sigma}_{t,n}^R$ is an estimate of the local long-run covariance matrix at time t . The theoretical guarantees for the test based on this test statistic would utilize the recent results from Mies [Mie24] about strong Gaussian approximations with random multipliers.

Second, it is possible to develop a test for the local null hypothesis of conditional independence

$$\tilde{X}_{[un],n,i,a}(u) \perp\!\!\!\perp \tilde{Y}_{[un],n,j,b}(u) \mid \tilde{Z}_{[un],n}(u) \text{ for all } (i, j, a, b) \in \mathcal{D}_n, \quad (24)$$

for a particular rescaled time $u \in \mathcal{U}_n$ (i.e. instead of for all \mathcal{U}_n as in (23)) by using, for example, the test statistic

$$\max_{m=(i,j,a,b) \in \mathcal{D}_n} \left| \frac{1}{\sqrt{T_n h_{n,m}}} \sum_{t \in \mathcal{T}_n} K\left(\frac{t/n - u}{h_n}\right) \hat{R}_{t,n,m} \right| / \hat{\sigma}_{n,m}^R(u),$$

for some bandwidths $h_{n,m} \rightarrow 0$ and local long-run variance estimates $(\hat{\sigma}_{n,m}^R(u))^2$. The main idea for this local conditional independence test is that since $\mathbb{E}_P(\tilde{R}_{P,[un],n,m}(u)) = 0$ under the null and the process of error products is “approximately stationary” over short periods of time, we can expect that the means of $R_{P,t,n,m} = \tilde{R}_{P,t,n,m}(t/n)$ for rescaled times t/n near u are also close to zero under the null. We can make this mathematically precise by using the technical tools developed for nonlinear locally stationary processes from Dahlhaus et al. [DRW19].

Third, it is also possible to simultaneously test whether conditional independence

$$\tilde{X}_{[un],n,i,a}(u) \perp\!\!\!\perp \tilde{Y}_{[un],n,j,b}(u) \mid \tilde{Z}_{[un],n}(u) \text{ for all } (i, j, a, b) \in \mathcal{D}_n, \quad (25)$$

holds at each rescaled time $u \in \mathcal{U}_n$ (i.e. instead of for a particular $u \in \mathcal{U}_n$ as in (24)). This can be done by creating simultaneous confidence bands (i.e. over time) for expected conditional covariance curves $(\rho_{P,n,m}(u))_{u \in \mathcal{U}_n}$ for each $m = (i, j, a, b) \in \mathcal{D}_n$. Depending on whether or not estimates of the local long-run variances $(\hat{\sigma}_{n,m}^R(u))^2$ are used, these simultaneous confidence bands will have time-varying or time-invariant widths, respectively. The main idea is that the local null hypothesis of conditional independence at rescaled time $u \in \mathcal{U}_n$ for some dimension/time-offset tuple $m = (i, j, a, b) \in \mathcal{D}_n$ can be rejected if zero is not included in the corresponding confidence interval for the local expected conditional covariance $\rho_{P,n,m}(u)$. This can be done using similar arguments as Bai and Wu [BW23], which focuses on inferring time-varying correlation curves. However, due to the problem of post-selection inference [KKK22], this would require either stronger assumptions (e.g. Donsker-type), data decomposition techniques (e.g. splitting, fission, or thinning) for nonstationary time series, or two independent realizations of the same nonstationary process—rarely possible outside of experimental settings.

An approach for inferring expected conditional covariance curves would have a range of applications outside of testing for conditional independence, since this functional frequently appears in the causal

inference literature [Ken24; Rob+08; Rob+09; Li+11; Rob+17; NR18]. We suspect that similar approaches can be used to infer curves based on other functionals of interest in causal inference. Hence, this line of work would be of significant interest to the emerging field of time series causal inference [Sag+20; RGR22; Run+23; Run+19b; Run18a; WNR24]. Lastly, we note that it may be possible to extend the tests discussed here to the piecewise locally stationary setting, however we leave the details for future work.

C.7 Piecewise locally stationary processes

We briefly describe how to extend the Sieve-dGCM test from Section 4 from locally stationary processes [Dah97; ZW09; Dah12; DRW19] to a more general class of nonstationary processes known as piecewise locally stationary (PLS) processes introduced in Zhou [Zho13]. Specifically, the class of PLS processes generalizes the stochastic Lipschitz nonstationarity condition from Assumption 4.6 by allowing for finitely many breakpoints [Zho13; WZ24; DWZ19]. We emphasize that PLS processes are included in the even more general class of nonstationary processes from Section 3 with the total variation-type nonstationarity condition from Assumption 3.6.

The main idea is to identify the breakpoints, fit a separate sieve model on each locally stationary segment, and run Algorithm 1 on all the residuals. If the breakpoints are known exactly, then the same arguments can be used to show that the sieve time-varying regression estimators achieve the required convergence rates (i.e. within each locally stationary segment). If the breakpoints must be identified, then our arguments must be extended to account for this. As far as we know, Wu and Zhou [WZ24] is the most relevant work on identifying breakpoints for PLS processes. We leave the full details of this extension for future work.

C.8 Weakening the assumptions on the error processes

In Assumption 3.5, we assume that there are distribution-uniform upper bounds on the L^∞ norms and L^∞ functional dependence measures of the error processes. We use this assumption to show inequality (21) in the proof of Theorem 3.1. Afterwards, we use the time-uniform convergence rates for the time-varying regression estimators to show Step 1.2 in the proof of Theorem 3.1. It is possible to weaken the assumptions imposed on the error processes by making stronger assumptions about the time-varying regression estimators, or more complicated assumptions about the terms in (21). Instead, we opt for simpler assumptions on the errors and prediction errors for the sake of transparency.

Lastly, we note that the Sieve-dGCM test from Section 4 performs well even when the error processes violate Assumption 3.5, at least in the settings we considered for our simulations in Section 5. To satisfy Assumption 3.5, we can make minor modifications to the data generating processes in Section 5. For example, by replacing the Gaussian error processes with, say, truncated Gaussian error processes.

C.9 A note on causal discovery for nonstationary time series

A common practice in time series analysis is to difference a time series until the augmented Dickey-Fuller (ADF) test says it is stationary. One may presume that a similar approach can be taken in the context of causal discovery for time series. That is, applying a causal discovery algorithm for stationary time series to a transformed (e.g. d -times differenced) time series in the process of identifying a causal graph for the originally nonstationary time series. Unfortunately, as discussed in Malinsky and Spirtes [MS19], even differencing once can induce spurious correlations among the time series. Consequently, the resultant causal graph for the original time series can be wildly incorrect.

In contrast, causal discovery algorithms based on conditional independence tests for nonstationary time series (e.g. our dGCM test) can identify more interpretable causal graphs because the time series need not be stationary. Instead, we can consider “locally stationary transformations” of the original time series, such as log growth rates, percent changes, or single differences. In other words, users of dGCM need not assume that the transformed time series is stationary over the entire time period, only that it is approximately stationary over short time windows. In practice, this means that causal graphs identified by dGCM-based causal discovery algorithms can be easier to interpret, because they can be constructed for locally stationary transformations rather than stationary transformations (e.g. d -times differenced time series). To streamline the development of dGCM-based causal discovery algorithms

for nonstationary nonlinear time series, we explain how to use our framework for (unconditional) independence testing in Section C.10.

C.10 Independence testing with nonstationary nonlinear time series

The theoretical framework in Section 3 can also be used for (unconditional) independence testing between multivariate nonstationary nonlinear time series. This independence test works for the practical setting in which only one realization of the process (X, Y) is observed. Naturally, this corresponds to the special case in which the covariate process is taken to be a constant, so that each of the σ -algebras generated by the covariates $\mathbf{Z}_{t,n}$ at times $t \in \mathcal{T}_n$ is simply the trivial σ -algebra. Theorem 3.1 states that this independence test will have uniformly asymptotic Type-I error control.

Recall the variety of null hypotheses of conditional independence from Section 2.2. Analogously, the independence test is for the unconditional versions of these null hypotheses. Specifically, the univariate test is for the null hypothesis

$$X_{t,n,i,a} \perp\!\!\!\perp Y_{t,n,j,b} \text{ for all } t \in \mathcal{T}_n,$$

for one dimension/time-offset tuple $(i, j, a, b) \in \mathcal{D}_n$, and the multivariate test is for the null hypothesis

$$X_{t,n,i,a} \perp\!\!\!\perp Y_{t,n,j,b} \text{ for all } t \in \mathcal{T}_n, \text{ for all } (i, j, a, b) \in \mathcal{D}_n,$$

or for the “groups of time series” setting with multiple locations as in Section 2.2.

The independence test is based on the fact that under the null hypothesis, the corresponding combinations of dimensions and time-offsets of X, Y will have zero covariance. The practical implementation of this independence test is given by Algorithm 1, where we use estimates of the unconditional means at each time rather than estimates of the conditional means at each time.

In the context of the sieve estimator from Section 4, the unconditional means of the processes X and Y at each time can be estimated by using only one basis function for the (constant) covariate value and multiple basis functions for time as usual. Theorem 4.1 provides a theoretical guarantee for this independence test based on using the sieve estimator for estimating the unconditional means at each time. For estimators based on kernel smoothing, we refer readers to Dahlhaus et al. [DRW19].

C.11 Granger causality

Throughout this paper, we have discussed causality as in Pearl [Pea09] and causal discovery as in Spirtes et al. [SGS01]. However, most economists are more familiar with the concept of Granger causality, particularly in the stationary linear time series setting. Therefore, it may be helpful to review different notions of Granger causality in the nonstationary nonlinear time series setting. To do so, we provide a highly condensed summary of Sections 2 and 4 in Shojaie and Fox [SF22].

The original definition of Granger causality from Granger [Gra69] is about prediction. Informally, a process X is said to be Granger noncausal of another process Y if, for all times t , the variance of the error from the optimal prediction of Y_t based on all relevant information up to time $t - 1$ is not reduced by including the history of X up to time $t - 1$. See Section 2 in Shojaie and Fox [SF22] for the exact definition and the stringent conditions under which this predictive definition corresponds to genuine causality as in Pearl [Pea09]. While this original definition does not assume linear dynamics, much of the following methodology revolves around the identification of coefficients in linear vector autoregressive (VAR) models with p time series [Gra80; Lüt05; BSM15].

Another definition of Granger causality, referred to as *strong* Granger causality [FM82], is stated in terms of conditional independence relationships among stochastic processes. Let $(X^i)_{i \in [p]}$ be p signals used to predict the target Y . The process X^i is said to be (strongly) Granger noncausal of Y if, for all times t , Y_t is conditionally independent of the history of the signal X^i up to time $t - 1$ given the history of the other signals $(X^j)_{j \in [p] \setminus \{i\}}$ up to time $t - 1$. See Definition 2 in Shojaie and Fox [SF22] for the exact definition, and the rest of Section 4 therein for more discussion.

Notably, Eichler [Eic12] introduced a comprehensive graphical modeling framework for time series based on strong Granger causality, which can be detected using conditional independence tests for nonlinear time series [SP11; SW07; BRT12; SW21; ZZZ22]. In a similar vein, our proposed conditional independence test can be used to detect strong Granger causality for nonlinear time series with time-varying dynamics. This can be incorporated into graphical modeling frameworks for nonstationary nonlinear time series, analogous to Basu and Rao [BR23].

There are various techniques for assessing nonlinear Granger causality that do not use conditional independence testing. For instance, the neural Granger causality method from Tank et al. [Tan+21] extracts Granger causal structures by using sparsity-inducing penalties on the weights of structured multilayer perceptrons (MLPs) and recurrent neural networks (RNNs). Additionally, there is an influential strand of literature connecting Granger causality and directed information theory [AM12; QKC15]. See Section 4 of Shojaie and Fox [SF22] for more discussion of nonlinear Granger causality.

C.12 All test results for the real data application

The following table displays the results of each hypothesis test for independence and conditional independence for the daily log returns of the stock market indices as described in Section 6. For the sake of simplicity, we base the conditional independence graph in Section 6 on the unadjusted p-values. For better or worse, reporting graphs based on unadjusted p-values is a relatively common practice in causal discovery.

Null Hypothesis	P-value	Adjusted P-value
$S\&P(t) \perp\!\!\!\perp FTSE(t)$ for all times $t \in \mathcal{T}_n$	0.024	0.030
$S\&P(t) \perp\!\!\!\perp HangSeng(t)$ for all times $t \in \mathcal{T}_n$	0.004	0.008
$S\&P(t) \perp\!\!\!\perp Nikkei(t)$ for all times $t \in \mathcal{T}_n$	0.042	0.051
$FTSE(t) \perp\!\!\!\perp HangSeng(t)$ for all times $t \in \mathcal{T}_n$	0.002	0.006
$FTSE(t) \perp\!\!\!\perp Nikkei(t)$ for all times $t \in \mathcal{T}_n$	0.002	0.006
$HangSeng(t) \perp\!\!\!\perp Nikkei(t)$ for all times $t \in \mathcal{T}_n$	8.000×10^{-4}	0.004
$S\&P(t) \perp\!\!\!\perp FTSE(t) \mid HangSeng(t)$ for all times $t \in \mathcal{T}_n$	2.000×10^{-4}	0.004
$S\&P(t) \perp\!\!\!\perp FTSE(t) \mid Nikkei(t)$ for all times $t \in \mathcal{T}_n$	4.000×10^{-4}	0.004
$S\&P(t) \perp\!\!\!\perp HangSeng(t) \mid FTSE(t)$ for all times $t \in \mathcal{T}_n$	0.529	0.529
$S\&P(t) \perp\!\!\!\perp HangSeng(t) \mid Nikkei(t)$ for all times $t \in \mathcal{T}_n$	0.008	0.012
$S\&P(t) \perp\!\!\!\perp Nikkei(t) \mid FTSE(t)$ for all times $t \in \mathcal{T}_n$	0.128	0.135
$S\&P(t) \perp\!\!\!\perp Nikkei(t) \mid HangSeng(t)$ for all times $t \in \mathcal{T}_n$	0.103	0.116
$FTSE(t) \perp\!\!\!\perp HangSeng(t) \mid S\&P(t)$ for all times $t \in \mathcal{T}_n$	0.003	0.006
$FTSE(t) \perp\!\!\!\perp HangSeng(t) \mid Nikkei(t)$ for all times $t \in \mathcal{T}_n$	0.005	0.008
$FTSE(t) \perp\!\!\!\perp Nikkei(t) \mid S\&P(t)$ for all times $t \in \mathcal{T}_n$	8.000×10^{-4}	0.004
$FTSE(t) \perp\!\!\!\perp Nikkei(t) \mid HangSeng(t)$ for all times $t \in \mathcal{T}_n$	0.006	0.009
$HangSeng(t) \perp\!\!\!\perp Nikkei(t) \mid S\&P(t)$ for all times $t \in \mathcal{T}_n$	0.002	0.006
$HangSeng(t) \perp\!\!\!\perp Nikkei(t) \mid FTSE(t)$ for all times $t \in \mathcal{T}_n$	0.005	0.008

Table 1: P-values and adjusted p-values from the Benjamini-Hochberg method [BH95] for each test.

C.13 Prior work on distribution-uniform inference

We briefly review prior work on distribution-uniform inference. First, we discuss the conditional independence testing literature. Recently, there has been a great deal of work on distribution-uniform conditional independence testing frameworks due to the hardness result and conditional independence testing framework from Shah and Peters [SP20]. For instance, Lundborg et al. [LSP22] introduced many distribution-uniform convergence results for separable Banach and Hilbert spaces. Recently, Christgau et al. [CPH22] introduced a distribution-uniform “conditional local independence” testing framework for the setting in which n realizations of a point process are observed. Christgau et al. [CPH22] also introduce a distribution-uniform version of Rebolledo’s martingale central limit theorem [Reb80] and extend many distribution-uniform convergence results from Lundborg et al. [LSP22] to metric spaces.

Second, we mention some relevant work from the literature on anytime-valid inference. Recently, Waudby-Smith and Ramdas [WR23] introduced a distribution-uniform strong (almost-sure) Gaussian approximation for the full sum of iid random variables. The work in Waudby-Smith and Ramdas [WR23] is motivated by prior work on asymptotic anytime-valid inference from Waudby-Smith et al. [Wau+24], in which the authors defined the concept of an “asymptotic confidence sequence”. In particular, Waudby-Smith et al. [Wau+24] introduced asymptotic confidence sequences for iid random

variables and a Lindeberg-type asymptotic confidence sequence which can capture time-varying means under martingale dependence.

Third, we discuss other areas in which distribution-uniform inference is studied under different names. There is a vast literature discussing the importance of distribution-uniform inference under the name of “honest” or “uniform” inference, see [Li89; Kas18; Tib+18; RWG19; KBW23]. Also, there is a plethora of literature on distribution-uniform moment inequality testing [IM04; RS08; AG09; AS10; AB12; RSW14]. Most recently, Li et al. [LLZ22] developed a distribution-uniform test for general functional inequalities which admits conditional moment inequalities as a special case. In their supplementary appendix, Li et al. [LLZ22] introduce a distribution-uniform strong Gaussian approximation for the full sum of a high-dimensional mixingale.

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