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TESTING FOR SERIAL CORRELATION OF UNKNOWN FORM USING WAVELET METHODS

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A wavelet-based consistent test for serial correlation of unknown form is proposed. As a spatially adaptive estimation method, wavelets can effectively detect local features such as peaks and spikes in a spectral density, which can arise as a result of strong autocorrelation or seasonal or business cycle periodicities in economic and financial time series. The proposed test statistic is constructed by comparing a wavelet-based spectral density estimator and the null spectral density. It is asymptotically one-sided N(0,1) under the null hypothesis of no serial correlation and is consistent against serial correlation of unknown form. The test is expected to have better power than a kernel-based test (e.g., Hong, 1996, *Econometrica* 64, 837–864) when the true spectral density has significant spatial inhomogeneity. This is confirmed in a simulation study. Because the spectral densities of time series arising in practice usually have unknown smoothness, the wavelet-based test is a useful complement to the kernel-based test in practice.

1. INTRODUCTION

Wavelet analysis originated as a new analytic method alternative to Fourier analysis in signal analysis and has rapidly grown through interactions with mathematics over the last decade or so. As a spatially adaptive analytic method, wavelets provide a new useful tool for nonparametric function estimation. Because wavelets are spatially varying orthonormal bases with two parameters—scale and translation—they are fundamentally different from the Fourier basis or Gabor basis (i.e., the windowed Fourier basis) and have some appealing statistical advantages over traditional estimators such as kernel and spline methods in esti-

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mating a function with unknown smoothness (e.g., Donoho and Johnstone, 1994, 1995a, 1995b; Donoho et al., 1996). In particular, wavelets have good time-frequency localization properties. Both smooth and nonsmooth functions can be effectively reconstructed using wavelet bases.

Although the development of wavelet analysis has been rapidly growing, its application to time series analysis is relatively sparse. In time series spectral analysis, Gao (1993) uses wavelet shrinkage to estimate the spectral density of a stationary Gaussian process. Neumann (1996) derives the asymptotic normal distribution of empirical wavelet coefficients used to estimate the spectral density of a non-Gaussian process. He shows in simulations that wavelet estimators outperform kernel estimators in capturing such spatially inhomogeneous features as peaks and spikes in the spectral density. In another development, Priestley (1996) provides useful links between wavelet analysis and nonstationary evolutionary spectral analysis. See also Subba Rao and Indukumar (1996) for wavelet application to nonlinear and nonstationary time series.

There have been some applications of wavelet analysis to economics and econometrics. Goffe (1994) illustrates the application of the wavelet method to some nonstationary macroeconomic time series. Gilbert (1995) uses wavelets to estimate and test structural changes. Jensen (2000) proposes a wavelet-based algorithm to estimate a long memory model via the maximum likelihood estimation method. Wang (1995) applies the wavelet method to detect jumps and sharp cusps in stock market returns. Ramsey and his coauthors, in a series of papers (e.g., Ramsey, 1999; Ramsey and Lampart, 1998a, 1998b; Ramsey and Zhang, 1996, 1997; Ramsey, Usikov, and Zaslavsky, 1995), apply wavelets to various economic and financial time series and obtain some interesting results.

In this paper, we illustrate how wavelets can be used to effectively detect serial correlation of unknown form. Detection and inference of serial correlation have long been of interest in time series analysis (e.g., Anderson, 1993; Andrews and Ploberger, 1996; Box and Pierce, 1970; Durbin and Watson, 1950, 1951; Durlauf, 1991; Godfrey, 1978a, 1978b; Hong, 1996; Robinson, 1991; Whang, 1997). Among various existing tests for serial correlation, Hong (1996) proposes a consistent test for serial correlation using a Parzen (1957) kernel estimator for the spectral density of a stationary time series. The test is shown to have good power against both short and long memory processes. However, spatially nonadaptive estimation methods such as the kernel method cannot effectively detect spatially varying local features (e.g., kinks, peaks, or jumps). It is well known, for example, that kernel estimators tend to underestimate a mode in a spectral density (e.g., Priestley, 1981). Thus, it is expected that a kernelbased test may have relatively poor power when the spectral density has significant spatial inhomogeneity. Nonsmooth spectral densities with kinks, peaks, or spikes are not uncommon for time series arising in practice. They can arise as a result of strong autocorrelation or seasonal or business cycle periodicities in time series (e.g., Wen, 1998). It is therefore important to develop test procedures with good power against these alternatives. The wavelet method is particularly suitable for such a purpose. Here we propose a new consistent test for serial correlation using wavelets. The test is constructed by comparing a waveletbased spectral density estimator and the null spectral density. We establish the asymptotic theory for the proposed test. The null distribution of the test statistic is asymptotically one-sided N(0,1). No formulation of an alternative model is required, and the test is consistent against serial correlation of unknown form.

A simulation study compares the proposed wavelet-based test with the kernelbased test of Hong (1996). It confirms our conjecture that the wavelet method outperforms the kernel method in detecting spatially inhomogeneous spectral features. When the spectral density has distinctive peaks or spikes, the waveletbased test has better power than the kernel-based test. On the other hand, when the spectral density is smooth, the kernel-based test performs better. Because the power of the tests depends on the shape of the unknown spectrum, the proposed test is a useful complement to the kernel-based test in practice.

In Section 2, we introduce the wavelet framework and construct the test statistic. The null asymptotic normality is derived in Section 3, and consistency is established in Section 4. Section 5 presents a simulation study on the finite sample performances of the proposed test and the kernel-based test of Hong (1996). All the proofs are given in the Appendixes. Throughout, A^* denotes the complex conjugate of A; Re(A) the real part of A; $\mathbb{Z} = \{0, \pm 1, ...\}$ and $\mathbb{Z}^+ = \{1, 2, ...\}$ the sets of integers, and positive integers, respectively; $C \in (0, \infty)$ a generic constant that may differ from place to place. All convergences are taken as the sample size $n \to \infty$.

2. BASIC FRAMEWORK AND TEST STATISTICS

2.1. Wavelet Analysis

Let $\psi : \mathbb{R} \to \mathbb{R}$ be an orthonormal wavelet such that the doubly infinite sequence $\{\psi_{jk}(x) = 2^{j/2}\psi(2^{j}x - k)\}$ forms a complete orthonormal basis of the $L_2(\mathbb{R})$ -space of square integrable functions, where j and $k \in \mathbb{Z}$ are integers corresponding to scale (dilation or compression) and translation (displacement), respectively. Any function $f(x) \in L_2(\mathbb{R})$ can be expressed as a sum of wavelets $\{\psi_{jk}(\cdot)\}$, which are generated from the single function $\psi(\cdot)$, the socalled mother wavelet. For excellent intuitive accounts of wavelet analysis, see, for example, Priestley (1996) and Ramsey (1998).

Throughout, we consider *multiresolution analysis*, which is the most commonly used analytic method in the wavelet literature (cf. Daubechies, 1992; Hernández and Weiss, 1996; Priestley, 1996; Strang and Nguyen, 1996).

DEFINITION. A multiresolution analysis is a sequence of subspaces $\{V_j, j \in \mathbb{Z}\}$ of $L_2(\mathbb{R})$ satisfying the following requirements:

- (i) $V_j \subset V_{j+1}$ and $\cap V_j = \{\mathbf{0}\}, \ \overline{\cap V_j} = L_2;$
- (ii) $f(x) \in V_j$ if and only if $f(2x) \in V_{j+1}$;

- (iii) $f(x) \in V_j$ if and only if $f(x k) \in V_j$, for all $k \in \mathbb{Z}$;
- (iv) V_0 has an orthonormal basis $\{\phi(\cdot k), k \in \mathbb{Z}\}$, where $\phi: \mathbb{R} \to \mathbb{R}$ such that $\int_{-\infty}^{\infty} \phi(x) dx = 1$.

From the nested structure of subspace V_j , the orthogonal complement W_j of V_j can be defined as $V_j \bigoplus W_j = V_{j+1}$, where \bigoplus denotes the orthogonal sum. Let V_0 be the initial subspace; then

$$V_0 \bigoplus \sum_{j=0}^{\infty} W_j = L_2(\mathbb{R}).$$

Through dilation and translation of the scale function $\phi(\cdot)$ (also called father wavelet) and mother wavelet $\psi(\cdot)$, the sequences $\{\phi_{ik}(x) = 2^{j/2}\phi(2^{j}x - k)\}$ and $\{\psi_{ik}(x) = 2^{j/2}\psi(2^{j}x - k)\}$ constitute a complete orthonormal basis of V_{i} and W_j , respectively. Each subspace V_j encodes the information of the signal at resolution level j, which can be represented by scale functions $\{\phi_{ik}(\cdot), k \in \mathbb{Z}\}$. Each subspace W_i orthogonal to V_i encodes the details, namely, the difference of the information between the signals seen at two resolutions V_i and V_{i+1} . Details at level j can be represented by wavelets $\{\psi_{ik}(\cdot), k \in \mathbb{Z}\}$. Thus, signals at level j combined with details at level j provide signals at level j + 1. Intuitively, a small j or a low resolution level can capture smooth components of the signal, whereas a large *j* or a high resolution level can capture variable components of the signal. Moreover, given a resolution level i, various values of translation parameter k allow one to capture local features of the signal. In our application, the scale function $\phi(\cdot)$ will capture the smoothest component of the spectral density, whereas the wavelets $\{\psi_{ik}(\cdot)\}$ will capture the differences such as peaks and spikes. We note that requirement (i) implies that (a) the signal seen at a given resolution level contains all the information of the signal seen at coarser resolution levels and (b) any function in $L_2(\mathbb{R})$ can be approximated arbitrarily well by a sufficiently fine resolution (i.e., a sufficiently large *j*). Requirements (ii) and (iii) represent scale and shift invariance, respectively.

We first impose a standard condition on the mother wavelet $\psi(\cdot)$.

Assumption 1. $\psi : \mathbb{R} \to \mathbb{R}$ is an orthonormal wavelet such that $\int_{-\infty}^{\infty} \psi(x) dx = 0$, $\int_{-\infty}^{\infty} |\psi(x)| dx < \infty$, $\int_{-\infty}^{\infty} \psi(x) \psi(x - k) dx = 0$ for all $k \in \mathbb{Z}, k \neq 0$, and $\int_{-\infty}^{\infty} \psi^2(x) dx = 1$.

The orthonormality of $\psi(\cdot)$ implies that the doubly infinite sequence $\{\psi_{jk}(\cdot)\}$ constitutes an orthonormal basis for $L_2(\mathbb{R})$, that is,

$$\int_{-\infty}^{\infty}\psi_{jk}(x)\psi_{lm}(x)dx=\delta_{jl}\delta_{km}, \quad j,l,k,m\in\mathbb{Z},$$

where $\delta_{jl} = 1$ if j = l and $\delta_{jl} = 0$ otherwise (cf. Daubechies, 1992). Assumption 1 ensures that the Fourier transform of $\psi(\cdot)$ defined by

$$\hat{\psi}(\omega) = (2\pi)^{-1/2} \int_{-\infty}^{\infty} \psi(x) e^{-i\omega x} dx, \quad i = \sqrt{-1}$$

exists and is continuous in ω almost everywhere. Note that $\hat{\psi}(0) = (2\pi)^{-1/2} \times \int_{-\infty}^{\infty} \psi(x) dx = 0$, which implies that the mother wavelet $\psi(\cdot)$ must have alternating signs. This is one of the characteristic properties of wavelets and a reason why wavelets are sensitive to changes or singularities. Note also that $\hat{\psi}^*(\omega) = \hat{\psi}(-\omega)$ for all $\omega \in \mathbb{R}$.

Most orthonormal wavelets $\psi(\cdot)$ are constructed from a father wavelet $\phi(\cdot)$. The mother wavelet $\psi(\cdot)$ can have bounded or unbounded support. A well-known compactly supported wavelet is the Haar wavelet,

$$\psi(x) = \begin{cases} 1, & 0 \le x < 1/2 \\ -1, & 1/2 \le x < 1 \\ 0 & \text{otherwise.} \end{cases}$$
(1)

It is generated from a linear combination of $\phi(x) = 1 (0 \le x < 1)$, where $1(\cdot)$ is the indicator function. The sequence $\{\psi_{jk}(x) = 2^{j/2}\psi(2^{j}x - k), j, k \in \mathbb{Z}\}$ forms a complete orthonormal basis of $L_2(\mathbb{R})$. (See, e.g., Hernández and Weiss, 1996, pp. 59–61.)

An example of wavelets with unbounded support is the Shannon wavelet

$$\psi(x) = -2 \frac{\sin(2\pi x) + \cos(\pi x)}{\pi(2x+1)}, \quad x \in \mathbb{R}.$$
 (2)

This is generated from the scaling function $\phi(x) = \frac{\sin(\pi x)}{(\pi x)}$. See Hernández and Weiss (1996, pp. 61–62) for more discussion.

We impose an additional condition for good localization of $\psi(\cdot)$ in the frequency domain.

Assumption 2.

- (i) $|\hat{\psi}(\omega)| \le C(1+|\omega|)^{-\alpha}$ for some $\alpha > \frac{3}{2}$ and some constant $C \in (0,\infty)$;
- (ii) $\hat{\psi}(\omega) = e^{i\omega/2}b(\omega)$ or $\hat{\psi}(\omega) = -ie^{i\omega/2}b(\omega)$, where $b(\cdot)$ is a real-valued function.

Most commonly used wavelets satisfy these conditions (cf. Hernández and Weiss, 1996). For example, the Lemarie–Meyer family of wavelets is of the form $\hat{\psi}(\omega) = e^{i\omega/2}b(\omega)$, where $b(\cdot)$ is a real-valued and symmetric function on \mathbb{R} . An example of this family is the Meyer wavelet, defined via the Fourier transforms of $\phi(\cdot)$ and $\psi(\cdot)$:

$$\hat{\phi}(\omega) = \begin{cases} (2\pi)^{-1/2}, & |\omega| \le 2\pi/3\\ (2\pi)^{-1/2} \cos\left[\frac{\pi}{2} v\left(\frac{3}{2\pi} |\omega| - 1\right)\right], & 2\pi/3 < |\omega| \le 4\pi/3, \\ 0, & \text{otherwise}, \end{cases}$$
(3)

$$\hat{\psi}(\omega) = \begin{cases} e^{i\omega/2} (2\pi)^{-1/2} \sin\left[\frac{\pi}{2} v\left(\frac{3}{2\pi} |\omega| - 1\right)\right], & 2\pi/3 \le |\omega| \le 4\pi/3 \\ e^{i\omega/2} (2\pi)^{-1/2} \cos\left[\frac{\pi}{2} v\left(\frac{3}{4\pi} |\omega| - 1\right)\right], & 4\pi/3 < |\omega| \le 8\pi/3, \\ 0, & \text{otherwise} \end{cases}$$
(4)

where v(x) = 0 if $x \le 0$ and v(x) = 1 if $x \ge 1$. For $x \in [0,1]$, v(x) can be chosen in terms of the regularity (e.g., $v(x) = x^2(3 - 2x)$) with the restriction that v(x) + v(1 - x) = 1. The Meyer wavelet has compact support in the frequency domain and fast decay in the time domain.

Another family of wavelets that satisfy Assumption 2 is the spline wavelets of positive order $m \in \mathbb{Z}^+$. When *m* is odd, this family is of the form $\hat{\psi}(\omega) = e^{i\omega/2}b(\omega)$, where $b(\cdot)$ is a real-valued and symmetric function; when *m* is even, it is of the form $\hat{\psi}(\omega) = -ie^{i\omega/2}b(\omega)$, where $b(\cdot)$ is a real-valued and odd function (cf. Hernández and Weiss, 1996, (2.16), p. 161). One example of this family is the first order spline wavelet, often called the Franklin wavelet. It is given by

$$\hat{\phi}(\omega) = (2\pi)^{-1/2} \, \frac{\sin^2(\omega/2)}{(\omega/2)^2} \, (P_3(\omega/2))^{-1/2},\tag{5}$$

$$\hat{\psi}(\omega) = e^{i\omega/2} (2\pi)^{-1/2} \frac{\sin^4(\omega/4)}{(\omega/4)^2} \left(\frac{P_3(\omega/4 + \pi/4)}{P_3(\omega/2)P_3(\omega/4)}\right)^{1/2},\tag{6}$$

where $P_3(\omega) = \frac{2}{3} + \frac{1}{3}\cos(2\omega)$. Another example is the second order spline wavelet, given by

$$\hat{\phi}(\omega) = (2\pi)^{-1/2} \, \frac{\sin^3(\omega/2)}{(\omega/2)^3} \, (P_5(\omega/2))^{-1/2},\tag{7}$$

$$\hat{\psi}(\omega) = -ie^{i\omega/2} (2\pi)^{-1/2} \frac{\sin^6(\omega/4)}{(\omega/4)^3} \left(\frac{P_5(\omega/4 + \pi/4)}{P_5(\omega/2)P_5(\omega/4)}\right)^{1/2},\tag{8}$$

where $P_5(\omega) = \frac{1}{30} \cos^2(2\omega) + \frac{13}{30} \cos(2\omega) + \frac{8}{15}$. The Franklin wavelet and the second order spline wavelet are constructed from piecewise linear and quadratic functions, respectively. They have compact support in the time domain and have exponential decay in the frequency domain (cf. Hernández and Weiss, 1996, p. 149). In fact, the Harr wavelet (1) is the 0th order spline wavelet, but it does not satisfy Assumption 2 because its Fourier transform $\hat{\psi}(\omega) = -ie^{i\omega/2}(2\pi)^{-1/2}\sin^2(\omega/4)/(\omega/4)$ decays to 0 as $|\omega| \to \infty$ only at the rate of $|\omega|^{-1}$.

2.2. Wavelet Representation of the Spectral Density

The signal described previously can be a regression function or a probability density function in the time domain or a spectral density function in the frequency domain. We now consider a wavelet representation of the normalized spectral density function $f(\omega)$ of a covariance-stationary real-valued process $\{X_t, t \in \mathbb{Z}\}$. Because $f(\omega)$ is 2π -periodic, it is not square integrable over \mathbb{R} . We need to construct a wavelet basis $\{\Phi_{jk}(\cdot), \Psi_{jk}(\cdot)\}$ for the $L_2(\Pi)$ -space of 2π periodic functions, where $\Pi = [-\pi, \pi]$. Given an orthonormal wavelet basis $\{\phi_{jk}(\cdot), \psi_{jk}(\cdot)\}$ of $L_2(\mathbb{R})$, we can always construct an orthonormal wavelet basis $\{\Phi_{jk}(\cdot), \Psi_{jk}(\cdot)\}$ of $L_2(\Pi)$ by periodizing $\{\phi_{jk}(\cdot), \psi_{jk}(\cdot)\}$ via the formula

$$\Phi_{jk}(\omega) = (2\pi)^{-1/2} \sum_{m=-\infty}^{\infty} \phi_{jk} \left(\frac{\omega}{2\pi} + m\right),$$
(9)

$$\Psi_{jk}(\omega) = (2\pi)^{-1/2} \sum_{m=-\infty}^{\infty} \psi_{jk} \left(\frac{\omega}{2\pi} + m\right)$$
(10)

(cf. Daubechies, 1992, Ch. 9; Hernández and Weiss, 1996, Ch. 4). Both $\Phi_{jk}(\cdot)$ and $\Psi_{jk}(\cdot)$ are real valued.

Because (9) and (10) are an infinite sum, it is convenient to use compactly supported wavelets so that only a finite number of terms are nonzero. Alternatively, when $\psi(\cdot)$ has unbounded support, one can compute $\{\Phi_{jk}(\cdot), \Psi_{jk}(\cdot)\}$ from their Fourier transforms $\{\hat{\Phi}_{ik}(\cdot), \hat{\Psi}_{ik}(\cdot)\}$ via the formula

$$\Phi_{jk}(\omega) = (2\pi)^{-1/2} \sum_{h=-\infty}^{\infty} \hat{\Phi}_{jk}(h) e^{i\omega h},$$
(11)

$$\Psi_{jk}(\omega) = (2\pi)^{-1/2} \sum_{h=-\infty}^{\infty} \hat{\Psi}_{jk}(h) e^{i\omega h},$$
(12)

where

$$\hat{\Phi}_{jk}(h) = (2\pi)^{-1/2} \int_{-\pi}^{\pi} \Phi_{jk}(\omega) e^{-i\omega h} d\omega,$$
(13)

$$\hat{\Psi}_{jk}(h) = (2\pi)^{-1/2} \int_{-\pi}^{\pi} \Psi_{jk}(\omega) e^{-i\omega h} d\omega.$$
(14)

By the periodization techniques (9) and (10), and change of variables, we have

$$\hat{\Phi}_{jk}(h) = (2\pi)^{1/2} \hat{\phi}_{jk}(2\pi h) = e^{-i2\pi hk/2^{j}} (2\pi/2^{j})^{1/2} \hat{\phi}(2\pi h/2^{j}),$$
(15)

$$\hat{\Psi}_{jk}(h) = (2\pi)^{1/2} \hat{\psi}_{jk}(2\pi h) = e^{-i2\pi hk/2^{j}} (2\pi/2^{j})^{1/2} \hat{\psi}(2\pi h/2^{j}).$$
(16)

Note that the orthonormality of the periodized wavelet basis $\{\Phi_{jk}(\cdot), \Psi_{jk}(\cdot)\}$ implies

$$1 = \int_{-\pi}^{\pi} \Psi_{jk}^{2}(\omega) d\omega = \sum_{h=-\infty}^{\infty} |\hat{\Psi}_{jk}(h)|^{2} = (2\pi/2^{j}) \sum_{h=-\infty}^{\infty} |\hat{\psi}(2\pi h/2^{j})|^{2},$$
(17)

where the second equality follows from Parseval's identity and the last one from (16). By utilizing the properties of wavelets and their Fourier transforms, one can see how dilation and translation in periodized wavelets play their roles. Dilation parameter j varies dyadically, and translation parameter k varies as the modulation.

Recall that $\{X_t, t \in \mathbb{Z}\}$ is a covariance-stationary real-valued time series with normalized spectral density $f(\omega), \omega \in [-\pi, \pi]$. Fourier series are most often used in practice to represent $f(\omega)$, where the Fourier coefficients are autocorrelations at various lags; that is,

$$f(\omega) = (2\pi)^{-1} \sum_{h=-\infty}^{\infty} \rho(h) e^{-ih\omega}, \quad \omega \in [-\pi, \pi],$$
(18)

where $\rho(h) = R(h)/R(0)$, $R(h) = \text{Cov}(X_t, X_{t-|h|})$. The wavelet basis is, however, more effective in capturing nonsmooth features of $f(\omega)$. Let V_0 be the initial subspace in $L_2(\Pi)$. Then, we can partition $L_2(\Pi) = V_0 \bigoplus \sum_{j=0}^{\infty} W_j$. Without loss of generality, we restrict $k \in [1, 2^j] \cap \mathbb{Z}$, because of the use of the periodized wavelet basis $\{\Phi_{jk}(\cdot), \Psi_{jk}(\cdot)\}$. Thus, with the orthonormal wavelet basis $\{\Phi_{jk}(\cdot), \Psi_{jk}(\cdot)\}$ in $L_2(\Pi)$, $f(\omega)$ can be expressed as

$$f(\omega) = \beta_{00} \Phi_{00}(\omega) + \sum_{j=0}^{\infty} \sum_{k=1}^{2^{j}} \alpha_{jk} \Psi_{jk}(\omega), \qquad \omega \in [-\pi, \pi],$$
(19)

where the wavelet coefficients

$$\boldsymbol{\beta}_{00} = \int_{-\pi}^{\pi} f(\boldsymbol{\omega}) \Phi_{00}(\boldsymbol{\omega}) d\boldsymbol{\omega},$$
(20)

$$\alpha_{jk} = \int_{-\pi}^{\pi} f(\omega) \Psi_{jk}(\omega) d\omega.$$
(21)

The coefficients β_{00} and $\{\alpha_{jk}\}$ are the orthogonal projections of $f(\omega)$ on wavelet bases. They are real valued. Without loss of generality, we can choose a scale function $\phi(\cdot)$ such that $\hat{\phi}(\omega) = 0$ for $|\omega| > \pi$ or $|\hat{\phi}(\omega)|$ is continuous. It follows that $(2\pi)^{1/2}\hat{\phi}(2k\pi) = 0$ for $k \in \mathbb{Z}, k \neq 0$, and $(2\pi)^{1/2}\hat{\phi}(0) = 1$ (cf. Hernández and Weiss, 1996, Proposition 2.17, p. 64). Thus, $\Phi_{00}(\omega) = (2\pi)^{-1/2}$ for $\omega \in [-\pi, \pi]$ and $\beta_{00} = (2\pi)^{-1/2}$. Consequently, we can write (19) as

$$f(\omega) = (2\pi)^{-1} + \sum_{j=0}^{\infty} \sum_{k=1}^{2^j} \alpha_{jk} \Psi_{jk}(\omega), \quad \omega \in [-\pi, \pi].$$
(22)

By Parseval's identity and (16), we can also express α_{jk} in the time domain, namely,

$$\alpha_{jk} = (2\pi)^{-1/2} \sum_{h=-\infty}^{\infty} \rho(h) \hat{\Psi}_{jk}(h)$$

= $\sum_{h=-\infty}^{\infty} \rho(h) \hat{\psi}_{jk}(2\pi h) = \sum_{h=-\infty}^{\infty} \rho(h) \hat{\psi}_{jk}^*(2\pi h),$ (23)

where $\{\hat{\psi}_{jk}(\cdot)\}\$ is given in (16) and the last equality follows given $\hat{\psi}^*(z) = \hat{\psi}(-z)$. Unlike the Fourier coefficients, $\{\alpha_{jk}\}\$ do not represent autocorrelations at different lags. They are the weighted average of autocorrelations centered at varying locations. It is intuitively clear from the expression of $\{\alpha_{jk}\}\$ why wavelets can capture the peaks of $f(\omega)$. Suppose, for example, that $f(\omega)$ has a peak at some frequency, say, $\omega = 0$, which can arise when $\{\rho(h)\}\$ have the same, positive sign and decay to zero slowly. Such a pattern can be effectively captured by $\{\alpha_{ik}\}\$ with sufficiently large j's.

2.3. Wavelet Spectral Density Estimator and Test Statistic

Suppose that we observe a sample $\{X_t\}_{t=1}^n$. We define the sample autocorrelation function, $\hat{\rho}(h) = \hat{R}(h)/\hat{R}(0)$, where $\hat{R}(h) = n^{-1} \sum_{t=|h|+1}^n (X_t - \overline{X}_n) \times (X_{t-|h|} - \overline{X}_n), \overline{X}_n = n^{-1} \sum_{t=1}^n X_t$. A natural choice of the estimator for α_{jk} is the empirical wavelet coefficient

$$\hat{\alpha}_{jk} = \sum_{h=1-n}^{n-1} \hat{\rho}(h) \hat{\psi}_{jk}(2\pi h) = \sum_{h=1-n}^{n-1} \hat{\rho}(h) \hat{\psi}_{jk}^*(2\pi h).$$
(24)

Then a wavelet estimator of the spectral density $f(\omega)$ can be given by

$$\hat{f}(\omega) = (2\pi)^{-1} + \sum_{j=0}^{J} \sum_{k=1}^{2^{j}} \hat{\alpha}_{jk} \Psi_{jk}(\omega), \qquad \omega \in [-\pi, \pi],$$
(25)

where $J \equiv J_n$ is the finest scale corresponding to the highest resolution level used in the approximation. The degree of approximation (or bias) depends on J. The larger J is, the smaller the bias. On the other hand, J also affects the sampling variation (i.e., variance) of $\hat{f}(\omega)$. The larger J is, the larger the variance of $\hat{f}(\omega)$. Given each sample size n, a suitable J should be chosen to balance the variance and the squared bias so that $\hat{f}(\omega)$ will be consistent for $f(\omega)$. There are a total of $\sum_{j=0}^{J} 2^j = 2^{J+1} - 1$ empirical wavelet coefficients in (25). The finest scale J should be smaller than $\log_2 n$. Proper conditions on J will be given to ensure that the proposed test statistic has a well-defined limit distribution. In our simulation, we will choose J via an automatic data-driven method by Walter (1994), which, to a certain extent, lets the data themselves determine a proper J given each n. Now we construct a consistent test for serial correlation of unknown form, with expected good power against alternatives with nonsmooth spectrum. The hypotheses of interest are

$$H_0: \rho(h) = 0$$
 for all $h \in \mathbb{Z}, h \neq 0$

versus

 $H_A: \rho(h) \neq 0$ for some $h \in \mathbb{Z}, h \neq 0$.

Note that H_A includes all possible serially autocorrelated alternatives. Under the null hypothesis H_0 , the spectral density $f(\omega)$ becomes

$$f_0(\omega) = (2\pi)^{-1}$$
 for all $\omega \in [-\pi, \pi]$. (26)

Hence, all the wavelet coefficients $\{\alpha_{jk}\}\$ are zero under H_0 . Under H_A , however, $f(\omega)$ is not a constant function of ω . At least one wavelet coefficient is nonzero. Thus, testing for serial uncorrelatedness is the same as testing whether all the wavelet coefficients $\{\alpha_{jk}\}\$ are jointly zero.

We now propose a test for H_0 versus H_A using a quadratic form, defined as

$$Q(\hat{f};f_0) = \int_{-\pi}^{\pi} (\hat{f}(\omega) - f_0(\omega))^2 d\omega = \sum_{j=0}^{J} \sum_{k=1}^{2^j} \hat{\alpha}_{jk}^2,$$
(27)

where the second equality follows from the orthonormality that

$$\int_{-\pi}^{\pi} \Psi_{jk}(\omega) \Psi_{lm}(\omega) d\omega = \delta_{jl} \delta_{km}, \qquad j, k, l, m \in \mathbb{Z}.$$
(28)

Divergence measures other than the quadratic form could be used also, but (27) is convenient because it involves no numerical integration over frequency ω . Our test statistic is constructed by properly standardizing (27); that is,

$$W_n = \left\{ 2\pi n \sum_{j=0}^{J} \sum_{k=1}^{2^j} \hat{\alpha}_{jk}^2 - (2^{J+1} - 1) \right\} / \{4(2^{J+1} - 1)\}^{1/2},$$
(29)

where $2^{J+1} - 1$ and $4(2^{J+1} - 1)$ are approximately the mean and variance of $2\pi nQ(\hat{f};f_0)$. We could use 2^{J+1} to replace $2^{J+1} - 1$, but the latter is expected to give better finite sample performances when J is small.

Hong (1996) considers a class of consistent tests for serial correlation of unknown form using the Parzen (1957) kernel estimator for the spectral density $f(\omega)$, which depends on Fourier analysis. Unlike the Fourier transform, (22) represents $f(\omega)$ via the wavelet transform. For the kernel-based test, kernels typically put more weight on low order autocorrelations and less weight on high order autocorrelations. In contrast, wavelets do not necessarily weigh down high order autocorrelations. Instead, they impose different weights via different scales and translations. In this sense, the wavelet-based test is expected to have better power than the kernel-based test when $f(\omega)$ has distinctive local features such as peaks and spikes. If $f(\omega)$ is smooth without singularities, however, the kernel-based test is expected to perform well. Our simulation study, which follows, confirms these claims.

3. ASYMPTOTIC NULL DISTRIBUTION

To derive the null limit distribution of W_n , we impose the following condition.

Assumption 3. $\{X_t\}_{t=-\infty}^{\infty}$ is independent and identically distributed with $E(X_t - \mu)^2 = \sigma^2$ and $E(X_t - \mu)^4 = \mu_4 < \infty$, where $\mu = EX_t$. A random sample $\{X_t\}_{t=1}^n$ of size $n \in \mathbb{Z}^+$ is observed.

Assumption 3 allows non-Gaussian processes as are common for economic and financial time series. We now establish the asymptotic normality of the wavelet-based test.

THEOREM 1. Suppose that Assumptions 1–3 hold and $J \equiv J_n \rightarrow \infty$, $2^{2J}/n \rightarrow 0$. Then $W_n \rightarrow^d N(0,1)$.

The conditions on the finest scale J ensure the asymptotic normality of W_n . They are analogous to the conditions on the bandwidth (or the lag order) for the kernel test of Hong (1996). The finest scale J is restricted to increase at a slower rate than $\frac{1}{2} \log_2 n$. In our simulation, we use an automatic data-driven method proposed by Walter (1994) to choose J. Although both the finest scale J in wavelet estimation and the bandwidth in kernel estimation are smoothing parameters, they are conceptually different. In particular, J is not a lag order; it is the integer corresponding to the finest resolution in wavelet decomposition. At each level $j \leq J$, all the sample autocorrelations $\{\hat{\rho}(h)\}_{h=1}^{n-1}$ are used to obtain the empirical wavelet coefficients $\{\hat{\alpha}_{jk}\}$ when $\hat{\psi}(\cdot)$ has unbounded support (see (24)).

4. CONSISTENCY

To establish consistency of the proposed test under the alternative hypothesis H_A , we impose a condition on the temporal dependence of $\{X_t\}$.

Assumption 4. $\{X_t\}_{t=-\infty}^{\infty}$ is fourth order stationary with $\sum_{h=-\infty}^{\infty} R^2(h) < \infty$ and $\sum_{j=-\infty}^{\infty} \sum_{k=-\infty}^{\infty} \sum_{l=-\infty}^{\infty} |\kappa(j,k,l)| < \infty$, where $\kappa(j,k,l)$ is the fourth order cumulant of the joint distribution of $\{X_t, X_{t+j}, X_{t+k}, X_{t+l}\}$, where $j, k, l \in \mathbb{Z}$.

The fourth order cumulant $\kappa(j, k, l)$ is defined as

$$\kappa(j,k,l) = E(X_t X_{t+j} X_{t+k} X_{t+l}) - E(\widetilde{X}_t \widetilde{X}_{t+j} \widetilde{X}_{t+k} \widetilde{X}_{t+l}),$$
(30)

where $\{\tilde{X}_t\}$ is a Gaussian sequence with the same mean and covariance function as $\{X_t\}$. Cumulant conditions are widely used in time series analysis (e.g.,

Anderson, 1971; Andrews, 1991; Hannan, 1970). The cumulant condition in Assumption 4 holds trivially for Gaussian processes. It is also satisfied if $\{X_t\}$ is a fourth order stationary linear process with absolutely summable coefficients and finite fourth moment (cf. Hannan, 1970). Andrews (1991, Lemma 1) provides a primitive mixing condition to ensure the cumulant condition. Neumann (1996) also uses a higher order cumulant condition.

THEOREM 2. Suppose that Assumptions 1, 2, and 4 hold and $J \equiv J_n \to \infty$, $2^{3J/2}/n \to 0$. Let $Q(f;f_0)$ be defined as $Q(\hat{f};f_0)$ in (27) with $\hat{f}(\cdot)$ replaced by $f(\cdot)$. Then

$$\frac{2(2^{J+1}-1)^{1/2}}{n} W_n \to^p 2\pi Q(f;f_0).$$

Theorem 2 implies that W_n is consistent against H_A , because $Q(f;f_0) > 0$ if and only if H_A holds. In particular, it is consistent against fractionally integrated processes, I(d), for $d < \frac{1}{4}$. The conditions on the finest scale J are weaker than those under Theorem 1.

Following reasoning analogous to that of Hong (1996), it can be shown that Bahadur's (1960) asymptotic slope of the wavelet test W_n under H_A does not depend on the choice of the mother wavelet $\psi(\cdot)$. In other words, the asymptotic power of W_n does not depend on the choice of $\psi(\cdot)$. This is in contrast to the kernel-based test of Hong (1996, Sect. 5), for which the asymptotic power depends on the choice of a kernel function, with the Daniell kernel being optimal within a class of kernel functions.

5. FINITE SAMPLE PERFORMANCE

We now study the finite sample performances of the wavelet-based test W_n in comparison with the kernel-based test of Hong (1996). To study the impact of the choice of mother wavelet $\psi(\cdot)$ on the size and power of W_n in finite samples, we use three wavelets—Meyer, Franklin, and the second order spline wavelets. For the Meyer wavelet, we choose v(x) = x for $x \in (0,1)$. We also consider more regular forms such as $v(x) = x^2(3 - 2x)$ for $x \in (0,1)$, but simulations show that the choice of v(x) has little impact on the size and power of W_n .

Both the wavelet- and the kernel-based tests involve the choice of smoothing parameters—the finest scale and the lag order, which are not directly comparable because they are conceptually different. It is thus critical to choose the smoothing parameters by appropriate data-driven methods, which determine the smoothing parameters based on data information. For W_n , we employ the algorithm of Walter (1994) to choose J, where the change in the integrated mean squared error (IMSE) from one scale to the next finer scale is proportional to the sum of squared empirical wavelet coefficients. The IMSE at the scale J is

given by $e_J = E \int_{-\pi}^{\pi} [\hat{f}_J(\omega) - f(\omega)]^2 d\omega$, where $\hat{f}_J(\omega)$ is a wavelet spectral density estimator (25) using the finest scale *J*. The change in IMSE from J - 1 to *J* is proportional to $\sum_{k=1}^{2^J} \hat{\alpha}_{Jk}^2$, where $\hat{\alpha}_{Jk}$ is the empirical wavelet coefficient at the scale *J*. One starts from the initial scale J = 0 and checks how much the error changes from zero to one. The grid search is iterated until we get the scale *J* at which the error increases most rapidly. Then, one obtains the finest scale. Here, we choose the finest scale *J* for which the change in error between *J* and J + 1 exceeds 100%.

The kernel-based test of Hong (1996) is constructed by comparing a Parzen (1957) kernel-based spectral density estimator and the null spectral density $f_0(\omega)$. A Parzen (1957) kernel-based spectral density estimator is given by

$$\hat{f}_{p}(\omega) = (2\pi)^{-1} \sum_{h=1-n}^{n-1} k(h/p)\hat{\rho}(h)e^{-ih\omega}, \qquad \omega \in [-\pi,\pi],$$

where $k(\cdot)$ is a kernel function and $p \equiv p_n$ is the bandwidth such that $p \to \infty$, $p/n \to 0$.

From a standardized version of a quadratic form, the test statistic is

$$K_n = \left\{ n \sum_{h=1}^{n-1} k^2(h/p) \hat{\rho}^2(h) - C_n(k) \right\} / \{2D_n(k)\}^{1/2},$$
(31)

where

$$C_n(k) = \sum_{h=1}^{n-1} (1 - h/n) k^2 (h/p),$$

$$D_n(k) = \sum_{h=1}^{n-2} (1 - h/n) (1 - (h+1)/n) k^4 (h/p).$$

For the choice of kernel $k(\cdot)$, we use the Daniell kernel, $k(z) = \sin(\pi z)/\pi z$, $z \in \mathbb{R}$, which maximizes the asymptotic power of K_n over a class of kernel functions (cf. Hong, 1996). We choose a data-driven bandwidth p using the cross-validation procedure of Beltrao and Bloomfield (1987). Here, the bandwidth is determined to maximize the cross-validated log likelihood in the frequency domain, which is asymptotically equivalent to minimizing a weighted IMSE. We use a grid search for the optimal integer-valued bandwidth p over the range from 2 to 15. The algorithm is implemented by fast Fourier transform. See Beltrao and Bloomfield (1987) for more discussion.

Two sample sizes, n = 64 and n = 128, are considered. First, we study the size under normal and nonnormal processes using a GAUSS pseudo random number generator on a personal computer. Nonnormal cases include lognormal and uniform processes, scaled to have mean zero and variance one. Table 1 reports the percentage rejections of the tests W_n and K_n at the 10% and 5%

			<i>n</i> = 64		<i>n</i> = 128			
		Normal	Uniform	Lognormal	Normal	Uniform	Lognormal	
$\overline{K_n}$	10%	0.115	0.138	0.097	0.124	0.138	0.098	
'n	5%	0.081	0.096	0.079	0.086	0.102	0.070	
W_{1n}	10%	0.123	0.135	0.112	0.152	0.127	0.111	
111	5%	0.091	0.096	0.089	0.096	0.088	0.072	
W_{2n}	10%	0.138	0.136	0.114	0.145	0.128	0.114	
2	5%	0.094	0.102	0.085	0.094	0.093	0.077	
<i>W</i> _{3<i>n</i>}	10%	0.174	0.169	0.138	0.214	0.167	0.156	
	5%	0.121	0.131	0.103	0.150	0.126	0.096	

TABLE 1. Size at the 10% and 5% levels^a

a1,000 iterations.

Abbreviations: K_n , kernel-based test (Daniell kernel); W_{1n} , wavelet-based test (Meyer wavelet); W_{2n} , wavelet-based test (Franklin wavelet); W_{3n} , wavelet-based test (Spline wavelet of order two).

significance levels, based on 1,000 iterations. Both K_n and W_n have reasonable sizes at the 10% level, but they have some overrejections at the 5% level. In most cases, W_n tends to overreject H_0 a little more than K_n . For all the three wavelets, W_n has more accurate sizes under the lognormal process than under both normal and uniform processes at the 10% level.

Next we study the size-corrected power using empirical critical values, which ensure fair power comparison for the tests under study. The empirical critical values are obtained from 1,000 iterations for normal and nonnormal processes, respectively. Alternatives are chosen according to the shapes of their spectral densities. They are

$$ARMA(4,4): X_{t} + 0.1X_{t-1} + 0.3X_{t-4} = \varepsilon_{t} + \varepsilon_{t-4},$$

$$AR(4): X_{t} = 0.3X_{t-4} + \varepsilon_{t},$$

$$ARIMA(0,0.2,0): (1 - B)^{0.2}X_{t} = \varepsilon_{t}, \qquad BX_{t} = X_{t-1},$$

$$AR(1): X_{t} = 0.3X_{t-1} + \varepsilon_{t},$$

where ε_t is i.i.d.(0,1). We consider normal, lognormal, and uniform distributions for ε_t , respectively. The first two alternatives generate spatially inhomogeneous features such as peaks or spikes in the spectral densities. Both ARMA(4,4) and AR(4) can arise from quarterly data. ARIMA(0,0.2,0) is a fractionally integrated process, a well-known long memory time series process. The spectral density of the long memory process is $f(\omega) = (1/2\pi)|1 - e^{-i\omega}|^{-2d}$, for 0 < d < 0.5. It is infinite at $\omega = 0$ but is quite smooth with no jumps elsewhere. Figure 1 describes the autocorrelation function $\rho(h)$ and spectral density function $f(\omega)$ for each of the four alternatives. The autocorrelations are generated up to lag 20 using



FIGURE 1. Sample autocorrelation (1) and true spectral density (2): ARMA(4,4) and AR(4).

a realization of the random sample $\{X_t\}_{t=1}^{10,000}$. The spectral densities are depicted for $\omega \in [0, \pi]$ (the spectral density of the long memory process is depicted for $\omega \in (0, \pi]$). There exist distinctive local features (kinks or spikes) for ARMA(4,4) and AR(4). On the other hand, the spectral density of AR(1) has no spike at any frequency.



FIGURE 1 (CONTINUED). Sample autocorrelation (1) and true spectral density (2): ARFIMA(0, d, 0) and AR(1).

Table 2 reports the power at the 10% and 5% levels when the innovations are normally distributed. For ARMA(4,4) and AR(4) processes, W_n has better powers than K_n regardless of the choice of the wavelet $\psi(\cdot)$ and the sample size n. Among the wavelets, the Meyer and Franklin wavelets have slightly better powers than the second order spline wavelet for the smaller sample size (n = 64).

	K_n		W _{1n}		W_{2n}		<i>W</i> _{3<i>n</i>}		
DGP	10%	5%	10%	5%	10%	5%	10%	5%	
	n = 64								
ARMA(4,4)	0.507	0.392	0.719	0.534	0.707	0.533	0.674	0.516	
AR(4)	0.310	0.219	0.476	0.340	0.482	0.347	0.447	0.332	
ARIMA(0,0.2,0)	0.410	0.298	0.352	0.246	0.341	0.238	0.331	0.235	
AR(1)	0.643	0.523	0.463	0.320	0.468	0.333	0.460	0.352	
	n = 128								
ARMA(4,4)	0.914	0.842	0.981	0.961	0.980	0.955	0.964	0.898	
AR(4)	0.599	0.515	0.774	0.684	0.760	0.673	0.710	0.597	
ARIMA(0,0.2,0)	0.710	0.650	0.618	0.521	0.601	0.522	0.544	0.435	
AR (1)	0.912	0.869	0.731	0.652	0.723	0.636	0.683	0.585	

TABLE 2. Size-corrected power under normal innovations^a

^{*a*}1,000 iterations. Power is computed using empirical critical values obtained from 1,000 iterations under an i.i.d. N(0,1) process.

Abbreviations: K_n , kernel-based test (Daniell kernel); W_{1n} , wavelet-based test (Meyer wavelet); W_{2n} , wavelet-based test (Franklin wavelet); W_{3n} , wavelet-based test (Spline wavelet of order two).

For the larger sample size (n = 128), all the three wavelets deliver similar powers, which is consistent with the asymptotic theory. For the long memory process, W_n performs similarly to K_n . Although there is a peak at zero frequency, the spectral density of the long memory process is rather smooth elsewhere, and the kernel method performs well for this alternative. On the other hand, K_n outperforms W_n in detecting AR(1), which has a spatially homogeneous spectral density with no peaks.

Tables 3 and 4 report the power when the innovations are lognormally and uniformly distributed, respectively. Again, W_n is superior to K_n against ARMA(4,4) and AR(4). Each wavelet performs quite similarly and dominates K_n , particularly against AR(4). For the long memory process, W_n has better power than K_n for n = 128 when innovations are lognormally distributed, but when innovations are uniformly distributed, K_n outperforms W_n . Also, K_n obviously outperforms W_n against AR(1) under both lognormal and uniform innovations. This evidence, though limited, confirms the theoretical claims that the wavelet method can more effectively detect spatial inhomogeneity of the spectrum than the kernel method. The power of the tests clearly depends on how spatially inhomogeneous the spectral density is.

In summary, we observe that (1) the choice of wavelets, in general, has little effect on the power of the wavelet-based test W_n , as expected from the asymptotic theory and (2) the wavelet-based test W_n has better power than the kernel-based test K_n when the spectral density exhibits distinct local features. When the spectral density is smooth, however, the kernel-based test K_n performs better.

	K _n		W_{1n}		W_{2n}		W _{3n}			
DGP	10%	5%	10%	5%	10%	5%	10%	5%		
	n = 64									
ARMA(4,4)	0.559	0.322	0.697	0.453	0.697	0.440	0.657	0.371		
AR(4)	0.285	0.147	0.427	0.252	0.434	0.245	0.392	0.194		
ARIMA(0,0.2,0)	0.676	0.538	0.689	0.552	0.688	0.531	0.707	0.557		
AR(1)	0.658	0.440	0.371	0.222	0.381	0.215	0.385	0.204		
	n = 128									
ARMA(4,4)	0.938	0.842	0.982	0.937	0.980	0.942	0.972	0.920		
AR(4)	0.630	0.467	0.808	0.672	0.787	0.688	0.773	0.640		
ARIMA(0,0.2,0)	0.834	0.782	0.933	0.875	0.914	0.861	0.941	0.891		
AR(1)	0.953	0.890	0.726	0.599	0.713	0.612	0.722	0.600		

TABLE 3. Size-corrected power under lognormal innovations^a

^{*a*}1,000 iterations. Power is computed using empirical critical values obtained from 1,000 iterations under an i.i.d. lognormal process.

Abbreviations: K_n , kernel-based test (Daniell kernel); W_{1n} , wavelet-based test (Meyer wavelet); W_{2n} , wavelet-based test (Franklin wavelet); W_{3n} , wavelet-based test (Spline wavelet of order two).

6. CONCLUSION

We propose a wavelet-based consistent test for serial correlation of unknown form. The test is constructed by comparing a wavelet-based spectral density

	K_n		W_{1n}		W_{2n}		W_{3n}			
DGP	10%	5%	10%	5%	10%	5%	10%	5%		
	n = 64									
ARMA(4,4)	0.500	0.367	0.699	0.507	0.680	0.500	0.658	0.480		
AR(4)	0.280	0.212	0.486	0.330	0.479	0.334	0.449	0.290		
ARIMA(0,0.2,0)	0.373	0.288	0.287	0.191	0.277	0.185	0.263	0.177		
AR(1)	0.629	0.520	0.449	0.301	0.435	0.289	0.435	0.291		
	n = 128									
ARMA(4,4)	0.875	0.746	0.979	0.943	0.979	0.950	0.952	0.942		
AR(4)	0.570	0.441	0.785	0.692	0.769	0.681	0.746	0.716		
ARIMA(0,0.2,0)	0.704	0.601	0.590	0.503	0.573	0.504	0.535	0.518		
AR (1)	0.903	0.837	0.755	0.646	0.748	0.668	0.725	0.706		

TABLE 4. Size-corrected power under uniform innovations^a

^{*a*}1,000 iterations. Power is computed using empirical critical values obtained from 1,000 iterations under an i.i.d. uniform process.

Abbreviations: K_n , kernel-based test (Daniell kernel); W_{1n} , wavelet-based test (Meyer wavelet); W_{2n} , wavelet-based test (Franklin wavelet); W_{3n} , wavelet-based test (Spline wavelet of order two).

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estimator and the null spectral density. The null asymptotic distribution of the proposed test is one-sided standard normal. Our simulation study shows that the wavelet-based test is more powerful than the kernel-based test of Hong (1996) when the data generating process has distinctive local spectral features, which confirms the theoretical claims that wavelets can effectively detect spatial inhomogeneous features. On the other hand, when the spectral density is smooth with no peaks or spikes, the kernel-based test outperforms the wavelet-based test. Because the spectral densities of time series arising in practice usually have unknown smoothness, the wavelet-based test is a useful complement to the kernel-based test in practice.

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APPENDIX A

To prove Theorems 1 and 2, we first state a useful lemma, which is proved in Appendix B.

LEMMA A.1. Suppose that Assumptions 1 and 2 hold and $J \rightarrow \infty, 2^J/n \rightarrow 0$. Define

$$b_J(h,m) = a_J(h,m) + a_J(-h,-m) + a_J(h,-m) + a_J(-h,m),$$

where $a_J(h,m) = 2\pi \sum_{j=0}^{J} \sum_{k=1}^{2^j} \hat{\psi}_{jk}(2\pi h) \hat{\psi}_{jk}^*(2\pi m)$. Then

- (i) $b_J(h,m)$ is real valued, $b_J(0,m) = b_J(h,0) = 0$, and $b_J(h,m) = b_J(m,h)$;
- (ii) $\sum_{h=1}^{n-1} \sum_{m=1}^{n-1} h^{\nu} |b_J(h,m)| = O(2^{(1+\nu)J})$ for $0 \le \nu \le 1/2$;

- (iii) $\sum_{h=1}^{n-1} \{\sum_{m=1}^{n-1} |b_J(h,m)|\}^2 = O(2^J);$ (iv) $\sum_{h=1}^{n-1} \sum_{h_2=1}^{n-1} \{\sum_{m=1}^{m-1} |b_J(h_1,m)b_J(h_2,m)|\}^2 = O\{(J+1)2^J\};$ (v) $\sum_{h=1}^{n-1} b_J(h,h) = (2^{J+1} 1)\{1 + O[(J+1)/2^J + (2^J/n)^{2\alpha-1}]\},$ where α is in Assumption 2;
- (vi) $\sum_{h=1}^{n-1} \sum_{m=1}^{n-1} b_J^2(h,m) = 2(2^{J+1}-1)\{1+o(1)\}.$

Proof of Theorem 1. For simplicity and without loss of generality, we assume $E(X_t) = 0$ and consider $\hat{R}(h) = n^{-1} \sum_{t=|h|+1}^n X_t X_{t-|h|}$. It can be shown that subtracting the sample mean $\overline{X}_n = n^{-1} \sum_{t=1}^n X_t$ from $\{X_t\}$ has no impact on the limit distribution of the test statistic W_n . Using (24), we have

$$2\pi n \sum_{j=0}^{J} \sum_{k=1}^{2^{j}} \hat{\alpha}_{jk}^{2} = 2\pi n \sum_{j=0}^{J} \sum_{k=1}^{2^{j}} \left(\sum_{h=1-n}^{n-1} \hat{\rho}(h) \hat{\psi}_{jk}(2\pi h) \right) \left(\sum_{m=1-n}^{n-1} \hat{\rho}(m) \hat{\psi}_{jk}^{*}(2\pi m) \right)$$
$$= n \sum_{h=1-n}^{n-1} \sum_{m=1-n}^{n-1} a_{j}(h,m) \hat{\rho}(h) \hat{\rho}(m)$$
$$= n \sum_{h=1}^{n-1} \sum_{m=1}^{n-1} b_{j}(h,m) \hat{\rho}(h) \hat{\rho}(m), \qquad (A.1)$$

where the last equality follows from reindexing and the definition of $b_J(h, m)$.

Using $\hat{\rho}(h) = \hat{R}(h)/\hat{R}(0)$ and $\hat{R}(0) - \sigma^2 = O_P(n^{-1/2})$ by Assumption 3, we have

$$n\sum_{h=1}^{n-1}\sum_{m=1}^{n-1}b_{J}(h,m)\hat{\rho}(h)\hat{\rho}(m) = \sigma^{-4}n\sum_{h=1}^{n-1}\sum_{m=1}^{n-1}b_{J}(h,m)\hat{R}(h)\hat{R}(m) + \{\hat{R}^{-2}(0) - \sigma^{-4}\}n\sum_{h=1}^{n-1}\sum_{m=1}^{n-1}b_{J}(h,m)\hat{R}(h)\hat{R}(m) = \sigma^{-4}n\sum_{h=1}^{n-1}\sum_{m=1}^{n-1}b_{J}(h,m)\hat{R}(h)\hat{R}(m) + O_{P}(2^{J}/n^{1/2}), \quad (A.2)$$

where the second term is of the indicated order of magnitude because

$$E\sum_{h=1}^{n-1}\sum_{m=1}^{n-1} |b_J(h,m)\hat{R}(h)\hat{R}(m)| \le Cn^{-1}\sum_{h=1}^{n-1}\sum_{m=1}^{n-1} |b_J(h,m)| = O(2^J/n)$$

given $E\hat{R}^2(h) \leq Cn^{-1}$ and Lemma A.1(ii).

We now focus on the first term in (A.2). Given $\hat{R}(h) = n^{-1} \sum_{t=h+1}^{n} X_t X_{t-h}$,

$$n\sum_{h=1}^{n-1}\sum_{m=1}^{n-1}b_{J}(h,m)\hat{R}(h)\hat{R}(m) = n^{-1}\sum_{h=1}^{n-1}\sum_{m=1}^{n-1}b_{J}(h,m)\sum_{t=h+1}^{n}\sum_{s=m+1}^{n}X_{t}X_{t-h}X_{s}X_{s-m}$$
$$= n^{-1}\sum_{h=1}^{n-1}\sum_{m=1}^{n-1}b_{J}(h,m)$$
$$\times \left(\sum_{t=1}^{n}\sum_{s=1}^{n}-\sum_{t=1}^{h}\sum_{s=m+1}^{n}-\sum_{t=1}^{n}\sum_{s=1}^{m}\right)X_{t}X_{t-h}X_{s}X_{s-m}$$
$$= \hat{A}_{n} + \hat{B}_{1n} - \hat{B}_{2n} - \hat{B}_{3n}, \qquad (A.3)$$

where

$$\hat{A}_{n} = n^{-1} \sum_{h=1}^{n-1} \sum_{m=1}^{n-1} b_{J}(h,m) \left(\sum_{t=2}^{n} \sum_{s=1}^{t-1} + \sum_{s=2}^{n} \sum_{t=1}^{s-1} \right) X_{t} X_{t-h} X_{s} X_{s-m}$$

$$= 2n^{-1} \sum_{h=1}^{n-1} \sum_{m=1}^{n-1} b_{J}(h,m) \sum_{t=2}^{n} \sum_{s=1}^{t-1} X_{t} X_{t-h} X_{s} X_{s-m} \quad \text{given } b_{J}(h,m) = b_{J}(m,h),$$

$$\hat{B}_{1n} = n^{-1} \sum_{h=1}^{n-1} \sum_{m=1}^{n-1} b_{J}(h,m) \sum_{t=1}^{n} X_{t}^{2} X_{t-h} X_{t-m},$$

$$\hat{B}_{2n} = n^{-1} \sum_{h=1}^{n-1} \sum_{m=1}^{n-1} b_{J}(h,m) \sum_{t=1}^{n} \sum_{s=m+1}^{n} X_{t} X_{t-h} X_{s} X_{s-m},$$

$$\hat{B}_{3n} = n^{-1} \sum_{h=1}^{n-1} \sum_{m=1}^{n-1} b_{J}(h,m) \sum_{t=1}^{n} \sum_{s=1}^{m} X_{t} X_{t-h} X_{s} X_{s-m}.$$

Proposition 1 shows that the U-statistic \hat{A}_n is dominant.

PROPOSITION 1. Suppose that Assumptions 1–3 hold and $J \to \infty, 2^{2J}/n \to 0$. Then $2^{-J/2} \{2\pi n \sum_{j=0}^{J} \sum_{k=1}^{2^{j}} \hat{\alpha}_{jk}^{2} - (2^{J+1} - 1)\} = 2^{-J/2} \sigma^{-4} \hat{A}_{n} + o_{P}(1).$

We now decompose \hat{A}_n into the terms with t - s > q and $t - s \le q$, for some $q \in \mathbb{Z}^+$:

$$\hat{A}_{n} = 2n^{-1} \sum_{h=1}^{n-1} \sum_{m=1}^{n-1} b_{J}(h,m) \left(\sum_{t=q+2}^{n} \sum_{s=1}^{t-q-1} + \sum_{t=2}^{n} \sum_{s=\max(t-q,1)}^{t-1} \right) X_{t} X_{t-h} X_{s} X_{s-m}$$

= $\hat{B}_{n} + \hat{B}_{4n}$, say. (A.4)

Furthermore, we decompose

$$\hat{B}_{n} = 2n^{-1} \left(\sum_{h=1}^{q} \sum_{m=1}^{q} + \sum_{h=1}^{q} \sum_{m=q+1}^{n-1} + \sum_{h=q+1}^{n-1} \sum_{m=1}^{n-1} \right) b_{J}(h,m) \sum_{t=q+2}^{n} \sum_{s=1}^{t-q-1} X_{t} X_{t-h} X_{s} X_{s-m}$$
$$= \hat{U}_{n} + \hat{B}_{5n} + \hat{B}_{6n}, \quad \text{say,}$$
(A.5)

where B_{5n} and \hat{B}_{6n} are the contributions from m > q and h > q, respectively.

Proposition 2 shows that \hat{A}_n can be approximated arbitrarily well by \hat{U}_n under a proper condition on q.

PROPOSITION 2. Suppose that Assumptions 1–3 hold, $J \to \infty, 2^{2J}/n \to 0$, and $q \equiv q_n \to \infty, q/2^J \to \infty, q^2/n \to 0$. Then $2^{-J/2}\hat{A}_n = 2^{-J/2}\hat{U}_n + o_P(1)$.

It is much easier to show the asymptotic normality of \hat{U}_n than of \hat{A}_n , because for \hat{U}_n , $\{X_t X_{t-h}\}$ and $\{X_s X_{s-m}\}$ are independent given t - s > q and $0 < h, m \le q$.

PROPOSITION 3. Suppose that Assumptions 1–3 hold and $J \to \infty, 2^{2J}/n \to 0$, $q/2^J \to \infty, q^2/n \to 0$. Let $\lambda_n^2 = E\hat{U}_n^2$. Then $4(2^{J+1} - 1)\sigma^8/\lambda_n^2 \to 1$, and $\lambda_n^{-1}\hat{U}_n \to d^{-1}N(0,1)$.

Propositions 1–3 and the Slutsky theorem imply $W_n \rightarrow^d N(0,1)$. The proof of Theorem 1 will be completed provided Propositions 1–3 are shown.

Proof of Proposition 1. By (A.1)-(A.3), we obtain

$$2\pi n \sum_{j=0}^{J} \sum_{k=1}^{2^{j}} \hat{\alpha}_{jk}^{2} - (2^{j+1} - 1) = \sigma^{-4} \{ \hat{A}_{n} + (\hat{B}_{1n} - \sigma^{4}(2^{j+1} - 1)) - \hat{B}_{2n} - \hat{B}_{3n} \} + O_{P}(2^{j}/n^{1/2}).$$

We shall show that (i) $2^{-J/2}\{\hat{B}_{1n} - \sigma^4(2^{J+1} - 1)\} \rightarrow^p 0$; (ii) $2^{-J/2}\hat{B}_{2n} \rightarrow^p 0$; (iii) $2^{-J/2}\hat{B}_{3n} \rightarrow^p 0$.

(i) Given Assumption 3 and $E(X_t^2 X_{t-h} X_{t-m}) = \sigma^4 \delta_{hm}$ for h, m > 0, we have

$$E\left[\sum_{t=1}^{n} \left(X_t^2 X_{t-h} X_{t-m} - \sigma^4 \delta_{hm}\right)\right]^2 \le Cn \quad \text{for any } h, m > 0.$$

It follows by Minkowski's inequality and Lemma A.1(ii) that

$$\begin{split} E(\hat{B}_{1n} - E\hat{B}_{1n})^2 &\leq n^{-2} \left\{ \sum_{h=1}^{n-1} \sum_{m=1}^n |b_J(h,m)| \left[E\left(\sum_{t=1}^n \left(X_t^2 X_{t-h} X_{t-m} - \sigma^4 \delta_{hm}\right)\right)^2 \right]^{1/2} \right\}^2 \\ &\leq C n^{-1} \left\{ \sum_{h=1}^{n-1} \sum_{m=1}^{n-1} |b_J(h,m)| \right\}^2 = O(2^{2J}/n). \end{split}$$

Hence, by Chebyshev's inequality, we have

$$\hat{B}_{1n} - E\hat{B}_{1n} = O_P(2^J/n^{1/2}).$$
(A.6)

Next, given $E(X_t^2 X_{t-h} X_{t-m}) = \sigma^4 \delta_{hm}$ for any h, m > 0, and Lemma A.1(v), we have

$$E\hat{B}_{1n} = \sigma^4 \sum_{h=1}^{n-1} b_J(h,h) = \sigma^4 (2^{J+1} - 1) \{1 + O[(J+1)/2^J + (2^J/n)^{2\alpha - 1}]\}.$$
 (A.7)

Combining (A.6), (A.7), $2^{2J}/n \to 0, J \to \infty$, and $\alpha > \frac{3}{2}$ then yields

$$2^{-J/2} \{ \hat{B}_{1n} - \sigma^4 (2^{J+1} - 1) \} = 2^{-J/2} \{ \hat{B}_{1n} - E \hat{B}_{1n} + E \hat{B}_{1n} - \sigma^4 (2^{J+1} - 1) \}$$

= $O_P (2^{J/2}/n^{1/2} + (J+1)/2^{J/2} + 2^{J(2\alpha - 1/2)}/n^{2\alpha - 1})$
= $O_P (1).$

(ii) Next, we consider \hat{B}_{2n} . Given Assumption 3, we have that for any h, m > 0,

$$E\left(\sum_{t=1}^{h}\sum_{s=m+1}^{n}X_{t}X_{t-h}X_{s}X_{s-m}\right)^{2} \leq \left[E\left(\sum_{t=1}^{h}X_{t}X_{t-h}\right)^{4}E\left(\sum_{s=m+1}^{n}X_{s}X_{s-m}\right)^{4}\right]^{1/2} \leq Cnh.$$

It follows by Minkowski's inequality and Lemma A.1(ii) that

$$\begin{split} E\hat{B}_{2n}^2 &\leq 4n^{-2} \left\{ \sum_{h=1}^{n-1} \sum_{m=1}^{n-1} |b_J(h,m)| \left[E\left(\sum_{t=1}^h \sum_{s=m+1}^n X_t X_{t-h} X_s X_{s-m} \right)^2 \right]^{1/2} \right\}^2 \\ &\leq 4Cn^{-1} \left\{ \sum_{h=1}^{n-1} \sum_{m=1}^{n-1} h^{1/2} |b_J(h,m)| \right\}^2 = O(2^{3J}/n). \end{split}$$

Hence, $2^{-J/2}\hat{B}_{2n} = O_P(2^J/n^{1/2}) = o_P(1)$ by Chebyshev's inequality and $2^{2J}/n \to 0$. (iii) By reasoning similar to (ii), we can obtain $2^{-J/2}\hat{B}_{3n} = O_P(2^J/n^{1/2}) = o_P(1)$.

Proof of Proposition 2. Given (A.4) and (A.5), $\hat{A}_n = \hat{U}_n + \hat{B}_{4n} + \hat{B}_{5n} + \hat{B}_{6n}$. It suffices to show $2^{-J/2}\hat{B}_{jn} \rightarrow^p 0$ for j = 4,5,6. (i) We first consider \hat{B}_{4n} . Given Assumption 3,

$$E\left(\sum_{t=2}^{n}\sum_{s=\max(t-q,1)}^{t-1}X_{t}X_{t-h}X_{s}X_{s-m}\right)^{2} \leq Cnq \quad \text{for } h, m > 0.$$

It follows by Minkowski's inequality and Lemma A.1(ii) that

$$E\hat{B}_{4n}^2 \leq 4n^{-2} \left\{ \sum_{h=1}^{n-1} \sum_{m=1}^{n-1} |b_J(h,m)| \left[E\left(\sum_{t=2}^n \sum_{s=t-q}^{t-1} X_t X_{t-h} X_s X_{s-m}\right)^2 \right]^{1/2} \right\}^2$$
$$\leq 4Cqn^{-1} \left\{ \sum_{h=1}^{n-1} \sum_{m=1}^{n-1} |b_J(h,m)| \right\}^2 = O(q2^{2J}/n).$$

This, along with Chebyshev's inequality and $q^{2}/n \rightarrow 0, 2^{2J}/n \rightarrow 0$, implies $2^{-J/2}\hat{B}_{4n} =$ $O_P(q^{1/2}2^{J/2}/n^{1/2}) = o_P(1).$

(ii) Next, we consider \hat{B}_{5n} . Define the partial sum

$$S_{t-q-1}(m) = \sum_{s=1}^{t-q-1} X_s X_{s-m}.$$
 (A.8)

Noting that $X_t X_{t-h}$ is independent of $S_{t-q-1}(m)$ for $0 \le h \le q$ and m > 0, we have

$$E\hat{B}_{5n}^{2} = 4n^{-2}E\left\{\sum_{t=q+2}^{n}\sum_{h=1}^{q}X_{t}X_{t-h}\sum_{m=q+1}^{n-1}b_{J}(h,m)S_{t-q-1}(m)\right\}^{2}$$
$$= 4\sigma^{8}n^{-2}\sum_{t=q+2}^{n}\sum_{h=1}^{q}\sum_{s=1}^{q-q-1}\sum_{m=q+1}^{n-1}b_{J}^{2}(h,m)$$
$$\leq 2\sigma^{8}\sum_{h=1}^{n-1}\sum_{m=q+1}^{n-1}b_{J}^{2}(h,m) = o(2^{J}),$$

where the last inequality follows from $q \to \infty, q/2^J \to \infty$, and Lemma A.1(vi). Therefore, $2^{-J/2}\hat{B}_{5n} \rightarrow^{p} 0$ by Chebyshev's inequality.

(iii) Finally, we consider \hat{B}_{6n} . Because X_t is independent of $X_{t-h}S_{t-q-1}(m)$ for h, m > 0,

$$E\hat{B}_{6n} = 4\sigma^2 n^{-2} \sum_{t=q+2}^{n} E\left[\sum_{h=q+1}^{n-1} \sum_{m=1}^{n-1} b_J(h,m) X_{t-h} S_{t-q-1}(m)\right]^2.$$
 (A.9)

We now decompose the expectation in (A.9):

$$E\left[\sum_{h=q+1}^{n-1}\sum_{m=1}^{n-1}b_{J}(h,m)X_{t-h}S_{t-q-1}(m)\right]^{2} = E\left[\sum_{h=q+1}^{n-1}\sum_{m=1}^{n-1}\sum_{s=1}^{t-q-1}b_{J}(h,m)X_{t-h}X_{s}X_{s-m}\right]^{2} \le 4(C_{1nt}+C_{2nt}+C_{3nt}),$$
(A.10)

where

$$\begin{split} C_{1nt} &= E \left[\sum_{h=q+1}^{n-1} \sum_{m=1}^{n-1} \sum_{s=1}^{t-q-1} b_J(h,m) X_{t-h} X_s X_{s-m} \mathbf{1}(t-h>s) \right]^2, \\ C_{2nt} &= E \left[\sum_{h=q+1}^{n-1} \sum_{m=1}^{n-1} \sum_{s=1}^{t-q-1} b_J(h,m) X_{t-h} X_s X_{s-m} \mathbf{1}(t-h=s) \right]^2, \\ C_{3nt} &= E \left[\sum_{h=q+1}^{n-1} \sum_{m=1}^{n-1} \sum_{s=1}^{t-q-1} b_J(h,m) X_{t-h} X_s X_{s-m} \mathbf{1}(t-h$$

For C_{1nt} , because X_{t-h} is independent of $\{X_s X_{s-m}\}_{s=1}^{t-q-1}$ for t-h > s and m > 0, we have

$$C_{1nt} = \sigma^{6} \sum_{h=q+1}^{n-1} \sum_{s=1}^{t-q-1} \sum_{m=1}^{n-1} b_{J}^{2}(h,m) \mathbf{1}(t-h>s) \le \sigma^{6} t \sum_{h=q+1}^{n-1} \sum_{m=1}^{n-1} b_{J}^{2}(h,m).$$
(A.11)

For C_{2nt} , noting that $X_{t-h}X_sX_{s-m} = X_{t-h}^2X_{t-h-m}$ given t - h = s, we have

$$C_{2nt} = E\left(\sum_{h=q+1}^{n-1} \sum_{m=1}^{n-1} b_J(h,m) X_{t-h}^2 X_{t-h-m}\right)^2 \le C\left(\sum_{h=q+1}^{n-1} \sum_{m=1}^{n-1} |b_J(h,m)|\right)^2.$$
 (A.12)

For C_{3nt} , noting that X_s is independent of $\{X_{t-h}X_{s-m}\}$ for t-h < s and m > 0, we have

$$C_{3nt} = \sigma^2 \sum_{s=1}^{t-q-1} E\left(\sum_{h=q+1}^{n-1} \sum_{m=1}^{n-1} b_J(h,m) X_{t-h} X_{s-m} \mathbb{1}(t-h < s)\right)^2,$$

where

$$E\left(\sum_{h=q+1}^{n-1}\sum_{m=1}^{n-1}b_{J}(h,m)X_{t-h}X_{s-m}1(t-h< s)\right)^{2}$$

$$\leq 2E\left(\sum_{h=q+1}^{n-1}\sum_{m=1}^{n-1}b_{J}(h,m)X_{t-h}X_{s-m}1(t-h\neq s-m)\right)^{2}$$

$$+ 2E\left(\sum_{h=q+1}^{n-1}\sum_{m=1}^{n-1}b_{J}(h,m)X_{t-h}X_{s-m}1(t-h=s-m)\right)^{2}$$

$$\leq C\sum_{h=q+1}^{n-1}\sum_{m=1}^{n-1}b_{J}^{2}(h,m) + C\left(\sum_{m=1}^{n-1}|b_{J}(m+t-s,m)|\right)^{2}.$$

It follows that

$$C_{3nt} \le Ct \sum_{h=q+1}^{n-1} \sum_{m=1}^{n-1} b_J^2(h,m) + C \left(\sum_{h=1}^{n-1} \sum_{m=1}^{n-1} |b_J(h,m)| \right)^2.$$
(A.13)

Combining (A.9)–(A.13), Lemma A.1(ii and vi), and $q \to \infty, q/2^J \to \infty$ yields

$$E\hat{B}_{6n}^2 \leq C\sum_{h=q+1}^{n-1}\sum_{m=1}^{n-1}b_J^2(h,m) + Cn^{-1}\left(\sum_{h=1}^{n-1}\sum_{m=1}^{n-1}|b_J(h,m)|\right)^2 = o(2^J) + O(2^{2J}/n).$$

Thus, $2^{-J/2}\hat{B}_{6n} \rightarrow^p 0$ by Chebyshev's inequality and $2^{2J}/n \rightarrow 0$, This completes the proof.

Proof of Proposition 3. We write $\hat{U}_n = n^{-1} \sum_{t=q+2}^n U_{nt}$, where

$$U_{nt} = 2X_t \sum_{h=1}^{q} X_{t-h} H_{t-q-1}(h),$$
(A.14)

where $H_{t-q-1}(h) = \sum_{m=1}^{q} b_J(h, m) S_{t-q-1}(m)$ and $S_{t-q-1}(m)$ is defined in (A.8). Let \mathcal{F}_t be the sigma field consisting of $X_s, s \leq t$. Because $\{X_t X_{t-h}\}$ is independent of $H_{t-q-1}(h)$ for $0 < h \leq q$, $\{U_{nt}, \mathcal{F}_{t-1}\}$ is an adapted martingale difference sequence, with

$$\lambda_n^2 = n^{-2} \sum_{t=q+2}^n EU_{nt}^2$$

= $4\sigma^8 n^{-2} \sum_{t=q+2}^n (t-q-1) \sum_{h=1}^q \sum_{m=1}^q b_J^2(h,m)$
= $2\sigma^8 (1-q/n)(1-(q+1)/n) \sum_{h=1}^q \sum_{m=1}^q b_J^2(h,m)$
= $4(2^{J+1}-1)\sigma^8 \{1+o(1)\},$ (A.15)

by Lemma A.1(vi) and $q \to \infty, q/2^J \to \infty, q^2/n \to 0$. It follows that $4(2^{J+1} - 1) \sigma^8/\lambda_n^2 \to 1$.

We apply Brown's (1971) martingale limit theorem by verifying his two conditions: (i) $\lambda_n^{-2} n^{-2} \sum_{t=q+2}^n E[U_{nt}^2 1(|U_{nt}| \ge \epsilon n \lambda_n)] \to 0$ for all $\epsilon > 0$ and (ii) $\lambda_n^{-2} n^{-2} \times \sum_{t=q+2}^n E(U_{nt}^2 | \mathcal{F}_{t-1}) \to^p 1$. Because

$$\lambda_n^{-2} n^{-2} \sum_{t=q+2}^n E[U_{nt}^2 1(|U_{nt}| > \epsilon n \lambda_n)] = \lambda_n^{-2} n^{-2} \sum_{t=q+2}^n \int_{|u| > \epsilon n \lambda_n} u^2 dF_{nt}(u)$$

$$\leq \lambda_n^{-4} n^{-4} \epsilon^{-2} \sum_{t=q+2}^n EU_{nt}^4,$$

where $F_{nt}(u)$ is the cumulative distribution function (c.d.f.) of U_{nt} , it suffices for (i) if $\lambda_n^{-4} n^{-4} \sum_{tq+2}^n EU_{nt}^4 \to 0$. Given the independence between X_{t-h} and $H_{t-q-1}(h)$ for $0 < h \le q$, we can use the iterated expectation $EU_{nt}^4 = E[E(U_{nt}^4 | \mathcal{F}_{t-q-1})]$ and obtain

$$EU_{nt}^{4} = 16\mu_{4}E\left[\sum_{h=1}^{q}X_{t-h}H_{t-q-1}(h)\right]^{4}$$

$$\leq 48\mu_{4}\left\{\sum_{h=1}^{q}(EX_{t-h}^{4})^{1/2}(EH_{t-q-1}^{4}(h))^{1/2}\right\}^{2}$$

$$\leq Ct^{2}\left\{\sum_{h=1}^{q}\sum_{m=1}^{q}b_{J}^{2}(h,m)\right\}^{2} = O(t^{2}2^{2J})$$

by Lemma A.1(vi), where we used the fact that given Assumption 3,

$$EH_{t-q-1}^{4}(h) = E\left(\sum_{s=1}^{t-q-1} X_{s} \sum_{m=1}^{q} b_{J}(h,m) X_{s-m}\right)^{4}$$

$$\leq C\left(\sum_{s=1}^{t-q-1} (EX_{s}^{4})^{1/2} \left[E\left(\sum_{m=1}^{q} b_{J}(h,m) X_{s-m}\right)^{4}\right]^{1/2}\right)^{2}$$

$$\leq Ct^{2}\left(\sum_{m=1}^{q} b_{J}^{2}(h,m)\right)^{2}.$$
(A.16)

It follows that condition (i) holds because $\lambda_n^{-4} n^{-4} \sum_{t=q+2}^n EU_{nt}^4 = O(n^{-1})$. Next, we verify condition (ii), which holds if $\lambda_n^{-4} E(\tilde{U}_n^2 - \lambda_n^2) \to 0$, where

$$\widetilde{U}_n^2 = n^{-2} \sum_{t=q+2}^n E(U_{nt}^2 | \mathcal{F}_{t-1}).$$

Given $E(X_{t-h_1}X_{t-h_2}) = \sigma^2 \delta_{h_1h_2}$ for $h_1, h_2 > 0$, we have

$$E(U_{nt}^{2}|\mathcal{F}_{t-1}) = 4\sigma^{2} \left(\sum_{h=1}^{q} X_{t-h} H_{t-q-1}(h)\right)^{2}$$

= $4\sigma^{4} \sum_{h=1}^{q} H_{t-q-1}^{2}(h) + 4\sigma^{2} \sum_{h_{2}=1}^{q} \sum_{h_{1}=1}^{q} (X_{t-h_{1}} X_{t-h_{2}} - \sigma^{2} \delta_{h_{1}h_{2}})$
 $\times H_{t-q-1}(h_{1}) H_{t-q-1}(h_{2})$
= $4\sigma^{4} \sum_{h=1}^{q} H_{t-q-1}^{2}(h) + V_{1nt}$, say. (A.17)

Next, using the definitions of $H_{t-q-1}(h)$ and $S_{t-q-1}(m)$ in (A.14) and (A.8), we decompose

$$4\sigma^{4} \sum_{h=1}^{q} H_{t-q-1}^{2}(h) = 4\sigma^{4} \sum_{h=1}^{q} \sum_{m_{2}=1}^{q} \sum_{m_{1}=1}^{q} b_{J}(h,m_{1})b_{J}(h,m_{2})S_{t-q-1}(m_{1})S_{t-q-1}(m_{2})$$

$$= 4\sigma^{4} \sum_{h=1}^{q} \sum_{m_{2}=1}^{q} \sum_{m_{1}=1}^{q} b_{J}(h,m_{1})b_{J}(h,m_{2}) \sum_{s=1}^{t-q-1} X_{s}^{2}X_{s-m_{1}}X_{s-m_{2}}$$

$$+ 8\sigma^{4} \sum_{h=1}^{q} \sum_{m_{2}=1}^{q} \sum_{m_{1}=1}^{q} b_{J}(h,m_{1})b_{J}(h,m_{2})$$

$$\times \sum_{s_{2}=2}^{t-q-1} \sum_{s_{1}=1}^{s_{2}-1} X_{s_{2}}X_{s_{2}-m_{2}}X_{s_{1}}X_{s_{1}-m_{1}}$$

$$= W_{nt} + V_{2nt}, \quad \text{say.}$$
(A.18)

Finally, noting that $E(X_s X_{s-m_1} X_{s-m_2}) = \sigma^4 \delta_{m_1 m_2}$ for $m_1, m_2 > 0$, we obtain

$$W_{nt} = 4(t-q-1)\sigma^8 \sum_{h=1}^q \sum_{m=1}^q b_J^2(h,m)$$

+ $4\sigma^4 \sum_{h=1}^q \sum_{m_2=1}^q \sum_{m_1=1}^q b_J(h,m_1)b_J(h,m_2) \sum_{s=1}^{t-q-1} (X_s^2 X_{s-m_1} X_{s-m_2} - \sigma^4 \delta_{m_1m_2})$
= $EU_{nt}^2 + V_{3nt}$, say. (A.19)

It follows from (A.17)–(A.19), C_r inequality, Lemma A.2, which follows, and (A.15) that

$$\lambda_n^{-4} E(\tilde{U}_n^2 - \lambda_n^2)^2 = \lambda_n^{-4} E\left(\sum_{j=1}^3 n^{-2} \sum_{t=q+2}^n V_{jnt}\right)^2 \le 4\lambda_n^{-4} \sum_{j=1}^3 E\left(n^{-2} \sum_{t=q+2}^n V_{jnt}\right)^2$$
$$= O\{q/n + (J+1)/2^J\} \to 0$$

given $q^2/n \to 0, J \to \infty$. Hence, condition (ii) of Brown (1971) holds, and so $\lambda_n^{-1} \hat{U}_n \to^d N(0,1)$. The proof of Proposition 3 will be completed if Lemma A.2 is shown.

LEMMA A.2. Suppose that the conditions of Proposition 3 hold. Then

(i) $E(n^{-2}\sum_{t=q+2}^{n}V_{1nt})^2 = O(q2^{2J}/n);$ (ii) $E(n^{-2}\sum_{t=q+2}^{n}V_{2nt})^2 = O\{q2^{2J}/n + (J+1)2^J\};$ (iii) $E(n^{-2}\sum_{t=q+2}^{n}V_{3nt})^2 = O(2^{2J}/n).$

Proof of Lemma A.2. (i) We first write

$$E\left(\sum_{t=q+2}^{n} V_{1nt}\right)^{2} \leq 2\sum_{t_{2}=q+2}^{n} \sum_{t_{1}=q+2}^{t_{2}} E(V_{1nt_{2}}V_{1nt_{1}})$$

$$= 2\sum_{t_{2}=q+2}^{n} \sum_{t_{1}=\max(q+2, t_{2}-q)}^{t_{2}} E(V_{1nt_{2}}V_{1nt_{1}}) + 2\sum_{t_{2}=2q+3}^{n} \sum_{t_{1}=q+2}^{t_{2}-q-1} E(V_{1nt_{2}}V_{1nt_{1}}).$$

(A.20)

Because $\{X_{t_2-h}\}_{h=1}^{q}$ is independent of $\{H_{t_2-q-1}(h)\}_{h=1}^{q}$, we have $E(V_{1nt_2}|\mathcal{F}_{t_2-q-1}) = 0$. Moreover, $\{X_{t_2-h}\}_{n=1}^{q}$ is also independent of V_{1nt_1} for $t_2 - t_1 > q$. Hence, we have $E(V_{1nt_2}|\mathcal{F}_{t_2-q-1})V_{1nt_1}] = 0$ when $t_2 - t_1 > q$. Thus, the second term in (A.20) is zero.

We now compute the order of magnitude for the first term. Again, using the facts that $\{X_{t-h}\}_{h=1}^{q}$ is independent of $\{H_{t-q-1}(h)\}_{h=1}^{q}$ and $EV_{1nt}^{2} = E[E(V_{1nt}^{2}|\mathcal{F}_{t-q-1})]$, we obtain

$$EV_{1nt}^{2} = E \left[8\sigma^{2} \sum_{h_{2}=1}^{q} \sum_{h_{1}=1}^{q} (X_{t-h_{1}}X_{t-h_{2}} - \sigma^{2}\delta_{h_{1}h_{2}})H_{t-q-1}(h_{1})H_{t-q-1}(h_{2}) \right]^{2}$$

$$\leq C \sum_{h_{2}=1}^{q} \sum_{h_{1}=1}^{q} E \{H_{t-q-1}^{2}(h_{1})H_{t-q-1}^{2}(h_{2})\}$$

$$\leq Ct^{2} \left(\sum_{h=1}^{q} \sum_{m=1}^{q} b_{j}^{2}(h,m) \right)^{2} = O(t^{2}2^{2J})$$
(A.21)

given Lemma A.1(vi), where we made use of the fact that

$$\begin{split} E\{H_{t-q-1}^2(h_1)H_{t-q-1}^2(h_2)\} &\leq \{EH_{t-q-1}^4(h_1)EH_{t-q-1}^4(h_2)\}^{1/2} \\ &\leq Ct^2\sum_{m=1}^q b_J^2(h_1,m)\sum_{m=1}^q b_J^2(h_2,m) \end{split}$$

by Cauchy–Schwarz inequality and (A.16). Hence, we have $E(n^{-2}\sum_{l=q+2}^{n}V_{1nl})^2 = O(q2^{2J}/n)$ from (A.20) and (A.21) and Cauchy–Schwarz inequality.

(ii) We decompose V_{2nt} into the sums with $s_2 - s_1 \le q$ and $s_2 - s_1 > q$:

$$V_{2nt} = 8\sigma^{4} \sum_{h=1}^{q} \sum_{m_{2}=1}^{q} \sum_{m_{1}=1}^{q} b_{J}(h,m_{1})b_{J}(h,m_{2}) \left(\sum_{s_{2}=2}^{t-q-1} \sum_{s_{1}=\max(1,s_{2}-q)}^{s_{2}-1} + \sum_{s_{2}=q+2}^{t-q-1} \sum_{s_{1}=1}^{s_{2}-q-1}\right) \times X_{s_{2}}X_{s_{2}-m_{2}}X_{s_{1}}X_{s_{1}-m_{1}} = V_{21nt} + V_{22nt}, \text{ say.}$$
(A.22)

We first consider V_{21nt} . For any $m_1, m_2 > 0$, we have

$$E\left(\sum_{s_2=2}^{t-q-1} X_{s_2} \sum_{s_1=\max(1,s_2-q)}^{s_2-1} X_{s_2-m_2} X_{s_1} X_{s_1-m_1}\right)^2$$

= $\sigma^2 \sum_{s_2=2}^{t-q-1} E\left(\sum_{s_1=\max(1,s_2-q)}^{s_2-1} X_{s_2-m_2} X_{s_1} X_{s_1-m_1}\right)^2$
= $\sigma^2 \sum_{s_2=2}^{t-q-1} \sum_{s_1=\max(1,s_2-q)}^{s_2-1} E(X_{s_2-m_2} X_{s_1} X_{s_1-m_1})^2 \le Ctq.$

Hence, by Minkowski's inequality and Lemma A.1(iii),

$$E\left(n^{-2}\sum_{t=q+2}^{n}V_{21nt}\right)^{2} \leq Cqn^{-1}\left\{\sum_{h=1}^{q}\sum_{m_{2}=1}^{q}\sum_{m_{1}=1}^{q}|b_{J}(h,m_{1})b_{J}(h,m_{2})|\right\}^{2} = O(q2^{2J}/n).$$
(A.23)

Next, we consider V_{22nt} . Put

$$Z_{s-1}(h) = \sum_{m=1}^{q} b_J(h,m) X_{s-m}.$$

Then we can write

$$V_{22nt} = 8\sigma^4 \sum_{h=1}^{q} \sum_{s_2=q+2}^{t-q-1} \sum_{s_1=1}^{s_2-q-1} X_{s_2} Z_{s_2-1}(h) X_{s_1} Z_{s_1-1}(h).$$

Because X_s is independent of $Z_{s-1}(h)$ for h > 0, and because $\{X_{s_2}, Z_{s_2-1}(h)\}$ is independent of $\{X_{s_1}, Z_{s_1-1}(h)\}$ for $s_2 - s_1 > q$ and $0 < h \le q$, we have

$$\begin{split} E(V_{22nt}^2) &= 64\sigma^{10} \sum_{s_2=q+2}^{t-q-1} E\left\{\sum_{h=1}^q Z_{s_2-1}(h) \sum_{s_1=1}^{s_2-q-1} X_{s_1} Z_{s_1-1}(h)\right\}^2 \\ &= 64\sigma^{10} \sum_{s_2=q+2}^{t-q-1} \sum_{h_1=1}^q \sum_{h_2=1}^q E\{Z_{s_2-1}(h_1) Z_{s_2-1}(h_2)\} \\ &\times E\left\{\left(\sum_{s_1=1}^{s_2-q-1} X_{s_1} Z_{s_1-1}(h_1)\right) \left(\sum_{s_1'=1}^{s_2-q-1} X_{s_1'} Z_{s_1'-1}(h_2)\right)\right\} \right\} \\ &= 64\sigma^{12} \sum_{s_2=q+2}^{t-q-1} \sum_{s_1=1}^q \sum_{h_1=1}^q \sum_{h_2=1}^q E\{Z_{s_2-1}(h_1) Z_{s_2-1}(h_2)\} \\ &\times E\{Z_{s_1-1}(h_1) Z_{s_1-1}(h_2)\} \\ &\leq 32\sigma^{16}t^2 \sum_{h_1=1}^q \sum_{h_2=1}^q \left(\sum_{m=1}^q b_J(h_1,m) b_J(h_2,m)\right)^2, \end{split}$$

where the last inequality follows from the fact that

$$E\{Z_{s-1}(h_1)Z_{s-1}(h_2)\} = \sigma^2 \sum_{m=1}^q b_J(h_1,m)b_J(h_2,m).$$

It follows from Minkowski's inequality and Lemma A.1(iv) that

$$E\left(n^{-2}\sum_{t=q+2}^{n}V_{22nt}\right)^{2} \le \left(n^{-2}\sum_{t=q+2}^{n}(EV_{22nt}^{2})^{1/2}\right)^{2} = O\{(J+1)2^{J}\}.$$
(A.24)

Combining (A.22)–(A.24) yields $E(n^{-2}\sum_{t=q+2}^{n}V_{2nt})^2 = O\{q2^{2J}/n + (J+1)2^J\}.$ (iii) Write

$$V_{3nt} = 4\sigma^{4} \sum_{h=1}^{q} \sum_{m_{2}=1}^{q} \sum_{m_{1}=1}^{q} b_{J}(h, m_{1})b_{J}(h, m_{2})$$

$$\times \left\{ \sum_{s=1}^{t-q-1} (X_{s}^{2} - \sigma^{2})X_{s-m_{1}}X_{s-m_{2}} + \sum_{s=1}^{t-q-1} \sigma^{2}(X_{s-m_{1}}X_{s-m_{2}} - \sigma^{2}\delta_{m_{1}m_{2}}) \right\} = V_{31nt} + V_{32nt}, \quad \text{say.}$$
(A.25)

By Minkowski's inequality, we have

$$EV_{31nt}^{2} \leq 16\sigma^{8} \left\{ \sum_{h=1}^{q} \sum_{m_{2}=1}^{q} \sum_{m_{1}=1}^{q} |b_{J}(h,m_{1})b_{J}(h,m_{2})| \\ \times \left[E\left(\sum_{s=1}^{t-q-1} (X_{s}^{2} - \sigma^{2})X_{s-m_{1}}X_{s-m_{2}}\right)^{2} \right]^{1/2} \right\}^{2} \\ \leq Ct \left\{ \sum_{h=1}^{q} \left(\sum_{m=1}^{q} |b_{J}(h,m)| \right)^{2} \right\}^{2} = O(t2^{2J})$$
(A.26)

by Lemma A.1(iii). Similarly, we have

$$EV_{32nt}^2 \le Ct \left\{ \sum_{h=1}^q \left(\sum_{m=1}^q |b_J(h,m)| \right)^2 \right\}^2 = O(t2^{2J}).$$
(A.27)

It follows from (A.25)–(A.27) and Minkowski's inequality that $E(n^{-2}\sum_{t=q+2}^{n}V_{3nt})^2 = O(2^{2J}/n)$.

Proof of Theorem 2. Given $2^{3J/2}/n \to 0$, we have $\{2(2^{J+1} - 1)^{1/2}/n\}W_n = 2\pi Q(\hat{f};f_0) + o(1)$. To show $Q(\hat{f};f_0) \to^p Q(f;f_0)$, we decompose

$$Q(\hat{f};f_0) = Q(f;f_0) + Q(\hat{f};f) + 2\int_{-\pi}^{\pi} \{\hat{f}(\omega) - f(\omega)\}\{f(\omega) - f_0(\omega)\}d\omega.$$

It suffices to show $Q(\hat{f};f) \to^p 0$, because the last term is $o_P(1)$ by Cauchy–Schwarz inequality.

Now, define the pseudo estimator

$$\tilde{f}(\omega) = (2\pi)^{-1} + \sum_{j=0}^{J} \sum_{k=1}^{2^{j}} \tilde{\alpha}_{jk} \Psi_{jk}(\omega),$$
(A.28)

where $\tilde{\alpha}_{jk} = \sum_{h=1-n}^{n-1} \tilde{\rho}(h) \hat{\psi}_{jk}(2\pi h)$ and $\tilde{\rho}(h) = \hat{R}(h)/\sigma^2$. Also, put

$$f_J(\omega) = (2\pi)^{-1} + \sum_{j=0}^{J} \sum_{k=1}^{2^j} \alpha_{jk} \Psi_{jk}(\omega).$$
(A.29)

Noting that $\hat{f}(\omega) - f(\omega) = \hat{f}(\omega) - \tilde{f}(\omega) + \tilde{f}(\omega) - E\tilde{f}(\omega) + E\tilde{f}(\omega) - f_J(\omega) + f_J(\omega) - f(\omega)$, we have

$$Q(\hat{f};f) \le 8Q(\hat{f};\tilde{f}) + 8Q(\tilde{f};E\tilde{f}) + 8Q(E\tilde{f};f_J) + 8Q(f_J;f).$$

We shall show that (i) $Q(\hat{f};\tilde{f}) \to^p 0$; (ii) $Q(\tilde{f};E\tilde{f}) \to^p 0$; (iii) $Q(E\tilde{f};f_J) \to 0$; (iv) $Q(f_J;f) \to 0$.

(i) By the orthonormality (28) and noting that $\hat{\alpha}_{ik} = \{\sigma^2/\hat{R}(0)\}\tilde{\alpha}_{ik}$, we have

$$Q(\hat{f};\tilde{f}) = \sum_{j=0}^{J} \sum_{k=1}^{2^{j}} (\hat{\alpha}_{jk} - \tilde{\alpha}_{jk})^{2} = \{\sigma^{2}/\hat{R}(0) - 1\}^{2} \sum_{j=0}^{J} \sum_{k=1}^{2^{j}} \tilde{\alpha}_{jk}^{2} \to^{p} 0$$

given $\hat{R}(0) - \sigma^2 \rightarrow^p 0$ and

$$\sum_{j=0}^{J} \sum_{k=1}^{2^{j}} \tilde{\alpha}_{jk}^{2} \le 2Q(\tilde{f}; f_{J}) + 2\sum_{j=0}^{\infty} \sum_{k=1}^{2^{j}} \alpha_{jk}^{2} = O_{P}(1),$$

where $Q(\tilde{f}; f_J) \leq 4Q(\tilde{f}; E\tilde{f}) + 4Q(E\tilde{f}; f_J) \rightarrow^p 0$ by (ii) and (iii) as shown subsequently.

(ii) By the orthonormality (28) and noting that $\tilde{\alpha}_{jk}$ is real valued, we have

$$Q(\tilde{f}; E\tilde{f}) = \sum_{j=0}^{J} \sum_{k=1}^{2^{j}} (\tilde{\alpha}_{jk} - E\tilde{\alpha}_{jk})^{2} = \sigma^{-4} \sum_{j=0}^{J} \sum_{k=1}^{2^{j}} \left| \sum_{h=1-n}^{n-1} [\hat{R}(h) - E\hat{R}(h)] \hat{\psi}_{jk}(2\pi h) \right|^{2}.$$

Recalling the definitions of $a_J(h,m)$ and $b_J(h,m)$ in Lemma A.1, we obtain

$$0 \leq EQ(\tilde{f}; E\tilde{f}) = \sigma^{-4} \sum_{h=1-n}^{n-1} \sum_{m=1-n}^{n-1} |a_J(h,m)| \operatorname{Cov}\{\hat{R}(h), \hat{R}(m)\}$$

$$= \sigma^{-4} \sum_{h=1}^{n-1} \sum_{m=1}^{n-1} |b_J(h,m)| \operatorname{Cov}\{\hat{R}(h), \hat{R}(m)\}$$

$$\leq \sup_{0 \leq h \leq n} \operatorname{Var}\{\hat{R}(h)\} \sum_{h=1}^{n-1} \sum_{m=1}^{n-1} |b_J(h,m)| = O(2^J/n),$$
(A.30)

where the last inequality follows from Cauchy–Schwarz inequality and the last equality follows from Lemma A.1(ii) and $\sup_{0 \le h \le n} \operatorname{Var}{\hat{R}(h)} = O(n^{-1})$, which follows from $\sum_{l=1}^{\infty} R^2(l) \le \infty, \sum_{j=-\infty}^{\infty} \sum_{k=-\infty}^{\infty} \sum_{l=-\infty}^{\infty} |\kappa(j,k,l)| \le \infty$, and

$$\operatorname{Var}\{\hat{R}(h)\} = n^{-1} \sum_{l=1-n}^{n-1} (1 - |l|/n) \{R^2(l) + R(l-h)R(l+h) + \kappa(h, l, l+h)\}$$

(cf. Hannan, 1970, p. 209). Hence, $Q(\tilde{f}; E\tilde{f}) \to^p 0$ by (A.30) and Markov's inequality. (iii) We now show $Q(E\tilde{f}; f_J) \to 0$. From the definition of $\tilde{\alpha}_{jk}$ in (A.28), we have

$$E\tilde{\alpha}_{jk} - \alpha_{jk} = \sigma^{-2} \sum_{h=1-n}^{n-1} (1 - |h|/n) R(h) \hat{\psi}_{jk}(2\pi h) - \sigma^{-2} \sum_{h=-\infty}^{\infty} R(h) \hat{\psi}_{jk}(2\pi h)$$
$$= -\sigma^{-2} n^{-1} \sum_{h=1-n}^{n-1} |h| R(h) \hat{\psi}_{jk}(2\pi h) - \sigma^{-2} \sum_{|h| \ge n} R(h) \hat{\psi}_{jk}(2\pi h).$$

It follows that

$$Q(E\tilde{f};f_J) = \sum_{j=0}^{J} \sum_{k=1}^{2^J} (E\tilde{\alpha}_{jk} - \alpha_{jk})^2$$

$$\leq 2\sigma^{-4} n^{-2} \sum_{j=0}^{J} \sum_{k=1}^{2^J} \left\{ \sum_{h=1-n}^{n-1} |h| R(h) \hat{\psi}_{jk}(2\pi h) \right\}^2$$

$$+ 2\sigma^{-4} \sum_{j=0}^{J} \sum_{k=1}^{2^J} \left\{ \sum_{|h| \ge n} R(h) \hat{\psi}_{jk}(2\pi h) \right\}^2$$

$$= 2\sigma^{-4} M_{1n} + 2\sigma^{-4} M_{2n}, \text{ say.}$$
(A.31)

For the first term M_{1n} , by Cauchy–Schwarz inequality and (16), we have

$$M_{1n} \leq n^{-2} \left\{ \sum_{h=1-n}^{n-1} R^2(h) \right\} \left\{ \sum_{j=0}^{J} \sum_{k=1}^{2^j} \sum_{h=1-n}^{n-1} h^2 |\hat{\psi}_{jk}(2\pi h)|^2 \right\}$$

$$\leq n^{-2} \left\{ \sum_{h=1-n}^{n-1} R^2(h) \right\} \sum_{j=0}^{J} 2^{3j} \left\{ (2\pi/2^j) \sum_{h=1-n}^{n-1} (2\pi h/2^j)^2 |\hat{\psi}(2\pi h/2^j)|^2 \right\}$$

$$= O(2^{3J}/n^2)$$
(A.32)

given $\sum_{h=-\infty}^{\infty} R^2(h) < \infty$ and the fact that given Assumption 2(i),

$$(2\pi/2^{j})\sum_{h=1-n}^{n-1}(2\pi h/2^{j})^{2}|\hat{\psi}(2\pi h/2^{j})|^{2} \leq C\int_{-\infty}^{\infty}z^{2}/(1+|z|)^{2\alpha}dz < \infty, \quad \alpha > \frac{3}{2}.$$

For the second term M_{2n} in (A.31), we have, by Cauchy–Schwarz inequality and (16),

$$\sum_{j=0}^{J} \sum_{k=1}^{2^{J}} \left\{ \sum_{|h| \ge n} R(h) \hat{\psi}_{jk}(2\pi h) \right\}^{2} \le \left\{ \sum_{h \ge n} R^{2}(h) \right\} \sum_{j=0}^{J} \sum_{h \ge n} |\hat{\psi}(2\pi h/2^{j})|^{2} = o(2^{2\alpha J}/n^{2\alpha - 1}),$$
(A.33)

where the last equality follows from $\sum_{h\geq n} R^2(h) \to 0$ and

$$\sum_{j=0}^{J} \sum_{|h| \ge n} |\hat{\psi}(2\pi h/2^{j})|^{2} \le C^{2} \sum_{j=0}^{J} \sum_{|h| \ge n} |2\pi h/2^{j}|^{-2\alpha} = O(2^{2\alpha J}/n^{2\alpha - 1})$$
(A.34)

by Assumption 2(i). Combining (A.31)–(A.34), $2^{3J/2}/n \rightarrow 0$, and $\alpha > \frac{3}{2}$ yields $Q(E\hat{f};f_J) \rightarrow 0$.

(iv) By the orthonormality (28) and $\sum_{j=0}^{\infty} \sum_{k=1}^{2^j} \alpha_{jk}^2 = \int_{-\pi}^{\pi} f^2(\omega) d\omega = \sum_{l=-\infty}^{\infty} \rho^2(l) < \infty$, we have $Q(f_j; f) = \sum_{j=J+1}^{\infty} \sum_{k=1}^{2^j} \alpha_{jk}^2 \to 0$ as $J \to \infty$. This completes the proof.

APPENDIX B

Proof of Lemma A.1. (i) Given (16), $\hat{\psi}(0) = 0$, and $\hat{\psi}^*(-z) = \hat{\psi}(-z)$, we have $a_J(0,m) = a_J(h,0) = 0, a_J^*(h,m) = a_J(m,h) = a_J(-h,-m)$. Hence, $b_J(0,m) = b_J(h,0) = 0$, and

$$b_J(h,m) = a_J(h,m) + a_J^*(h,m) + a_J(h,-m) + a_J^*(h,-m)$$

= $a_J(h,m) + a_J(m,h) + a_J(h,-m) + a_J(m,-h).$ (B.1)

The first equality in (B.1) implies that $b_J(h,m)$ is real valued, and the second one implies that $b_J(h,m) = b_J(m,h)$.

(ii) Put
$$c_j(h,m) = 2^{-j} \sum_{k=1}^{2^j} e^{i2\pi (m-h)k/2^j}$$
. Then, using (16), we obtain
 $a_J(h,m) = 2\pi \sum_{j=0}^{J} c_j(h,m) \hat{\psi}(2\pi h/2^j) \hat{\psi}^*(2\pi m/2^j)$
 $= \begin{cases} 2\pi \sum_{j=0}^{J} \hat{\psi}(2\pi h/2^j) \hat{\psi}^*(2\pi m/2^j), & \text{if } m-h=2^jr & \text{for some } r \in \mathbb{Z}, \\ 0, & \text{otherwise,} \end{cases}$

where we used the well-known identity that $c_j(h, m) = 1$ if $m - h = 2^j r$ for some $r \in \mathbb{Z}$ and $c_j(h, m) = 0$ otherwise (cf. Priestley, 1981, (6.19), p. 392).

Now, by the triangle inequality, reindexing, and (B.2), we obtain

$$\begin{split} \sum_{h=1}^{n-1} \sum_{m=1}^{n-1} h^{\nu} |b_{J}(h,m)| &\leq \sum_{h=-\infty}^{\infty} \sum_{m=-\infty}^{\infty} |h|^{\nu} |a_{J}(h,m)| \\ &\leq 2\pi \sum_{j=0}^{J} \sum_{h=-\infty}^{\infty} \sum_{r=-\infty}^{\infty} |h|^{\nu} |\hat{\psi}(2\pi h/2^{j}) \hat{\psi}^{*}(2\pi h/2^{j}+2\pi r)| \\ &\leq \sum_{j=0}^{J} 2^{j(1+\nu)} \left((2\pi/2^{j}) \sum_{h=-\infty}^{\infty} |2\pi h/2^{j}|^{\nu} |\hat{\psi}(2\pi h/2^{j})| \right) \\ &\times \left(\sum_{r=-\infty}^{\infty} |\hat{\psi}(2\pi h/2^{j}+2\pi r)| \right) = O(2^{(1+\nu)J}), \end{split}$$

where we used the facts that given Assumption 2(i) and $\nu \leq \frac{1}{2}$,

$$(2\pi/2^{j})\sum_{h=-\infty}^{\infty} |2\pi h/2^{j}|^{\nu} |\hat{\psi}(2\pi h/2^{j})| \leq C \int_{-\infty}^{\infty} |z|^{\nu}/(1+|z|)^{\alpha} dz < \infty,$$
(B.3)
$$\sup_{z \in \mathbb{R}} \sum_{r=-\infty}^{\infty} |\hat{\psi}(z+2\pi r)| \leq C.$$
(B.4)

(iii) By reindexing and (B.2), we have

$$\begin{split} \sum_{h=1}^{n-1} \left(\sum_{m=1}^{n-1} |b_J(h,m)| \right)^2 &\leq C \sum_{h=-\infty}^{\infty} \left(\sum_{m=-\infty}^{\infty} |a_J(h,m)| \right)^2 \\ &\leq C \sum_{h=-\infty}^{\infty} \left(2\pi \sum_{j=0}^{J} |\hat{\psi}(2\pi h/2^j)| \sum_{r=-\infty}^{\infty} |\hat{\psi}(2\pi h/2^j + 2\pi r)| \right)^2 \\ &\leq C^2 \sum_{h=-\infty}^{\infty} \left(2\pi \sum_{j=0}^{J} |\hat{\psi}(2\pi h/2^j)| \right)^2 \text{ given (B.4)} \\ &\leq 2\pi C^2 \left\{ \sum_{j=0}^{J} 2^{j/2} \left[(2\pi/2^j) \sum_{h=-\infty}^{\infty} |\hat{\psi}(2\pi h/2^j)|^2 \right]^{1/2} \right\}^2 \\ &= O(2^J), \end{split}$$

where we used the orthonormality (17) in obtaining the last equality.

(iv) We first show
$$\sup_{h\in\mathbb{Z}} \sum_{m=1}^{n-1} |b_J(h,m)|^2 \leq C(J+1)$$
. By reindexing and (B.2),

$$\sum_{m=-\infty}^{\infty} |a_J(h,m)|^2 = (2\pi)^2 \sum_{j=0}^{J} \sum_{m=-\infty}^{\infty} |c_j(h,m)|^2 |\hat{\psi}(2\pi h/2^j)|^2 |\hat{\psi}(2\pi m/2^j)|^2
+ 2(2\pi)^2 \operatorname{Re} \sum_{d=1}^{J} \sum_{j=d}^{J} \sum_{m=-\infty}^{\infty} c_j(h,m) c_{j-d}^*(h,m)
\times \hat{\psi}(2\pi h/2^j) \hat{\psi}^* (2^d 2\pi h/2^j) \hat{\psi}^* (2\pi m/2^j) \hat{\psi}(2^d 2\pi m/2^j)
= 2\pi \sum_{j=0}^{J} |\hat{\psi}(2\pi h/2^j)|^2 \left(2\pi \sum_{r=-\infty}^{\infty} |\hat{\psi}(2\pi h/2^j + 2\pi r)|^2\right)
+ 4\pi \operatorname{Re} \sum_{d=1}^{J} \sum_{j=d}^{J} \hat{\psi}(2\pi h/2^j) \hat{\psi}^* (2^d 2\pi h/2^j)
\times \left(2\pi \sum_{r=-\infty}^{\infty} \hat{\psi}^* (2\pi h/2^j + 2\pi r) \hat{\psi}(2^d (2\pi h/2^j + 2\pi r))\right)
= \sum_{j=0}^{J} |\hat{\psi}(2\pi h/2^j)|^2,$$
(B.5)

where the last equality follows from the orthonormality that for any $d \ge 0$,

$$2\pi \sum_{r=-\infty}^{\infty} \hat{\psi}(z+2\pi r)\hat{\psi}^*(2^d(z+2\pi r)) = \delta_{0d} \quad \text{a.e. } z \in \mathbb{R}$$
(B.6)

(cf. Hernández and Weiss, 1996, (1.2) and its proof, pp. 101–102). It follows from Cauchy–Schwarz inequality, $b_J(h,m) = b_J(m,h)$, Lemma A.1(iii), (B.5), and $|\hat{\psi}(z)| \leq C$ that

$$\begin{split} \sum_{h_1=1}^{n-1} \sum_{h_2=1}^{n-1} \left(\sum_{m=1}^{n-1} |b_J(h_1, m) b_J(h_2, m)| \right)^2 \\ &\leq \sup_{h \in \mathbb{Z}} \sum_{m=-\infty}^{\infty} |b_J(h, m)|^2 \\ &\times \sum_{h_1=1}^{n-1} \sum_{h_2=1}^{n-1} \left(\sum_{m=1}^{n-1} |b_J(h_1, m) b_J(h_2, m)| \right) \\ &= \sup_{h \in \mathbb{Z}} \sum_{m=-\infty}^{\infty} |b_J(h, m)|^2 \left(\sum_{h=1}^{n-1} \left(\sum_{m=1}^{n-1} |b_J(h, m)| \right)^2 \right) \\ &= O\{(J+1)2^J\}. \end{split}$$

(v) By $a_J(0,h) = a_J(h,0) = 0$ and reindexing, we have

$$\sum_{h=1}^{n-1} b_J(h,h) = \sum_{h=1-n}^{n-1} a_J(h,h) + \sum_{h=1-n}^{n-1} a_J(h,-h).$$
(B.7)

Using (B.2) and the orthonormality (17), we have

$$\sum_{h=1-n}^{n-1} a_J(h,h) = \sum_{j=0}^{J} 2^j \left\{ (2\pi/2^j) \sum_{h=1-n}^{n-1} |\hat{\psi}(2\pi h/2^j)|^2 \right\}$$
$$= \sum_{j=0}^{J} 2^j \left\{ 1 - (2\pi/2^j) \sum_{|h| \ge n} |\hat{\psi}(2\pi h/2^j)|^2 \right\}$$
$$= (2^{J+1} - 1)\{1 + O[(2^J/n)^{2\alpha - 1}]\},$$
(B.8)

where we used $\sum_{j=0}^{J} 2^j = 2^{J+1} - 1$ and (A.34). For the second term in (B.7), using (B.2) and Assumptions 1 and 2(i), we obtain

$$\left|\sum_{h=1-n}^{n-1} a_J(h,-h)\right| \le \sum_{h=-\infty}^{\infty} |a_J(h,-h)| \le \sum_{j=0}^{J} \sum_{r=-\infty}^{\infty} |\hat{\psi}(\pi r)|^2 \le C(J+1).$$
(B.9)

Combining (B.7)–(B.9), $2^{3J/2}/n \to 0, J \to \infty$, and $\alpha > \frac{3}{2}$ then yields $\sum_{h=1}^{n-1} b_J(h,h) = (2^{J+1}-1)\{1 + O[(J+1)/2^J + (2^J/n)^{2\alpha-1}]\}$. (vi) Using a(h,0) = a(0,m) = 0 and reindexing, we have

$$\begin{split} \sum_{h=1}^{n-1} \sum_{m=1}^{n-1} b_J^2(h,m) &= \sum_{h=1}^{n-1} \sum_{m=1}^{n-1} \left[a_J(h,m) + a_J(-h,-m) + a_J(h,-m) + a_J(-h,m) \right]^2 \\ &= \sum_{h=1}^{n-1} \sum_{m=1}^{n-1} \left[a_J^2(h,m) + a_J^2(-h,-m) + a_J^2(h,-m) + a_J^2(-h,m) \right] \\ &+ 2 \sum_{h=1}^{n-1} \sum_{m=1}^{n-1} \left[a_J(h,m) a_J(-h,-m) + a_J(h,-m) a_J(-h,m) \right] \\ &+ 2 \sum_{h=1}^{n-1} \sum_{m=1}^{n-1} \left[a_J(h,m) a_J(h,-m) + a_J(-h,-m) a_J(-h,m) \right] \\ &+ 2 \sum_{h=1}^{n-1} \sum_{m=1}^{n-1} \left[a_J(h,m) a_J(-h,m) + a_J(-h,-m) a_J(-h,m) \right] \\ &= \sum_{h=1-n}^{n-1} \sum_{m=1-n}^{n-1} \left[a_J(h,m) a_J(-h,m) + \sum_{h=1-n}^{n-1} \sum_{m=1-n}^{n-1} \left| a_J(h,m) a_J(-h,m) + \sum_{h=1-n}^{n-1} \sum_{m=1-n}^{n-1} a_J(h,m) a_J(-h,m) \right|^2 \\ &+ \sum_{h=1-n}^{n-1} \sum_{m=1-n}^{n-1} a_J(h,m) a_J(h,-m) + \sum_{h=1-n}^{n-1} \sum_{m=1-n}^{n-1} a_J(h,m) a_J(-h,m) \\ &= A_{1n} + A_{2n} + A_{3n} + A_{4n}, \quad \text{say.} \end{split}$$

We first consider A_{1n} . Write

$$A_{1n} = \sum_{h=1-n}^{n-1} \sum_{m=-\infty}^{\infty} a_j^2(h,m) - \sum_{h=1-n}^{n-1} \sum_{|m| \ge n} a_j^2(h,m) = A_{11n} - A_{12n}, \quad \text{say.}$$
(B.11)

Following a reasoning analogous to that of (B.5), we have

$$\begin{split} A_{11n} &= (2\pi)^2 \sum_{d=-J}^{J} \sum_{j=|d|}^{J} \sum_{h=1-n}^{n-1} \sum_{m=-\infty}^{\infty} c_j(h,m) c_{j-|d|}(h,m) \hat{\psi}(2\pi h/2^j) \hat{\psi}(2^{|d|}2\pi h/2^j) \\ &\qquad \times \hat{\psi}^* (2\pi m/2^j) \hat{\psi}^* (2^{|d|}2\pi m/2^j) \\ &= (2\pi)^2 \sum_{d=-J}^{J} \sum_{j=|d|}^{J} \sum_{h=1-n}^{n-1} \sum_{r=-\infty}^{\infty} \hat{\psi}(2\pi h/2^j) \hat{\psi}(2^{|d|}2\pi h/2^j) \\ &\qquad \times \hat{\psi}^* (2\pi h/2^j + 2\pi r) \hat{\psi}^* (2^{|d|}(2\pi h/2^j + 2\pi r)) \qquad \text{by (B.2)} \\ &= 2\pi \sum_{d=-J}^{J} \sum_{j=|d|}^{J} \sum_{h=1-n}^{n-1} \hat{\psi}^* (2\pi h/2^j) \hat{\psi}(2^{|d|}2\pi h/2^j) \\ &\qquad \times 2\pi \sum_{r=-\infty}^{\infty} \hat{\psi}(2\pi h/2^j + 2\pi r) \hat{\psi}^* (2^{|d|}(2\pi h/2^j + 2\pi r)) \\ &\qquad \text{by (B.12) which foll} \end{split}$$

by (B.13), which follows

$$= 2\pi \sum_{j=0}^{J} \sum_{h=1-n}^{n-1} |\hat{\psi}(2\pi h/2^{j})|^2 \qquad \text{by (B.6)}$$

= $(2^{J+1}-1)\{1+o(1)\},$ (B.12)

where the third equality follows because for any $z \in \mathbb{R}$, and any $d, r \in \mathbb{Z}$,

$$\hat{\psi}(z)\hat{\psi}(2^{|d|}z)\hat{\psi}^*(z+2\pi r)\hat{\psi}^*(2^{|d|}(z+2\pi r))$$

= $\hat{\psi}^*(z)\hat{\psi}(2^{|d|}z)\hat{\psi}(z+2\pi r)\hat{\psi}^*(2^{|d|}(z+2\pi r)),$ (B.13)

given Assumption 2(ii). Also, the last equality follows from (B.8) and $2^{3J/2}/n \rightarrow 0$. Next, using $a_J(h,m) = a_J(m,h)^*$ and (B.5), we obtain

$$A_{12n} \leq \sum_{h=-\infty}^{\infty} \sum_{|m|\geq n} |a_J(h,m)|^2 = \sum_{h=-\infty}^{\infty} \sum_{|m|\geq n} |a_J(m,h)|^2$$
$$= 2\pi \sum_{j=0}^{J} \sum_{|m|\geq n} |\hat{\psi}(2\pi m/2^j)|^2 = o(2^J),$$
(B.14)

where the last equality follows from (A.34) and $2^{3J/2}/n \rightarrow 0$. Consequently, from (B.11), (B.12), and (B.14), we obtain

$$A_{1n} = (2^{J+1} - 1)\{1 + o(1)\}.$$
(B.15)

By reasoning similar to that of A_{1n} , we can also show

$$A_{2n} = (2^{J+1} - 1)\{1 + o(1)\}.$$
(B.16)

Now, we consider A_{3n} . By reindexing, (B.2), and $\hat{\psi}^*(z) = \hat{\psi}(-z)$, we can write

$$\begin{split} A_{3n} &= (2\pi)^2 \sum_{d=-J}^{J} \sum_{j=|d|}^{J} \sum_{h=1-n}^{n-1} \sum_{m=n-1}^{n-1} c_j(h,m) c_{j-|d|}(h,-m) \hat{\psi}(2\pi h/2^j) \hat{\psi}(2^{|d|}2\pi h/2^j) \\ &\times \hat{\psi}^*(2\pi m/2^j) \hat{\psi}^*(-2^{|d|}2\pi m/2^j). \end{split}$$

Given (B.2), we have $c_j(h,m)c_{j-|d|}(h,-m) = 1$ if $m - h = 2^j r$ and $m + h = 2^{j-|d|}r'$ for some $r, r' \in \mathbb{Z}$, and $c_j(h,m)c_{j-|d|}(h,-m) = 0$ otherwise. It follows that

$$\begin{aligned} |A_{3n}| &\leq (2\pi)^2 \sum_{d=-J}^{J} \sum_{j=|d|}^{J} \sum_{r'=-\infty}^{\infty} \sum_{r=-\infty}^{\infty} |\hat{\psi}(\pi r'/2^{|d|} - \pi r)\hat{\psi}(\pi r' - 2^{|d|}\pi r)| \\ &\times |\hat{\psi}(\pi r'/2^{|d|} + \pi r)\hat{\psi}(\pi r' + 2^{|d|}\pi r)| \\ &\leq C \sum_{d=-J}^{J} \sum_{j=|d|}^{J} \left\{ \sum_{l=-\infty}^{\infty} |\hat{\psi}(\pi l)| \right\} \left\{ \sum_{r=-\infty}^{\infty} |\hat{\psi}(2\pi r)| \right\} = O\{(J+1)^2\}, \end{aligned}$$
(B.17)

where the second inequality follows by change of variable $l = r' - 2^{|d|}r$ and $|\hat{\psi}(z)| \leq C$. Similarly, we have $A_{4n} = O\{(J+1)^2\}$. Combining this with (B.10) and (B.15)–(B.17), we obtain $\sum_{h=1}^{n-1} \sum_{m=1}^{n-1} b_J^2(h,m) = 2(2^{J+1} - 1)\{1 + o(1)\}$. This completes the proof.