Multivariate Trend Filtering for Lattice Data

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Abstract

We study a multivariate version of trend filtering, called Kronecker trend filtering or KTF, for the case in which the design points form a lattice in *d* dimensions. KTF is a natural extension of univariate trend filtering (Steidl et al., 2006; Kim et al., 2009; Tibshirani, 2014), and is defined by minimizing a penalized least squares problem whose penalty term sums the absolute (higher-order) differences of the parameter to be estimated along each of the coordinate directions. The corresponding penalty operator can be written in terms of Kronecker products of univariate trend filtering penalty operators, hence the name Kronecker trend filtering. Equivalently, one can view KTF in terms of an ℓ_1 -penalized basis regression problem where the basis functions are tensor products of falling factorial functions, a piecewise polynomial (discrete spline) basis that underlies univariate trend filtering.

This paper is a unification and extension of the results in Sadhanala et al. (2016, 2017). We develop a complete set of theoretical results that describe the behavior of k^{th} order Kronecker trend filtering in d dimensions, for every $k \ge 0$ and $d \ge 1$. This reveals a number of interesting phenomena, including the dominance of KTF over linear smoothers in estimating heterogeneously smooth functions, and a phase transition at d = 2(k + 1), a boundary past which (on the high dimension-to-smoothness side) linear smoothers fail to be consistent entirely. We also leverage recent results on discrete splines from Tibshirani (2020), in particular, discrete spline interpolation results that enable us to extend the KTF estimate to any off-lattice location in constant-time (independent of the size of the lattice n).

1 Introduction

We consider a standard nonparametric regression model, relating real-valued responses $y_i \in \mathbb{R}$, i = 1, ..., n to design points $x_i \in \mathcal{X} \subseteq \mathbb{R}^d$, i = 1, ..., n,

$$y_i = f_0(x_i) + \epsilon_i, \quad i = 1, \dots, n, \tag{1}$$

where $f_0 : \mathcal{X} \to \mathbb{R}$ is the (unknown) regression function to be estimated, and ϵ_i , i = 1, ..., n are mean zero stochastic errors. In this paper, we will focus on functions f_0 that display heterogeneous smoothness across the domain \mathcal{X} , in a sense we will make precise later. We will also focus on the case in which the design points form a *d*-dimensional lattice: that is, we assume $n = N^d$, and

$$\{x_1, \dots, x_n\} = \{1/N, 2/N, \dots, 1\}^d := Z_{n,d}.$$
(2)

The assumption of a uniformly-spaced lattice, embedded in the unit cube $[0, 1]^d$, is used only for simplicity. Essentially all results (both methods and theory) translate over to the case of a somewhat more general lattice structure, a Cartesian product $\{z_{i1}\}_{i=1}^{N_1} \times \{z_{i2}\}_{i=1}^{N_2} \times \cdots \times \{z_{id}\}_{i=1}^{N_d}$, where $\prod_{j=1}^d N_j = n$, and the sets in this product are otherwise arbitrary. We return to this point in Section 9.1.

This paper is a unification and extension of Sadhanala et al. (2016, 2017) (more will be said about the relationship to these papers in Section 1.3). The models of smoothness for f_0 that we will study are based on *total variation* (TV). For a univariate function $g: [a, b] \rightarrow \mathbb{R}$, recall that its total variation is defined as

$$TV(g; [a, b]) = \sup_{a < z_1 < \dots < z_{m+1} < b} \sum_{i=1}^m |g(z_i) - g(z_{i+1})|.$$

For a multivariate function $f : [0, 1]^d \to \mathbb{R}$, we will consider notions of smoothness that revolve around the following *discrete* version of multivariate total variation:

$$TV(f; Z_{n,d}) = \sum_{j=1}^{d} \sum_{\substack{x, z \in Z_{n,d} \\ z = x + e_j/N}} |f(x) - f(z)|.$$

Here we use e_j to denote the j^{th} standard basis vector in \mathbb{R}^d , and hence the inner sum is taken over all pairs of lattice points $x, z \in \mathbb{Z}_{n,d}$ that differ in the j^{th} coordinate by 1/N (and match in all other coordinates). We will also consider higher-order versions of discrete multivariate TV, which are based on higher-order differences in the summands in the above display. We will connect our discrete notions of TV smoothness to standard continuum notions of total variation in Section 3.

Broadly speaking, there are many multivariate nonparametric regression methods available. Many of the methods in common use are *linear smoothers*: estimators of the form $\hat{f}(x) = w(x)^{\mathsf{T}}y$, for a suitable weight function $w : \mathcal{X} \to \mathbb{R}^n$ (which can depend on the design x_1, \ldots, x_n), where we use $y = (y_1, \ldots, y_n) \in \mathbb{R}^n$ for the response vector. Examples include kernel smoothing, thin-plate splines, and reproducing kernel Hilbert space estimators. A critical shortcoming of linear smoothers is that they cannot be *locally adaptive*—they cannot adapt to different local levels of smoothness exhibited by f_0 over \mathcal{X} . This is a phenomenon that has been well-documented in various settings; see Section 1.3.

The limitations of linear smoothers—and the need for nonlinear adaptive methods—is also a central theme in this paper. The central method that drives this story is a multivariate extension of trend filtering, which is indeed nonlinear and locally adaptive, a claim that will be supported by experiments and theory in the coming sections. We must note at the outset that all of the developments in this paper hinge on the assumption of lattice data (meaning, a lattice structure for the design points). A multivariate extension of trend filtering for scattered data would require a completely different approach (unlike, say, kernel smoothing or reproducing kernel Hilbert space methods, which apply regardless of the structure of the design points). The intersection of multivariate nonparametric regression methods and locally adaptive methods is actually quite small, especially when we further intersect this with the set of simple methods that are easy to use in practice, are well-understood theoretically. For this reason, we see the contributions of the current paper, though limited to lattice data, as being worthwhile. The development of new multivariate trend filtering methods for scattered data is important, and an extension is discussed in Section 9.2, but a comprehensive study is left to future work.

1.1 Review: trend filtering

Before describing the main proposal, we review *trend filtering*, a relatively recent method for univariate nonparametric regression, independently proposed by Steidl et al. (2006); Kim et al. (2009). For a univariate design, equally-spaced (say) on the unit interval, $x_i = i/n$, i = 1, ..., n, and an integer $k \ge 0$, the k^{th} order trend filtering estimate is defined by the solution of the optimization problem:

$$\underset{\theta \in \mathbb{R}^n}{\text{minimize}} \ \frac{1}{2} \| y - \theta \|_2^2 + \lambda \| D_n^{(k+1)} \theta \|_1.$$
(3)

Here $\lambda \ge 0$ denotes a tuning parameter, $y = (y_1, \dots, y_n) \in \mathbb{R}^n$ is the response vector, and $D_n^{(k+1)} \in \mathbb{R}^{(n-k-1) \times n}$ is the difference operator of order k + 1, which we will also loosely call discrete derivative operator of order k + 1. This can be defined recursively in the following manner:

$$D_n = \begin{bmatrix} -1 & 1 & 0 & \dots & 0 & 0 \\ 0 & -1 & 1 & \dots & 0 & 0 \\ \vdots & & & & \\ 0 & 0 & 0 & \dots & -1 & 1 \end{bmatrix},$$

$$D_n^{(k+1)} = D_{n-k} D_n^{(k)}, \quad k = 1, 2, 3 \dots$$
(4)

The intuition behind problem (3) is as follows. The penalty term, which penalizes the discrete derivatives of θ of order k + 1, can be equivalently seen as penalizing the *differences* in k^{th} discrete derivatives of θ at adjacent design points (due to (4)). By the sparsity-inducing property of the ℓ_1 norm, the k^{th} discrete derivatives of the trend filtering solution $\hat{\theta}$ will be exactly equal at a subset of adjacent design points, and $\hat{\theta}$ will therefore exhibit the structure of a k^{th} degree piecewise polynomial, with adaptively-chosen knots.

Here is a summary of the properties of trend filtering, as a nonparametric regression tool.¹

• The discrete trend filtering estimate, which is defined over the design points, can be "naturally" extended to a k^{th} degree piecewise polynomial function, in fact, a k^{th} degree discrete spline, on [0, 1].

¹We have defined it in this subsection for an evenly-spaced design, for simplicity, but trend filtering can still be defined for an arbitrary design, and all of the following properties still hold; see Section 9.1.

- Trend filtering is computationally efficient (several fast algorithms exist for the structured, convex problem (3)), and is not much slower to compute than (say) the smoothing spline.
- Trend filtering is more locally adaptive than the smoothing spline (or any linear smoother). This not only carries theoretical backing (next point), but is clearly noticeable in practice as well.
- Trend filtering attains the minimax rate (in squared empirical norm) of $n^{-(2k+2)/(2k+3)}$ for estimating a function f_0 whose k^{th} weak derivative is of bounded variation. The minimax linear rate (best worst-case risk attained by a linear smoother) over this class is $n^{-(2k+1)/(2k+2)}$.

Support for the above facts can be found in Tibshirani (2014), and the discrete spline (numerical analytic) perspective behind trend filtering is further developed in in Tibshirani (2020). More will be said about all of these properties in the coming sections, as analogous properties will be developed for a multivariate extension of trend filtering.

To prepare for this multivariate extension, it helps to recast the discrete problem (3) in just a slightly different form. First, some notation: for a vector $\theta \in \mathbb{R}^n$, we will (when convenient) index it by the underlying design points, and write its components as $\theta(x_i)$, i = 1, ..., n in place of θ_i , i = 1, ..., n. Next, we define a difference operator, which we will again loosely refer to as a discrete derivative operator, by

$$(\Delta\theta)(x_i) = \begin{cases} \theta(x_{i+1}) - \theta(x_i) & \text{if } i \le n-1, \\ 0 & \text{else.} \end{cases}$$

Naturally, we can view $\Delta \theta$ as a vector in \mathbb{R}^n with components $(\Delta \theta)(x_i)$, i = 1, ..., n. Higher-order discrete derivatives are obtained by repeated application of the same formula; we abbreviate $(\Delta^2 \theta)(x_i) = (\Delta(\Delta \theta))(x_i)$, and so on. In this new notation, we can now rewrite problem (3) as

$$\underset{\theta \in \mathbb{R}^n}{\text{minimize}} \ \frac{1}{2} \sum_{i=1}^n \left(y_i - \theta(x_i) \right)^2 + \lambda \sum_{i=1}^n |(\Delta^{k+1}\theta)(x_i)|.$$
(5)

1.2 Kronecker trend filtering

Let us return to the setting of multivariate design points on a lattice, as in (2) (where recall we use $N = n^{1/d}$, assumed to be integral). Building from the univariate notation and definitions at the end of the last subsection, we now introduce multivariate analogs. For a vector $\theta \in \mathbb{R}^n$, we will (when convenient) index its components by their lattice positions, denoted $\theta(x), x \in Z_{n,d}$. For each $j = 1, \ldots, d$, we define the discrete derivative of θ in the j^{th} coordinate direction at a location x by

$$(\Delta_{x_j}\theta)(x) = \begin{cases} \theta(x+e_j/N) - \theta(x) & \text{if } x, x+e_j/N \in Z_{n,d}, \\ 0 & \text{else.} \end{cases}$$

We write $\Delta_{x_j}\theta \in \mathbb{R}^n$ for the vector with components $(\Delta_{x_j}\theta)(x)$, $x \in Z_{n,d}$. As before, higher-order discrete derivatives are simply defined by repeated application of the above definition; we use abbreviations $(\Delta_{x_j^2}\theta)(x) = (\Delta_{x_j}(\Delta_{x_j}\theta))(x)$, $(\Delta_{x_j,x_\ell}\theta)(x) = (\Delta_{x_j}(\Delta_{x_\ell}\theta))(x)$, and so on.

With this notation in place, we define a multivariate version of trend filtering, that we call *Kronecker trend filtering* (KTF). Given an integer $k \ge 0$, the k^{th} order KTF estimate is defined by the solution of the optimization problem:

$$\underset{\theta \in \mathbb{R}^n}{\text{minimize}} \ \frac{1}{2} \sum_{i=1}^n \left(y_i - \theta(x_i) \right)^2 + \lambda \sum_{j=1}^d \sum_{x \in Z_{n,d}} |(\Delta_{x_j^{k+1}}\theta)(x)|.$$
(6)

Note the close analogy between (5) and (6): the latter extends the former by adding up absolute discrete derivatives of θ of order k + 1 along each one of the *d* coordinate directions. A similar intuition carries over from the univariate case, regarding the role of the penalty in (6), and the structure of the solution. As we can see, the KTF problem penalizes the differences in k^{th} discrete derivatives of θ at lattice positions x and z, for all x and z that are adjacent along any one of the *d* coordinate directions. By the sparsifying nature of the ℓ_1 norm, the KTF solution $\hat{\theta}$ will have equal k^{th} discrete derivatives between neighboring points on the lattice (and more so for larger λ , generally speaking). Hence, along any line segment parallel to one of the coordinate axes, the KTF solution $\hat{\theta}$ will have the structure of a k^{th} degree piecewise polynomial, with adaptively-chosen knots. This intuition will be made rigorous in Section 2.3.

We can also rewrite the KTF problem (6) in a more compact form, so that it resembles (3):

$$\underset{\theta \in \mathbb{R}^{n}}{\text{minimize}} \ \frac{1}{2} \| y - \theta \|_{2}^{2} + \lambda \| D_{n,d}^{(k+1)} \theta \|_{1}, \tag{7}$$

where we define

$$D_{n,d}^{(k+1)} = \begin{bmatrix} D_N^{(k+1)} \otimes I_N \otimes \dots \otimes I_N \\ I_N \otimes D_N^{(k+1)} \otimes \dots \otimes I_N \\ \vdots \\ I_N \otimes I_N \otimes \dots \otimes D_N^{(k+1)} \end{bmatrix}.$$
(8)

Here $D_N^{(k+1)} \in \mathbb{R}^{(N-k-1)\times N}$ is the discrete derivative matrix from (4) (as would be used in k^{th} order univariate trend filtering on N points); $I_N \in \mathbb{R}^{N \times N}$ denotes the identity matrix; and $A \otimes B$ denotes the Kronecker product of matrices A, B. Each block of rows in (8) is made up of a total of d-1 Kronecker products (a total of d matrices). The Kronecker product structure behind the penalty matrix in (8) is what inspires the name Kronecker trend filtering. In a similar vein, we will also refer to $\|D_{n,d}^{(k+1)}\theta\|_1$ as the k^{th} order *Kronecker total variation* (KTV) of θ . Note that when k = 0, the KTF problem (7) reduces to anisotropic TV denoising on the d-dimensional lattice, and when d = 1, it reduces to k^{th} order univariate trend filtering (3).

Before we delve any deeper into its properties, which will start in Section 2, we give a simple example of KTF in Figure 1. The example portrays an underlying function f_0 in d = 2 dimensions that has two peaks in opposite corners of the unit square $[0, 1]^2$, one larger and one smaller, and is otherwise very smooth. Based on noisy observations of f_0 over a 2d lattice, the estimates from KTF of orders k = 0, 1, 2 are able to capture this behavior. Meanwhile, estimates from kernel smoothing are not, and they either oversmooth the larger peak or undersmooth the valleys, depending on the choice of the bandwidth. Advocates of kernel smoothing might say that this problem can be solved by giving the kernel a locally-varying bandwidth (that is, modeling the bandwidth itself as a function of $x \in [0, 1]^d$). While this can work in principle, for example, using Lepski's method (see Section 1.3), it can often be hard to implement in practice. More to the point: the comparison in Figure 1 is not meant to portray the kernel smoothing framework as being wholly untenable; rather, it is just meant to portray the differences in adaptivity of KTF versus kernel smoothing when each method is allowed only a single tuning parameter.

1.3 Related work

Our paper builds off a line of work on trend filtering, which, recall, enforces smoothness by penalizing the ℓ_1 norm of discrete derivatives of a given order (Steidl et al., 2006; Kim et al., 2009; Tibshirani, 2014; Wang et al., 2014; Ramdas and Tibshirani, 2016). This can be seen as a discrete analog of the *locally adaptive regression spline* estimator, which penalizes the TV of a given order derivative of a function (Koenker et al., 1994; Mammen and van de Geer, 1997). For an overview of trend filtering, its connection to classical theory on splines and divided differences, and related results that bridge discrete and continuum representations (such as that connecting trend filtering and locally adaptive regression splines), we refer to Tibshirani (2020). Trend filtering was extended to general graphs by Wang et al. (2016) (more on this below), and was studied in an additive model setting by Sadhanala and Tibshirani (2019).

Kronecker trend filtering was proposed in Sadhanala et al. (2017). Minimax theory for KTF has been developed in different special cases, distributed among several papers. Minimax results for TV denoising on lattices can be found in Hutter and Rigollet (2016) (upper bounds) and Sadhanala et al. (2016) (lower bounds); with respect to KTF and KTV classes, this covers the case of k = 0 and all dimensions d. Meanwhile, Sadhanala et al. (2017) derived minimax results (matching upper and lower bounds) for all k and d = 2. The current paper completes the landscape, deriving minimax theory for all smoothness orders k and all dimensions d. The majority of this paper (including the minimax theory) was completed in 2018 and can be found in the Ph.D. thesis of the first author (Sadhanala, 2019).

Graph-based TV methods. Our paper is complementary to the line of research on locally adaptive nonparametric estimation over graphs, such as Gavish et al. (2010); Sharpnack et al. (2013); Wang et al. (2016); Göbel et al. (2018); Padilla et al. (2018, 2020); Ye and Padilla (2021). The lattice structure that we consider in this paper can be cast as a particular *d*-dimensional grid graph with *n* nodes (that is, with all side lengths equal to $N = n^{1/d}$). However, many of the methods proposed in the aforementioned references seek to be far more general and operate over arbitrary graph structures. This generality comes with several challenges, from conceptual to theoretical.



Figure 1: Top left: underlying regression function f_0 evaluated over a square lattice with $n = 40^2 = 1600$ points, and associated responses (formed by adding noise) shown as black points. Top middle and top right: kernel smoothing (with a spherical Gaussian kernel) fit to this data using large and small bandwidth values, respectively. Bottom left, middle, and right: Kronecker trend filtering estimates of orders k = 0, 1, 2, respectively (recall, KTF with k = 0 reduces to anisotropic total variation denoising). We see that, in order to capture the larger of the two peaks in f_0 , kernel smoothing must significantly undersmooth the other peak (and surrounding areas); instead, with more regularization, it undersmooths throughout. The KTF estimates are able to adapt to heterogeneity in the smoothness of f_0 . Also, each exhibits a distinct structure based on the polynomial order k.

For example, Wang et al. (2016) developed graph trend filtering (GTF), which estimates a signal that takes values over the nodes of an arbitrary graph by penalizing the ℓ_1 norm its graph derivatives, which are defined via an iterated graph Laplacian operator. Applied to the *d*-dimensional grid graph, and translated into the notation of our paper, the penalty term used in k^{th} order GTF for a signal θ at a point in the lattice $x \in Z_{n,d}$ is:

$$\begin{cases} \sum_{j_1=1}^d \left| \sum_{j_2,\dots,j_q=1}^d \left(\Delta_{x_{j_1},x_{j_2}^2,\dots,x_{j_q}^2} \theta \right)(x) \right| & \text{for } k \text{ even, where } q = k/2, \\ \left| \sum_{j_1,\dots,j_q=1}^d \left(\Delta_{x_{j_1}^2,x_{j_2}^2,\dots,x_{j_q}^2} \theta \right)(x) \right| & \text{for } k \text{ odd, where } q = (k+1)/2 \end{cases}$$

Compared to the analogous penalty term used in KTF (6), which is just $\sum_{j=1}^{d} |(\Delta_{x_{j}^{k+1}}\theta)(x)|$, we see that the above is much is harder to interpret. GTF clearly considers some form of mixed derivatives (whereas KTF does not and is more anisotropic), but it is generally conceptually unclear what kind of smoothness GTF is promoting. Further, Sadhanala et al. (2017) show that using the GTF penalty operator to define a smoothness class for the analysis of multivariate signals is problematic, in the following sense: for any $k \ge 1$ and any dimension d, there are k^{th} order Holder smooth functions whose discretization to the lattice is arbitrarily nonsmooth as measured by the k^{th} order GTF penalty operator. This is due to issues in the way the GTF penalty operator measures smoothness on the boundaries of the lattice.

Continuous-time multivariate TV methods. There is a rich body of work in applied mathematics on the denoising of signals or images by promoting total variation smoothness, beginning with the seminal paper by Rudin et al. (1992), which gave rise to the so-called Rudin-Osher-Fatemi (ROF) functional. This was then further developed by Rudin and Osher (1994); Vogel and Oman (1996); Chambolle and Lions (1997); Chan et al. (2000), and others. Penalized versions of the ROF functional have also been combined with multiscale (Candès and Guo, 2002; Dong et al., 2011) and L^1 (Chan and Esedoglu, 2005) data-fidelity terms. Papers in this line of work tend to be cast in continuous-time, which means that the estimand, estimator, and typically even the data itself are each functions of one or more variables on a continuous domain (such as $[0, 1]^d$). A related line of work, inspired by ROF, considers discretization in its numerical analysis, see, for example, Chambolle (2004, 2005); Almansa et al. (2008).

More recently, del Álamo et al. (2021) studied estimation of a multivariate function of bounded variation under the white noise model, in arbitrary dimension d. This may be interpreted as a continuum analog of our setting, albeit for k = 0. They derive minimax rates under the L^p norm, for any $p \ge 1$. When p = 2 (matching our study), their estimator obtains (up to log factors), the minimax rate of $n^{-1/d}$ on the squared L^2 error scale, for any $d \ge 2$, which agrees with the minimax rate in our discrete setting. We note that our methods, while motivated from discrete principles, do in fact bear rigorous connections to continuous-time formulations of multivariate total variation; see Sections 2.3 and 3.

Other models for multivariate TV. There has been a recent surge of work studying different generalizations of TV and trend filtering penalties to multiple dimensions, for example, Bibaut and van der Laan (2019); Fang et al. (2021); Ortelli and van de Geer (2021); Chatterjee and Goswami (2021); Ki et al. (2021). Many of these works are based on the notion of Hardy-Krause variation of a multivariate function, or the related notion of Vitali variation. For a function to be smooth in the Hardy-Krause sense, it must exihibit an order of smoothness k that scales with the dimension d. This leads to estimation error rates that are (nearly) dimension-free. However, practically speaking, expecting the inherent smoothness of the regression function to be on par with the ambient dimension may not be a reasonable assumption in many applications. A distinctive feature of our work is that the smoothness order k (which translates into the max degree of the fitted local polynomial) is a user-defined parameter, and is not tied in any way to the dimension d.

Locally-adaptive kernel smoothing. Lepski's method, which originated in the seminal paper by Lepskii (1991), is a procedure for selecting a local bandwidth in kernel smoothing; roughly speaking, at each point x in the domain, it chooses the largest bandwidth (from a discrete set of possible values) such that the kernel estimate at x is within a carefully-defined error tolerance to estimates at smaller bandwidths. Since its introduction, many papers have studied and generalized Lepski's method, see, for example, Lepskii (1992, 1993); Lepski et al. (1997); Lepski and Spokoiny (1997); Kerkyacharian et al. (2001, 2008); Goldenshluger and Lepski (2008, 2009, 2011, 2013); Lepski (2015). Papers in this line of work are focused on establishing novel theoretical guarantees—for example, using kernel smoothing as a base estimator to achieve minimax rates over heterogeneous function classes, and not on applied considerations—like the computational difficulties or ambiguities (choice of constants) encountered in practical implementation.

Most related to the current paper is Kerkyacharian et al. (2001, 2008), which considered estimation in anisotropic Besov classes using Lepski's method, under the white noise model. These papers considered the "dense" and "sparse" cases, which roughly correspond to the cases s > 1/2 and $s \le 1/2$ in our theory, respectively (see Figure 2). This is revisited in Remark 12.

Tensor product and hyperbolic wavelets. Wavelets have a rich history in signal processing, approximation theory, and other disciplines; classic references include Daubechies (1992); Chui et al. (1992a); Meyer and Roques (1993); Mallat (2009). Seminal work by Donoho and Johnstone (1998), on minimax estimation over univariate Besov spaces using wavelet-based estimators, contributed greatly to the popularity of wavelets in statistics. The earliest work on multivariate wavelet approximation appears to have been Meyer (1987, 1990) and Mallat (1989b,a), which considered "separable" wavelet bases, formed from tensor products of univariate wavelet basis functions within each resolution level. A second approach, well-studied in approximation theory but less so in statistics, considers constructing "truly multivariate" wavelet bases by generating multiresolution subspaces in the ambient domain, from which a wavelet basis can be derived (Meyer, 1990; Riemenschneider and Shen, 1992; Chui et al., 1992b; Lorentz and Madych, 1992; DeVore and Lucier, 1992). This gives rise to nonseparable wavelet bases, but in both cases, the support of the wavelet function has the same scale in each coordinate direction. The desire to effectively represent functions with different degrees of smoothness in different coordinate directions led to the development of hyperbolic wavelets, whose basis functions are formed by taking tensor products of univariate wavelet basis functions *across* resolution levels (Neumann

and von Sachs, 1997; DeVore et al., 1998; Neumann, 2000). Of particular relevance to our work is the latter paper, and connections will be drawn in Remark 12.

1.4 Summary and outline

A summary of results in this paper and outline for this paper is given below.

- In Section 2, we derive some basic properties of KTF, including an equivalent continuous-time formulation for problem (7), which provides insights into the local structure of KTF estimates.
- In Section 3, we derive an expression for higher-order multivariate TV (in the standard measure-theoretic sense) in terms of an integrating univariate TV on line segments running parallel to the coordinate axes. We use this to motivate the definition of KTV smoothness, the central notion of smoothness used in this paper—it is simply a discretization of the aforementioned integral.
- In Section 4, we introduce the smoothness classes of interest for our study of minimax theory, and examine the relationships between them.
- In Section 5, we derive a complete set of results on the minimax estimation risk, as measured in the squared l₂ norm, over the set T^k_{n,d}(C_n) of vectors θ with kth order KTV smoothness satisfying ||D^(k+1)_{n,d}θ||₁ ≤ C_n, for a given sequence C_n > 0. We prove that KTF is minimax rate optimal (up to log factors) for any k, d, and derive lower bounds on the minimax linear risk (that is, the best worst-case risk over all linear smoothers) which show that linear estimators are suboptimal for any k, d. Interestingly, the minimax rates reveal a phase transition at 2(k + 1) = d, and in the low smoothness-to-dimension regime, linear smoothers fail to be consistent altogether. See Figure 2 for a more detailed summary.
- In Section 6, we describe and compare specialized convex optimization algorithms that can be used to compute the KTF solution in (7).
- In Section 7, we present an extremely efficient and simple algorithm for interpolating the discrete KTF estimate θ (the solution in (7)), defined over the lattice, into a function \hat{f} (the solution in (12)), defined over the underlying continuum domain $[0, 1]^d$. Remarkably, this interpolation method, which builds off discrete spline interpolation results from Tibshirani (2020), runs in *constant-time* (independent of n).
- In Section 8, we carry out empirical experiments that compare KTF and various other estimators, and examine whether the empirical error rates match the minimax theory derived in Section 5.
- In Section 9, we conclude with a discussion, and cover some extensions and directions for future work.

2 Basic properties

In this section, we cover a number of basis properties that reflect the structure and complexity of KTF estimates.

2.1 Unpenalized component

We start by examining the null space of the KTF penalty matrix in (8). A word on notation here, and in general: when convenient, we will use x_j to denote the j^{th} component of a vector x, which should not be confused with our use of x_i to denote the i^{th} design point (itself a *d*-dimensional vector). This is an unfortunate clash of notation, however, the meaning should always be clear form the context. (Furthermore, we will keep the use of indices i, j for the two cases consistent throughout to aid the interpretation—hence x_j will always be univariate, the j^{th} component of a vector x, and x_i will always be *d*-dimensional, the i^{th} design point.)

Proposition 1. The null space of the KTF penalty matrix in (8) has dimension $(k + 1)^d$. Furthermore, it is spanned by a polynomial basis made up of elements

$$p(x) = x_1^{a_1} x_2^{a_2} \cdots x_d^{a_d}, \quad x \in Z_{n,d},$$

for all $a_1, \ldots, a_d \in \{0, \ldots, k\}$.



Figure 2: Summary of the minimax results developed in this paper. The central object of our study is the set $\mathcal{T}_{n,d}^k(C_n^*)$ of vectors θ defined over the d-dimensional lattice $Z_{n,d}$, with k^{th} order KTV smoothness satisfying $\|D_{n,d}^{(k+1)}\theta\|_1 \leq C_n^*$, for a sequence $C_n^* > 0$ obeying what we call the canonical scaling, to be made precise later. The following two statements hold, generally (regardless of k, d):

- 1. KTF achieves the minimax rate (up to log factors) over $\mathcal{T}_{n,d}^k(C_n^*)$; and
- 2. no linear smoother is able to achieve the minimax rate over this class.

However, the story is more interesting, due to a phase transition occurring at 2(k+1) = d. Defining a notion of effective smoothness by s = (k+1)/d, this can be explained as follows. When s > 1/2, the minimax rate has the more classical form $n^{-2s/(2s+1)}$, matching the minimax rate for a k^{th} Holder class in dimension d (or an s^{th} order Holder class in the univariate case). Indeed, the lower bound on the minimax rate that we derive is given by embedding a Holder class into $\mathcal{T}_{n,d}^k(C_n^*)$. Meanwhile, the minimax linear risk (the best worst-case risk among linear smoothers) scales as $n^{-(2s-1)/(2s)}$, which can be interpreted as the rate of KTF (or any other minimax optimal method) for a problem with a half less degree of effective smoothness. When $s \leq 1/2$, the minimax rate takes on the less classical form n^{-s} , and the lower bound is obtained by embedding a suitable ℓ_1 ball into $\mathcal{T}_{n,d}^k(C_n^*)$. Further, the gap between the minimax linear and nonlinear rates is even more dramatic: the minimax linear rate is constant, which means no linear smoother is even consistent over $\mathcal{T}_{n,d}^k(C_n^*)$ (in the sense of worst-case risk). Finally, though not reflected in the figure, we note that when s < 1/2 the KTV class and its embedded Holder class exhibit different minimax rates, n^{-s} versus $n^{-2s/(2s+1)}$, respectively. Whether KTF can adapt to the latter (faster) Holder rate in the low smoothness-to-dimension regime, s < 1/2, is an open question.

The proof is elementary and is deferred to Appendix A (which contains all proofs in this paper). This proposition reveals that the KTF penalty matrix has quite a rich null space, thus KTF lets a significant component of the response vector y "pass through" unpenalized. In contrast to univariate trend filtering, which preserves univariate polynomials of degree k (precisely, it preserves the projection of y onto this subspace), KTF preserves "much more" than multivariate polynomials of degree k: it preserves multivariate polynomials of max degree k.² To give an example, when k = 1 and d = 2, the KTF estimator—which we might be tempted to call "linear-order" KTF (to use an analogous term as we do in univariate trend filtering)—preserves any polynomial of the form $p(x) = ax_1 + bx_2 + cx_1x_2$. This is of course not a linear function, but a bilinear one (due to the cross-product term x_1x_2).

²By a multivariate polynomial of degree k, we mean (adhering to the standard classification) a sum of terms of the form $b \prod_{j=1}^{d} x_{j}^{a_{j}}$, where the sum of degrees satisfies $\sum_{j=1}^{d} a_{j} \leq k$. By a multivariate polynomial of max degree k, we mean the same, but where the degrees satisfy $a_{j} \leq k$, $j = 1, \ldots, d$.

2.2 Review: trend filtering in continuous-time

For univariate trend filtering (3), an equivalent continuous-time formulation was derived in Tibshirani (2014):

$$\underset{f \in \mathcal{H}_{n}^{k}}{\text{minimize}} \quad \frac{1}{2} \sum_{i=1}^{n} \left(y_{i} - f(x_{i}) \right)^{2} + \lambda \operatorname{TV}(f^{(k)}).$$
(9)

Problems (3), (9) are equivalent in the sense that their solutions $\hat{\theta}$, \hat{f} , respectively, satisfy $\hat{\theta}_i = \hat{f}(x_i)$, i = 1, ..., n. In (9), we use $f^{(k)}$ to denote the k^{th} weak derivative of f, and $\text{TV}(\cdot) = \text{TV}(\cdot; [a, b])$ to denote the total variation operator defined with respect to any interval [a, b] containing the design points (henceforth, when convenient, we will drop the underlying domain from our notation for univariate TV). The optimization in (9) is performed over all functions f that lie in a space \mathcal{H}_n^k , which is the span of $h_{n,j}^k$, j = 1, ..., n, the following k^{th} degree piecewise polynomials:

$$h_{n,i}^{k}(x) = \frac{1}{(i-1)!} \prod_{j=1}^{i-1} (x-j/n), \quad i = 1, \dots, k+1,$$

$$h_{n,i}^{k}(x) = \frac{1}{k!} \prod_{j=i-k}^{i-1} (x-j/n) \cdot 1\{x > (i-1)/n\}, \quad i = k+2, \dots, n.$$
(10)

(Here, for convenience, we interpret the empty product to be equal to 1.) The functions in (10) are called the k^{th} degree *falling factorial basis*. Observe that they depend on the *n* underlying design points $1/n, 2/n, \ldots, 1$, which we express through the first subscript *n* in $h_{n,i}^k$. Note also the similarity between the above basis and the standard truncated power basis for splines; in fact, when k = 0 or k = 1, the two bases are equal and \mathcal{H}_n^k is just a space of splines with knots at the design points. However, when $k \ge 2$, this is no longer true—the falling factorial functions are k^{th} degree piecewise polynomials with (mildly) discontinuous derivatives of all orders $1, \ldots, k - 1$, and therefore they span a different space than that of k^{th} degree splines—they space the space of k^{th} degree *discrete splines*, which are piecewise polynomials that have continuous discrete derivatives (rather than derivatives) at their knot points. See Tibshirani (2020).

To be clear, the original formulation (3) is more computationally convenient (it is a structured convex problem for which several fast algorithms exist, discussed in Section 1.3). But the variational formulation (9) is important because it provides rigorous backing to the intuition that a k^{th} order trend filtering estimate exhibits the structure of a k^{th} degree piecewise polynomial, with adaptively-chosen knots (a feature of the ℓ_1 penalty in (3) or TV penalty in (9)); moreover, it shows how to extend the trend filtering estimate from a discrete sequence, defined over the design points, to a function on the continuum interval [0, 1].

2.3 Continuous-time

We now develop a similar continuous-time representation for KTF.

Proposition 2. Let $h_{N,i}^k : [0,1] \to \mathbb{R}$, i = 1, ..., N denote the k^{th} degree falling factorial functions (10) with respect to design points 1/N, 2/N, ..., 1, and $\mathcal{H}_{n,d}^k$ denote the space spanned by all d-wise tensor products of these functions. That is, abbreviating $h_i = h_{N,i}^k$, i = 1, ..., N, this is the space of all functions $f : [0,1]^d \to \mathbb{R}$ of the form

$$f(x) = \sum_{i_1,\dots,i_d=1}^N \alpha_{i_1,\dots,i_d} h_{i_1}(x_1) h_{i_2}(x_2) \cdots h_{i_d}(x_d), \quad x \in [0,1]^d,$$
(11)

for coefficients $\alpha \in \mathbb{R}^n$ (whose components we denote as α_{i_1,\ldots,i_d} , $i_1,\ldots,i_d \in \{1,\ldots,N\}$). Then the KTF estimator defined in (7) is equivalent to the optimization problem:

$$\underset{f \in \mathcal{H}_{n,d}^k}{\text{minimize}} \ \frac{1}{2} \sum_{i=1}^n \left(y_i - f(x_i) \right)^2 + \lambda \sum_{j=1}^d \sum_{x_{-j}} \text{TV}\left(\frac{\partial^k f(\cdot, x_{-j})}{\partial x_j^k} \right), \tag{12}$$

where $f(\cdot, x_{-j})$ denotes f as function of the j^{th} coordinate with all other dimensions fixed at x_{-j} , $(\partial^k/\partial x_j^k)(\cdot)$ denotes the k^{th} partial weak derivative operator with respect to x_j , and inner sum in the second term in (12) is interpreted as a sum over $Z_{m,d-1}$, where $m = N^{d-1}$ (that is, a sum over the (d-1)-dimensional uniformly-spaced lattice with N^{d-1} total points). The discrete (7) and continuous-time (12) problems are equivalent in the sense that at their solutions $\hat{\theta}, \hat{f}$, respectively, we have: $\hat{\theta}_i = \hat{f}(x_i), i = 1, ..., n$. **Remark 1.** Similar to the discrete-time terminology, we will refer to $\sum_{j=1}^{d} \sum_{x-j} \text{TV}(\partial^k f(\cdot, x_{-j})/\partial x_j^k)$, the penalty functional in the continuous-time representation (12) of the KTF problem, as the k^{th} order *Kronecker total variation* (KTV) of f. A key implication of Proposition 2 is that the solution \hat{f} in (12) not only interpolates the solution $\hat{\theta}$ in (7), but it is exactly as smooth in continuous-time as $\hat{\theta}$ is in discrete-time, as measured by KTV:

$$\|D_{n,d}^{(k+1)}\hat{\theta}\|_1 = \sum_{j=1}^d \sum_{x_{-j}} \operatorname{TV}\left(\frac{\partial^k \hat{f}(\cdot, x_{-j})}{\partial x_j^k}\right).$$

How to form the interpolant \hat{f} is discussed briefly in the remark after the next, and covered in detail in Section 7.

Remark 2. From (11), the basis underlying the continuous-time representation (12) of the KTF optimization problem, we can see that a k^{th} order KTF estimate exhibits the structure of a tensor product of k^{th} degree discrete splines, with adaptively knots, chosen to promote higher-order TV smoothness along the coordinate axes. In other words, locally, it exhibits the structure of a multivariate polynomial of max degree k. When k = 1 and d = 2, for example, the structure is locally of the form $\hat{f}(x) = ax_1 + bx_2 + cx_1x_2$, which is a bilinear function and has local curvature. Such curvature is somewhat visible in Figure 1 (bottom row, middle panel), and it will be even more apparent later when we discuss interpolation, see Figure 6.

Remark 3. The proof of Proposition 2 reveals that (12) has an equivalent form, transcribed here for convenience:

$$\underset{\alpha \in \mathbb{R}^{n}}{\operatorname{minimize}} \quad \frac{1}{2} \left\| y - \left(H_{N}^{(k+1)} \otimes \cdots \otimes H_{N}^{(k+1)} \right) \alpha \right\|_{2}^{2} + \lambda k! \left\| \begin{bmatrix} I_{N}^{0} \otimes H_{N}^{(k+1)} \otimes \cdots \otimes H_{N}^{(k+1)} \\ H_{N}^{(k+1)} \otimes I_{N}^{0} \otimes \cdots \otimes H_{N}^{(k+1)} \\ \vdots \\ H_{N}^{(k+1)} \otimes H_{N}^{(k+1)} \otimes \cdots \otimes I_{N}^{0} \end{bmatrix} \alpha \right\|_{1}, \quad (13)$$

where $H_N^{(k+1)} \in \mathbb{R}^{N \times N}$ is the falling factorial basis matrix (with columns given by evaluations of the falling factorial functions at the design points) and $I_N^0 \in \mathbb{R}^{(N-k-1) \times N}$ denotes the last N-k-1 rows of the identity I_N . Interestingly, the penalty in (13) is not a pure ℓ_1 penalty on the coefficients α (as it would be in basis form in the univariate case) but an ℓ_1 penalty on aggregated (positive linear combinations of) coefficients.

The basis formulation in (13) gives us a natural recipe for how to extend a KTF estimate from a discrete sequence, defined over the lattice points, to a function on the hypercube $[0, 1]^d$. This is simply:

$$\hat{f}(x) = \sum_{i_1,\dots,i_d=1}^{N} \hat{\alpha}_{i_1,\dots,i_d} h_{i_1}(x_1) h_{i_2}(x_2) \cdots h_{i_d}(x_d), \quad x \in [0,1]^d,$$
(14)

where $\hat{\alpha}$ is the solution in (13). Though it is not obvious from the basis expansion in (14), it turns out that at any point $x \in [0,1]^d$, we can form the prediction $\hat{f}(x)$ in constant-time, starting from the fitted values $\hat{\theta}_i = \hat{f}(x_i), i = 1, ..., n$. This leverages recent advances in discrete spline interpolation from Tibshirani (2020), and is covered in Section 7.

2.4 Degrees of freedom

Given data from a model (1), where the errors ϵ_i , i = 1, ..., n are i.i.d. with mean zero and variance σ^2 , recall that the *degrees of freedom* of an estimator $\hat{\theta}_i = \hat{f}(x_i)$, i = 1, ..., n of the means $\theta_{0i} = f_0(x_i)$, i = 1, ..., n is a quantitative reflection of its complexity, defined as (Efron, 1986; Hastie and Tibshirani, 1990):

$$df(\hat{\theta}) = \frac{1}{\sigma^2} \sum_{i=1}^{n} Cov(y_i, \hat{\theta}_i).$$

When the errors are Gaussian, Tibshirani and Taylor (2011, 2012) derived an expression for the degrees of freedom of any generalized lasso estimator, based on Stein's formula (Stein, 1981). This covers KTF in (7) as a special case, and thus translates into the following result for our setting: if $\epsilon_i \sim N(0, \sigma^2)$, i = 1, ..., n in (1), and $\hat{\theta}$ denotes the solution in (7) with active set

$$A = \operatorname{supp}\left(D_{n,d}^{(k+1)}\hat{\theta}\right) = \left\{i : \left[D_{n,d}^{(k+1)}\theta\right]_i \neq 0\right\},\$$

then

$$df(\hat{\theta}) = \mathbb{E}\Big[nullity\Big([D_{n,d}^{(k+1)}]_{-A}\Big)\Big],\tag{15}$$

where nullity (M) denotes the nullity (dimension of the null space) of a matrix M, and M_{-S} denotes the submatrix of M given by removing all rows indexed by a set S. From the above, we of course have the natural estimator of degrees of freedom

$$\widehat{\mathrm{df}}(\widehat{\theta}) = \mathrm{nullity}\Big(\big[D_{n,d}^{(k+1)}\big]_{-A}\Big),\tag{16}$$

which is unbiased for (15).

The expression in (16) is easy to interpret when k = 0: in this case, it reduces to the number of connected constant pieces in the KTF solution $\hat{\theta}$, where connectivity is interpreted with respect to the underlying *d*-dimensional grid graph. This follows from the fact that the penalty matrix $D_{n,d}^{(1)}$ in this case is the edge incidence operator on the grid graph, and any submatrix of this penalty matrix (defined by removing a subset of rows) is itself the edge incidence operator with respect to a subgraph of the original grid (induced by removing a subset of the edges). This result and its interpretation was already given in Tibshirani and Taylor (2011, 2012) in the context of TV denoising on a graph.

When $k \ge 1$, the unbiased estimator of degrees of freedom in (16) is not as easy to interpret. This is because, at a high level, there is no longer a clear link between the local structure exhibited by $\hat{\theta}$ and whether or not a particular entry of $D_{n,d}^{(k+1)}\hat{\theta}$ is nonzero. However, we show in Appendix D that it is possible to compute the right-hand side in (16) with a simple, direct algorithm that runs in linear time (more precisely, the algorithm requires O(ndk) operations).

3 Interlude: total variation on lines

In this section, we take a continuum perspective, looking at total variation defined over functions in \mathbb{R}^d , and connect it to the discrete notion of total variation used in the previous sections used to define the KTF estimator.

3.1 Measure-theoretic total variation

Let U be an open, bounded subset of \mathbb{R}^d and $L^p(U)$ denote the space of real-valued functions on U with finite L^p norm, $\int_U |f(x)|^p dx < \infty$. A function $f \in L^1(U)$ is said to be of *bounded variation* (BV) provided $\mathrm{TV}(f; U) < \infty$, where

$$\operatorname{TV}(f;U) = \sup\left\{\int_{U} f(x)\operatorname{div}\phi(x)\,dx: \phi \in C_{c}^{\infty}(U;\mathbb{R}^{d}), \ \|\phi(x)\|_{\infty} \le 1 \text{ for all } x \in U\right\}.$$
(17)

Above, $C_c^{\infty}(U; \mathbb{R}^d)$ denotes the space of infinitely continuously differentiable functions from U to \mathbb{R}^d with compact support, and div(·) denotes the divergence operator, div $\phi = \sum_{j=1}^d \partial \phi_j / \partial x_j$. We call $\mathrm{TV}(f; U)$ the *total variation* of f; this is the standard measure-theoretic definition used in modern analysis; see, for example, Chapter 5 of Evans and Gariepy (2015). To be clear, it would be more precise to call our definition in (17) the *aniostropic* total variation of f (due to the use of the ℓ_{∞} norm in the constraint in (17) on the test function ϕ), but we often drop the reference to the anisotropic qualifier for simplicity.

To build intuition, we note that if $f \in W^{1,1}(U)$, that is, f is in $L^1(U)$ and it is weakly differentiable and its weak derivative ∇f is also in $L^1(U)$, then

$$\mathrm{TV}(f;U) = \int_U \|\nabla f(x)\|_1 \, dx. \tag{18}$$

Writing BV(U) for the space of bounded variation functions, the above shows that $W^{1,1}(U) \subseteq BV(U)$. Importantly, this is a strict inclusion, because, for example, the indicator function of a set that has smooth boundary is of bounded variation, but it is not in $W^{1,1}(U)$ (it is not weakly differentiable).

3.2 Univariate total variation revisited

In order to connect the discrete notions of TV that we use in this paper to the standard measure-theoretic definition of TV defined in (17), we must first refine our definition of univariate TV. For a function $g : [a, b] \to \mathbb{R}$, we define its total variation as:

$$TV(g;[a,b]) = \sup_{\substack{a < z_1 < \dots < z_{m+1} < b \\ z_1,\dots,z_{m+1} \in AC(g)}} \sum_{i=1}^m |g(z_i) - g(z_{i+1})|,$$
(19)

where the supremum is only taken over the set AC(g) of points of approximate continuity of g. Approximate continuity is a weak notion of continuity that excludes, for example, point discontinuities; see Section 1.7.2 of Evans and Gariepy (2015). Observe that the definition in (19) differs from that given in the introduction in that the latter does not require that the supremum be taken over points of approximate continuity. Some authors, including Evans and Gariepy (2015), differentiate these definitions by calling the latter the *variation* of g and (19) the *essential variation* of g. An intuitive way of interpreting their connection is as follows: the essential variation of g is the infimum of the variation achievable by any function \tilde{g} that agrees with g Lebesgue almost everywhere.

The reason the refinement in (19) is important, when using the measure-theoretic definition in (17) as a basis for defining the BV space, is that BV functions (as with L^p functions and Sobolev functions) are only well-defined up to a set of Lebesgue measure zero. That is, if f and \tilde{f} agree Lebesgue almost everywhere, then their TV as defined in (17) (as with L^p norms or Sobolev norms) must also agree. Therefore, it should be clear that (19) is the proper univariate notion here, as otherwise redefining g at a point would change its univariate TV (without restricting the supremum to points of approximate continuity). Lastly, and reassuringly, the multivariate measure-theoretic definition in (17) reduces to the univariate definition in (19) once we take U = (a, b) (see Theorem 5.21 in Evans and Gariepy (2015)).

3.3 Total variation on lines

We now proceed in the opposite direction to the end of the last subsection: instead of reducing the multivariate definition to the univariate case, we will use the univariate definition of TV to approach the multivariate one. Interestingly, as we will see next, it turns out that (19) can be used as a building block for (17) for an open, bounded set $U \subseteq \mathbb{R}^d$. In words, the next result says that the multivariate notion of TV on U is given by aggregating the univariate notion on all line segments parallel to the coordinate axes, anchored at boundary points of U.

Theorem 1. Let $U \subseteq \mathbb{R}^d$ be an open, bounded, convex set. Then for any $f \in BV(U)$,

$$TV(f;U) = \sum_{j=1}^{d} \int_{U_{-j}} TV(f(\cdot, x_{-j}); I_{x_{-j}}) dx_{-j},$$
(20)

where for each j = 1, ..., d, we define $U_{-j} = \{x_{-j} : (x_j, x_{-j}) \in U \text{ for some } x_j\}$, and $I_{x_{-j}} = [a_{x_{-j}}, b_{x_{-j}}]$, with

$$a_{x_{-j}} = \inf\{x_j : (x_j, x_{-j}) \in U\},\b_{x_{-j}} = \sup\{x_j : (x_j, x_{-j}) \in U\}.$$

Recall $f(\cdot, x_{-j})$ denotes f as function of the j^{th} coordinate with all other dimensions fixed at x_{-j} . Lastly, the univariate *TV* operator in the integrand in (20) is to be interpreted in the essential variation sense, as in (19).

Remark 4. The assumption of convexity of U in Theorem 1 is used for simplicity, to ensure that each coordinatewise slice of U—intersecting it with a line segment parallel to the coordinate axis—is an interval. The proof trivially extends to the case in which each slice is a finite union of intervals; more complex structures could likely be handled via more complex arguments.

Remark 5. The above result is inspired by Theorem 5.22 of Evans and Gariepy (2015). In the proof, we mollify f in order to invoke the representation in (18) for the TV of a smooth function, and then we leverage the separability of the ℓ_1 norm (that is, we leverage the fact that the integrand in (18) decomposes into a sum of absolute partial derivatives) in order to derive (20). Curiously, an analogous result does not seem straightforward to derive for the case of isotropic TV; this being defined by using an ℓ_2 norm constraint on the test function ϕ in (17) (and for functions in $W^{1,2}(U)$, it would reduce to the integral of ℓ_2 norm of the weak derivative, instead of the ℓ_1 norm as in (18)).

3.4 Connection to KTV smoothness

We connect the representation in (20) to the KTF penalty functional, which we call KTV smoothness. First note that we can rewrite the definition of the anisotropic TV of a function f in (17) as

$$\mathrm{TV}(f;U) = \sum_{j=1}^{d} \sup_{U} \left\{ \int_{U} f(x) \, \frac{\partial \phi(x)}{\partial x_j} \, dx : \phi \in C_c^{\infty}(U), \ |\phi(x)| \le 1 \text{ for all } x \in U \right\},$$

where $C_c^{\infty}(U)$ is the space of infinitely continuously differentiable real-valued functions on U. Note that $V_j(f; U)$, as defined above, measures the variation of f along the j^{th} coordinate direction. Now consider the following definition of k^{th} order multivariate TV, for an integer $k \ge 0$:

$$\mathrm{TV}^{k}(f;U) = \sum_{j=1}^{d} V_{j}\left(\frac{\partial^{k} f}{\partial x_{j}^{k}};U\right),$$
(21)

where $\partial^k f / \partial x_j^k$ denotes the k^{th} partial weak derivative of f with respect to x_j . The formulation in (21) is, in a sense, among the many possible options for higher-order TV in the multivariate setting, the "most" anisotropic. It only looks at the variation in the partial derivatives along the coordinate directions with respect to which they are defined (that is, the variation in the j^{th} partial derivative along the j^{th} coordinate axis).

Thanks to the representation in Theorem 1, we can rewrite the definition of k^{th} order TV in (21) as:

$$\mathrm{TV}^{k}(f;U) = \sum_{j=1}^{d} \int_{U_{-j}} \mathrm{TV}\left(\frac{\partial^{k} f(\cdot, x_{-j})}{\partial x_{j}^{k}}\right) dx_{-j},$$
(22)

where we have made the dependence on the domain $I_{x_{-j}}$ in the TV operator in the integrand implicit. Finally, we are ready to draw the connection to KTF. Observe that the penalty functional underlying KTF, the second term in the criterion of its continuous-time formulation (12), is given by taking the notion of k^{th} order TV in (22) and approximating the integral via discretization; that is, the integral over U_{-i} is simply replaced by a sum over an embedded lattice.

4 Smoothness classes

We present various discrete smoothness classes, then connect them to each other and to traditional Holder smoothness classes defined in continuous-time, to derive what we refer to as *canonical scalings* for the radii of the discrete classes. This paves the way for the minimax analysis in the next section.

4.1 Discrete TV and Sobolev classes

First we define the k^{th} order *Kronecker total variation* (KTV) class, for a radius $\rho > 0$, by

$$\mathcal{T}_{n,d}^{k}(\rho) = \left\{ \theta \in \mathbb{R}^{n} : \|D_{n,d}^{(k+1)}\theta\|_{1} \le \rho \right\}.$$
(23)

It is a priori unclear what scaling for the radius ρ in (23) makes for an "interesting" smoothness class for theoretical analysis. This is discussed at length in Sadhanala et al. (2016), and was one of the original motivations for that paper. There, it is shown that taking ρ to be a constant (as $n \to \infty$) leads to seemingly very fast minimax rates, but in this regime trivial estimators turn out to be rate optimal (such as the sample mean estimator $\hat{\theta}_i = \bar{y}, i = 1, ..., n$).

In order to begin reasoning about scalings for ρ in (23), it helps to define the order k + 1 discrete ℓ_2 -Sobolev class:

$$\mathcal{W}_{n,d}^{k+1}(\rho) = \left\{ \theta \in \mathbb{R}^n : \|D_{n,d}^{(k+1)}\theta\|_2 \le \rho \right\}.$$
(24)

Observe that $\mathcal{W}_{n,d}^{k+1}(\rho)$ only considers partial derivatives of order k+1 aligned with one of the coordinate axes, rather than considering all mixed derivatives of total order k+1, as we would in a traditional Sobolev class. For simplicity, we drop reference to the ℓ_2 prefix when referring to (24) henceforth.

By the inequality $||v||_2 \leq \sqrt{p} ||\theta||_1$ for vectors $v \in \mathbb{R}^p$, and the fact that the number of rows of $D_{n,d}^{(k+1)}$ can be upper bounded by dn, we have the following embedding:

$$\mathcal{W}_{n,d}^{k+1}(\rho) \subseteq \mathcal{T}_{n,d}^k(\sqrt{dn}\rho), \quad \text{for any } \rho > 0.$$

This shows that any reasonable regime for analysis must have ρ varying with n in (23), or in (24) (or both), because a constant radius in one class would translate into a growing or diminishing radius in the other, by the above display. However, it still leaves unspecified what precise scalings for the radii in (23) and (24), would correspond to "interesting" classes, comparable in some sense to choices of radii in analogous continuous-time TV or Sobolev smoothness classes. We answer this question in the next subsection, by introducing discrete and continuum Holder classes, and pursuing further embeddings.

4.2 Discrete and continuum Holder classes

Now we recall the traditional definition for the k^{th} order Holder class of functions from $[0,1]^d$ to \mathbb{R} , of radius L > 0:

$$C^{k}(L;[0,1]^{d}) = \left\{ f:[0,1]^{d} \to \mathbb{R} : f \text{ is } k \text{ times differentiable and for all integers } \alpha_{1}, \dots, \alpha_{d} \ge 0, \\ \text{with } \alpha_{1} + \dots + \alpha_{d} = k, \ \left| \frac{\partial^{k} f(x)}{\partial x_{1}^{\alpha_{1}} \cdots \partial x_{d}^{\alpha_{d}}} - \frac{\partial^{k} f(z)}{\partial x_{1}^{\alpha_{1}} \cdots \partial x_{d}^{\alpha_{d}}} \right| \le L \|x - z\|_{2}, \text{ for all } x, z \in [0,1]^{d} \right\}.$$

We define a discretized version of this class by simply evaluating the functions in $C^k(L; [0, 1]^d)$ on the lattice $Z_{n,d}$:

$$\mathcal{C}_{n,d}^k(L) = \left\{ \theta \in \mathbb{R}^n : \text{there exists some } f \in C^k(L; [0, 1]^d) \text{ such that } \theta(x) = f(x), x \in Z_{n,d} \right\}.$$
(25)

The next proposition derives embeddings for the discrete Holder class (25) into the discrete Sobolev (24) and KTV (23) classes. It is a direct consequence of Lemma A.6 in Sadhanala et al. (2017) (which is a classical result of sorts that quantifies the error of the forward difference approximation of the derivative of a Holder function).

Proposition 3 (Sadhanala et al. 2017). *The discrete classes in* (23)–(25) *satisfy, for any* L > 0,

$$\mathcal{C}_{n,d}^{k}(L) \subseteq \mathcal{W}_{n,d}^{k+1}\left(c_{1}Ln^{\frac{1}{2}-\frac{k+1}{d}}\right) \subseteq \mathcal{T}_{n,d}^{k}\left(c_{2}Ln^{1-\frac{k+1}{d}}\right),\tag{26}$$

where $c_1, c_2 > 0$ are constants depending only on k, d.

Motivated by the last result, as in Sadhanala et al. (2017), we define the *canonical scalings* for the discrete Sobolev and KTV classes as

$$B_n^* = n^{\frac{1}{2} - \frac{k+1}{d}},\tag{27}$$

$$C_n^* = n^{1 - \frac{k+1}{d}},\tag{28}$$

so that $C_{n,d}^k(1) \subseteq W_{n,d}^{k+1}(c_1B_n^*) \subseteq \mathcal{T}_{n,d}^k(c_2C_n^*)$, for constants $c_1, c_2 > 0$ that depend only on k, d. Thus, by analogy to classical results on nonparametric estimation over Holder spaces, we should expect the minimax rate over $W_{n,d}^{k+1}(B_n^*)$ (in the squared ℓ_2 norm) to be $n^{-2(k+1)/(2(k+1)+d)}$. This is indeed the case, as we will show at the end of Section 5. The minimax rate over $\mathcal{T}_{n,d}^k(C_n^*)$, on the other hand, will turn out to be more exotic, and is the focus of the majority of the next section.

5 Estimation theory

We derive a number of results on estimation theory over KTV classes. We begin by deriving upper bounds on the error of the KTF estimator, and then study lower bounds. Throughout, we assume the data model in (1) with $\theta_{0i} = f_0(x_i)$, i = 1, ..., n and i.i.d. normal errors, to be precise:

$$y_i \sim N(\theta_{0,i}, \sigma^2), \quad \text{independently, for } i = 1, \dots, n.$$
 (29)

To set some basic notation, based on estimators $\hat{\theta}$ of the mean θ_0 in (29), we define for a subset $\mathcal{K} \subseteq \mathbb{R}^n$,

$$R(\mathcal{K}) = \inf_{\hat{\theta}} \sup_{\theta_0 \in \mathcal{K}} \frac{1}{n} \mathbb{E} \| \hat{\theta} - \theta_0 \|_2^2,$$

which is called the *minimax risk* over \mathcal{K} . Also of interest will be

$$R_L(\mathcal{K}) = \inf_{\hat{\theta} \text{ linear }} \sup_{\theta_0 \in \mathcal{K}} \frac{1}{n} \mathbb{E} \| \hat{\theta} - \theta_0 \|_2^2.$$

called the *minimax linear risk* over \mathcal{K} , the infimum being restricted to linear estimators $\hat{\theta}$ (that is, of the form $\hat{\theta} = Sy$ for a matrix $S \in \mathbb{R}^{n \times n}$). To finish our discussion of notation, for deterministic sequences a_n, b_n we write $a_n = O(b_n)$ when a_n/b_n is upper bounded for large enough n, we write $a_n = \Omega(b_n)$ when $a_n^{-1} = O(b_n^{-1})$, and $a_n \asymp b_n$ when both $a_n = O(b_n)$ and $a_n = \Omega(b_n)$. For random sequences A_n, B_n , we write $A_n = O_{\mathbb{P}}(B_n)$ when A_n/B_n is bounded in probability. In the theory that follows, all asymptotics are for $n \to \infty$ with k, d fixed.

As a side remark, although not the focus of the current paper, analogous theory can be established for graph trend filtering on grids (see Section 1.3 for a discussion of its relation to KTF), which we defer to Appendix B.

5.1 Upper bounds on estimation risk

To derive upper bounds on the risk of KTF, we leverage the following simple generalization of a key result from Wang et al. (2016). Here and henceforth, for an integer $a \ge 1$, we abbreviate $[a] = \{1, \ldots, a\}$.

Theorem 2 (Wang et al. 2016). Consider the generalized lasso estimator $\hat{\theta}$ with penalty matrix $D \in \mathbb{R}^{r \times n}$, defined by the solution of

$$\underset{\theta \in \mathbb{R}^n}{\text{minimize}} \ \frac{1}{2} \|y - \theta\|_2^2 + \lambda \|D\theta\|_1$$
(30)

Suppose that D has rank q, and denote by $\xi_1 \leq \cdots \leq \xi_q$ its nonzero singular values. Also let $u_1, \ldots, u_q \in \mathbb{R}^r$ be the corresponding left singular vectors. Assume that these vectors, except possibly for those in a set $I \subseteq [q]$, are incoherent, meaning that for a constant $\mu \geq 1$,

$$||u_i||_{\infty} \le \mu/\sqrt{n}, \quad i \in [q] \setminus I.$$

Then under the data model (29), choosing

$$\lambda \asymp \mu \sqrt{\frac{\log r}{n} \sum_{i \in [q] \setminus I} \frac{1}{\xi_i^2}},$$

the generalized lasso estimator satisfies

$$\frac{1}{n} \|\hat{\theta} - \theta_0\|_2^2 = O_{\mathbb{P}} \left(\frac{\text{nullity}(D)}{n} + \frac{|I|}{n} + \frac{\mu}{n} \sqrt{\frac{\log r}{n} \sum_{i \in [q] \setminus I} \frac{1}{\xi_i^2}} \cdot \|D\theta_0\|_1} \right).$$
(31)

We will now apply this result to KTF in (7), and choose the set I in order to balance the second and third terms on the right-hand side in (31). Throughout, we will make reference to the *effective degree of smoothness* (or effective smoothness for short), defined by

$$s = \frac{k+1}{d}.$$

Theorem 3. Let $\hat{\theta}$ denote the KTF estimator in (7). Under the data model (29), denote $C_n = \|D_{n,d}^{(k+1)}\theta_0\|_1$, and assume $C_n > 0$. Choosing

$$\lambda \asymp \begin{cases} \sqrt{\log n} & \text{if } s < 1/2, \\ \log n & \text{if } s = 1/2, \\ (\log n)^{\frac{1}{2s+1}} (n/C_n)^{\frac{2s-1}{2s+1}} & \text{if } s > 1/2, \end{cases}$$

the KTF estimator satisfies

$$\frac{1}{n} \|\hat{\theta} - \theta_0\|_2^2 = O_{\mathbb{P}} \left(\frac{1}{n} + \frac{\lambda}{n} C_n \right)$$

Remark 6. The result in the above theorem for the case of k = 0 (TV denoising) and any d was already established in Hutter and Rigollet (2016). The result for d = 2 and any k was given in Sadhanala et al. (2017). For d = 1 and any k, the result was established in Mammen and van de Geer (1997); Tibshirani (2014) (though the results in the latter papers are sharper by log factors).

Theorem 3 covers all k and d. Its proof, an application of Theorem 2 to KTF, requires checking the incoherence of the penalty matrix $D = D_{n,d}^{(k+1)}$, and bounding the partial sum of squared reciprocal singular values $\sum_{i \in [q] \setminus I} \xi_i^{-2}$, for an appropriate set I that corresponds to small eigenvalues. The former—checking the incoherence of the left singular vectors of the KTF penalty matrix—turns out to be the harder calculation. However, the hardest part of this calculation was already done in Sadhanala et al. (2017), who established the incoherence of $D_n^{(k+1)}$ (the trend filtering penalty matrix, or equivalently, the KTF penalty matrix when d = 1) by appealing to complex approximation arguments for the eigenvectors of Toeplitz matrices. To handle KTF in arbitrary dimension d, in the current paper, we use careful arguments that relate singular vectors of Kronecker products to singular vectors of their constituent matrices.

Remark 7. It is interesting to note that the case $s \le 1/2$ (that is, $2k + 2 \le d$) appears to be special, in that KTF ends up being adaptive to the underlying level of smoothness C_n . In other words, the prescribed choice of tuning parameter is $\lambda \asymp \sqrt{\log n}$ when s < 1/2, and $\lambda \asymp \log n$ when s = 1/2, neither of which depend on C_n .

Under the canonical scaling $C_n \asymp C_n^*$ in (28), the error bound in Theorem 3 reduces to the following.

Corollary 1. Assume the conditions of Theorem 3. Under the canonical scaling in (28) for the KTV smoothness level, $C_n \simeq C_n^* = n^{1-s}$, the KTF estimator satisfies

$$\frac{1}{n} \|\hat{\theta} - \theta_0\|_2^2 = \begin{cases} O_{\mathbb{P}} \left(n^{-s} \sqrt{\log n} \right) & \text{if } s < 1/2, \\ O_{\mathbb{P}} \left(n^{-s} \log n \right) & \text{if } s = 1/2, \\ O_{\mathbb{P}} \left(n^{-\frac{2s}{2s+1}} (\log n)^{\frac{1}{2s+1}} \right) & \text{if } s > 1/2. \end{cases}$$
(32)

Remark 8. Recall the continuous-time analog of the KTF penalty in (12). It turns out, as we show in Appendix A.6, that the assumption that the mean vector θ_0 is KTV smooth in Theorem 3 and Corollary 1 can be broadened into an assumption that the KTV of the mean *function* f_0 is KTV smooth, in the sense of the penalty functional in (12).

5.2 Lower bounds on estimation risk

Next we give lower bounds on the minimax risk over KTV classes.

Theorem 4. The minimax risk for KTV class defined in (23) satisfies, for any sequence $C_n \leq n$,

$$R\left(\mathcal{T}_{n,d}^{k}(C_{n})\right) = \Omega\left(\frac{1}{n} + \frac{C_{n}}{n} + \left(\frac{C_{n}}{n}\right)^{\frac{2}{2s+1}}\right).$$
(33)

Remark 9. The result in Theorem 4 for $s \ge 1/2$ (that is, $2k + 2 \ge d$) was already derived in Sadhanala et al. (2017). More precisely, these authors established the third term on the right-hand side in (33), by using the Holder embedding in (26) of Proposition 3, and suitably adapting classical results on minimax bounds for Holder spaces (Korostelev and Tsybakov, 2003; Tsybakov, 2009), which ends up being tight in the s > 1/2 regime. Moreover, the lower bound in the middle term in (33) was derived by Sadhanala et al. (2016), for k = 0 and all d, which was obtained by embedding an appropriate ℓ_1 ball into $\mathcal{T}_{n,d}^k(C_n)$, and appealing to minimax theory over ℓ_1 balls from Birge and Massart (2001). This ends up being tight for k = 0 and all d.

In the current work, we establish tight lower bounds over all k, d, by essentially combining these two strategies. As we will see comparing to upper bounds in the next remark, the Holder embedding ends up being tight for s > 1/2, and the ℓ_1 embedding for $s \le 1/2$.

Remark 10. Plugging in the canonical scaling $C_n \simeq C_n^* = n^{1-s}$ from (28), and simplifying to dominant terms, we see that (33) becomes

$$R(\mathcal{T}_{n,d}^k(C_n^*)) = \begin{cases} \Omega(n^{-s}) & \text{if } s \le 1/2, \\ \Omega(n^{-\frac{2s}{2s+1}}) & \text{if } s > 1/2. \end{cases}$$

By comparing this to (32), we can see that KTF is minimax rate optimal for estimation over KTV classes, under the canonical scaling, up to log factors.

Remark 11. From the upper bound in (32) and the Holder embedding in (26), we see that for $s \ge 1/2$, KTF achieves the rate of $n^{-2s/(2s+1)}$ (up to log factors) over $\mathcal{H}_d^k(1)$. This matches the optimal rate for estimation over Holder classes (see Sadhanala et al. (2017) for a formal statement and proof for the discretized class $\mathcal{H}_d^k(1)$), which means that KTF automatically adapts Holder smooth signals.

However, when s < 1/2, the same results show that KTF achieves a rate of n^{-s} (ignoring log factors) over $\mathcal{H}_d^k(1)$, which is slower than the optimal rate of $n^{-2s/(2s+1)}$. Whether this upper bound is pessimistic and KTF can actually adapt to $\mathcal{H}_d^k(1)$, or whether this upper bound is tight and KTF fails to adapt to Holder smooth signals, remains to be formally resolved. Later, we investigate this empirically in Section 8.3.

Remark 12. It is interesting to compare minimax results for anisotropic Besov spaces, under the white noise model. Past work on this topic includes Neumann (2000) with a focus on hyperbolic wavelets, as well as Kerkyacharian et al. (2001, 2008) with a focus on Lepski's method applied to kernel smoothing. A nice summary of past work, and what appears to be the most comprehensive results, can be found in Lepski (2015). It should be noted that this line of work considers a more general setup than ours (albeit in the white noise model), in a few ways: anisotropic Besov classes with an arbitrary smoothness index in each coordinate direction; error measured in L^p norm, for arbitrary $p \ge 1$; and so on. That said, translating their results to match our setting as best as we can (recalling general embeddings of BV spaces into Besov spaces; for example, DeVore and Lorentz (1993)), we find that the minimax rate under the squared L^2 loss, for the anisotropic Besov class with integrability index 1, smoothness index k + 1 in each coordinate direction, and any third index $q \ge 1$, is indeed $n^{-2s/(2s+1)}$ for s > 1/2 (or 2k + 2 > d). This regime is what Lepski and others refer to as the "dense zone".

In what they refer to as the "sparse zone", $s \le 1/2$, the minimax risk under the L^2 loss for the same Besov class is a constant. This is due to the fact that this Besov space fails to embed compactly into L^2 (see Section 5.5 of Johnstone (2015) for a general statement of the implications of this phenomenon). If the underlying regression function is itself additionally assumed to be bounded in L^{∞} norm, then we believe the minimax rate in their white noise setting will be n^{-s} , matching that in our discrete setting. Evidence of this claim includes the result in del Álamo et al. (2021) for the BV space (under the white noise model), who derive a rate of $n^{-1/d}$, assuming such L^{∞} boundedness; as well as the Besov results in Goldenshluger and Lepski (2014), who also assume L^{∞} boundedness, but study density estimation.

5.3 Minimax rates for linear smoothers

We now study whether linear smoothers can achieve the minimax rate over the KTV class in (23). Before stating our result, we define a truncated eigenmaps estimator based on $D_{n,d}^{(k+1)}$ as follows. Denote by $\xi_i \ge 0$, $i \in [N]^d$ its singular values (noting that $(k+1)^d$ of these are zero), where along each dimension in the multi-index, the singular values are sorted in increasing order. Denote also by $v_i \in \mathbb{R}^n$, $i \in [N]^d$ its corresponding right singular vectors. Then, for a subset $Q \subseteq [N]^d$, we denote by $V_Q \in \mathbb{R}^{n \times |Q|}$ the matrix with columns given by v_i , $i \in Q$, and define the projection estimator

$$\hat{\theta} = V_Q V_Q^{\mathsf{T}} y. \tag{34}$$

Note that this reduces to the Laplacian eigenmaps estimator (here the Laplacian is that of the grid graph) when k = 0.

Theorem 5. The minimax linear risk over the KTV class in (23) satisfies, for any sequence $C_n \leq \sqrt{n}$,

$$R_L(\mathcal{T}_{n,d}^k(C_n)) = \begin{cases} \Omega(1/n + C_n^2/n) & \text{if } s < 1/2, \\ \Omega(1/n + C_n^2/n\log(1 + n/C_n^2)) & \text{if } s = 1/2, \\ \Omega(1/n + (C_n^2/n)^{\frac{1}{2s}}) & \text{if } s > 1/2. \end{cases}$$
(35)

This is achieved in rate by the projection estimator in (34), by setting $Q = [\tau]^d$ for $\tau^d \simeq (C_n n^{s-1/2})^{1/s}$, in the case s > 1/2. When s < 1/2, the simple polynomial projection estimator, which projects onto all multivariate polynomials of max degree k (equivalently, the estimator in (34) with $Q = [k+1]^d$), achieves the rate in (35). When s = 1/2, either estimator achieves the rate in (35) up to a log factor. Lastly, if $C_n^2 = O(n^{\alpha})$ for $\alpha < 1$, and still s = 1/2, then either estimator achieves the rate in (35) without the additional log factor.

Remark 13. Plugging in the canonical scaling $C_n \simeq C_n^* = n^{1-s}$ from (28), and simplifying to dominant terms, we see that (35) becomes

$$R(\mathcal{T}_{n,d}^k(C_n^*)) \asymp \begin{cases} 1 & \text{if } s \le 1/2, \\ n^{-\frac{2s-1}{2s}} & \text{if } s > 1/2. \end{cases}$$

These minimax linear rates display a stark difference to the minimax rates for the KTV class in Remark 10 (achieved by KTF up to log factors, in (32)). When s > 1/2, the minimax linear rate of $n^{-(2s-1)/(2s)}$ can be interpreted as the rate of the optimal nonlinear estimator when the problem has a half degree less of effective smoothness (s - 1/2 in place of s). When $s \le 1/2$, even more dramatically, we see that *no linear smoother can even be consistent* over the KTV class, in the sense of worst-case risk. This contributes an interesting addition to the line of work on the suboptimality of linear smoothers for nonparametric regression over heterogeneous spaces, which begun with the seminal work of Donoho and Johnstone (1998).

5.4 Summary of rates

Table 1 presents a summary of the minimax rates derived in the previous three subsections. (It offers a more detailed summary than Figure 2.) The upper bound on minimax risk is from Corollary 1, the lower bound on the minimax risk from Theorem 4 and Remark 10, and the minimax linear risk is from Theorem 5 and Remark 13.

Regime	R, upper bound	R, lower bound	R_L , linear risk
s < 1/2	$n^{-s}\sqrt{\log n}$	n^{-s}	1
s = 1/2	$n^{-\frac{1}{2}}\log n$	$n^{-\frac{1}{2}}$	1
s > 1/2	$n^{-\frac{2s}{2s+1}}(\log n)^{\frac{1}{2s+1}}$	$n^{-\frac{2s}{2s+1}}$	$n^{-\frac{2s-1}{2s}}$

Table 1: Minimax rates over the KTV class $\mathcal{T}_{n,d}^k(C_n^*)$, where the canonical scaling is $C_n^* = n^{1-s}$, and recall s = (k+1)/d. We use the abbreviations $R = R(\mathcal{T}_{n,d}^k(C_n^*))$ and $R_L = R_L(\mathcal{T}_{n,d}^k(C_n^*))$.

5.5 Minimax rates over Sobolev classes

For completeness, we establish the minimax rate for the (discrete) Sobolev class defined in (24). The lower bounds are simply those from the Holder class (due to the embedding in Proposition 3), and as we show next, this is achieved in rate by the eigenmaps estimator in (34).

Theorem 6. The minimax risk over the (discrete) Sobolev class in (24) satisfies, for any sequence $B_n \leq \sqrt{n}$,

$$R(\mathcal{W}_{n,d}^{k+1}(B_n)) \simeq \Omega\left(\frac{1}{n} + \left(\frac{B_n^2}{n}\right)^{\frac{1}{2s+1}}\right).$$

The lower bound is due to the Holder embedding in (26) (and the lower bound on the discretized Holder class derived in Sadhanala et al. (2017)), and the upper bound is from the estimator in (34), with $Q = [\tau]^d$ for $\tau^d \simeq (B_n^2 n^{2s})^{1/(2s+1)}$. Finally, under the canonical scaling $B_n \simeq B_n^* = n^{1/2-s}$ from (27), the minimax rate simplifies to $n^{-2s/(2s+1)}$.

6 Optimization algorithms

In this section, we describe a number of numerical algorithms for solving the convex KTF problem (7), analyze their asymptotic time complexity, and benchmark their performance. In particular, we will describe a family of specialized ADMM algorithms that adapt to the structure of the KTF problem, and as we will show, can find moderately accurate solutions much faster than a general purpose "off-the-shelf" solver. This general purpose solver can be applied to the dual of (7), which is a simple box-constrained quadratic program, and is amenable to standard interior point methods. Details are deferred to the appendix.

6.1 Specialized ADMM algorithms

Motivated by the popularity of operating splitting methods in machine learning over the last decade, and specifically by their success in application to trend filtering and TV denoising problems (for example, see Ramdas and Tibshirani (2016); Wang et al. (2016); Barbero and Sra (2014); Tansey and Scott (2015)), we investigate application of similar methods to Kronecker trend filtering. We study a proximal Dykstra algorithm, Douglas-Rachford splitting, and a family of specialized ADMM algorithms. For brevity, the details on the former two are deferred to the appendix.

Our specialized ADMM approach is inspired by that of Ramdas and Tibshirani (2016), for univariate trend filtering. To reformulate (7) into "ADMM form", where the criterion decomposes as a sum of functions of separate optimization variables, we must introduce auxiliary variables. To do so, we rely on the following observation that decomposes the KTF penalty operator into the product of a block-diagonal matrix and a lower-order KTF penalty operator.

Proposition 4. For each j = 1, 2, ..., k + 1, the KTF penalty operator in (8) obeys (with $D_N^{(0)} = I_N$ for convenience):

$$D_{n,d}^{(k+1)} = \underbrace{\begin{bmatrix} D_N^{(k+1-j)} \otimes I_N \otimes \cdots \otimes I_N & & & \\ & I_N \otimes D_N^{(k+1-j)} \otimes \cdots \otimes I_N & & \\ & & \ddots & & \\ & & & I_N \otimes I_N \otimes \cdots \otimes D_N^{(k+1-j)} \end{bmatrix}}_{(k+1-j)} D_{n,d}^{(j)}$$

$$M_{n,d}^{(k+1-j)}$$

This follows directly from the the univariate recursion in (4), and the Kronecker structure in (8) (particularly, the mixed-product property of Kronecker products), and therefore we omit its proof. Observe that each diagonal block of the block-diagonal matrix $M_{n,d}^{(k+1-j)}$ is itself—possibly after appropriate permutation of the row and column order—a block-diagonal matrix with all diagonal blocks equal to $D_N^{(k+1-j)}$. This is a key fact that we will leverage shortly.

Now fix any $j \in \{1, \ldots, k+1\}$. We can reformulate (7) as:

$$\begin{array}{ll} \underset{\theta,z}{\text{minimize}} & \frac{1}{2} \|y - \theta\|_2^2 + \lambda \|M_{n,d}^{(k+1-j)}z\|_1 \\ \text{subject to } & z = D_{n,d}^{(j)}\theta. \end{array}$$

$$(36)$$

The augmented Lagrangian associated with (36), for an augmented Lagrangian parameter $\rho \ge 0$, is:

$$L_{\rho}(\theta, z, u) = \frac{1}{2} \|y - \theta\|_{2}^{2} + \lambda \|M_{n,d}^{(k+1-j)}z\|_{1} + \frac{\rho}{2} \|z - D_{n,d}^{(j)}\theta + u\|_{2}^{2} - \frac{\rho}{2} \|u\|_{2}^{2}.$$

ADMM iteratively performs a separate minimization over each of primal variables θ , z, then updates the dual variable u via gradient ascent. Namely, given some initialization $\theta^{(0)}$, $z^{(0)}$, $u^{(0)}$, it repeats the following, for t = 1, 2, 3, ...:

$$\theta^{(t+1)} = \left(I_n + \rho [D_{n,d}^{(j)}]^\mathsf{T} D_{n,d}^{(j)} \right)^{-1} \left(y + \rho [D_{n,d}^{(j)}]^\mathsf{T} (z^{(t)} + u^{(t)}) \right), \tag{37}$$

$$z^{(t+1)} = \operatorname{prox}_{\frac{\lambda}{\rho} \| M_{n,d}^{(k+1-j)}(\cdot) \|_1} \Big(D_{n,d}^{(j)} \theta^{(t+1)} - u^{(t)} \Big),$$
(38)

$$u^{(t+1)} = u^{(t)} + z^{(t+1)} - D_{n,d}^{(j)} \theta^{(t+1)}.$$
(39)

Here we use the notation $prox_h(\cdot)$ for the proximal operator associated with a function h. Below we make a few remarks on the computation.

- When j = 1, the matrix $[D_{n,d}^{(j)}]^{\mathsf{T}} D_{n,d}^{(j)}$ in (37) is the graph Laplacian of the *d*-dimensional grid, which decomposes into the Kronecker sum of Laplacians of (univariate) chain graphs. This can be diagonalized by a *d*-dimensional discrete cosine transform (DCT) (see, for example, the proof of Corollary 8 in Wang et al. (2016)). Computationally, this is essentially sequentially applying univariate DCTs to every dimension. This implies that the θ -update in (37) can be done in $O(n \log n)$ time. Further improvements (to linear-time) should be possible with multi-grid methods.
- By the key fact mentioned after the proposition, the z-update in (38) can be decomposed into dN^{d-1} univariate trend filtering problems, each of order k + 1 j and each with sample size N. These can be solved in parallel, in a total time that is nearly-linear in n; for example, using either the univariate ADMM approach of Ramdas and Tibshirani (2016) or the primal-dual interior point method (PDIP) of Kim et al. (2009). PDIP is likely the best option for reasonably small N, and using it, the update (38) can be done in $O(dN^{d-1}N^{1.5}) = O(n^{1+1/(2d)})$ time.
- When j = k or k + 1, the z-update (38) can be performed even more efficiently, in O(n) time. This is because it reduces to soft-thresholding for j = k + 1, and to separate univariate TV denoising problems for j = k. In the former case, the linear time complexity is obvious; in the latter, it is due to the dynamic programming (DP) of Johnson (2013).
- The case k = 0 is quite favorable, as we can use DCT in (37) and soft-thresholding in (38), by choosing j = 1, or simple coordinatewise shrinkage in (37) and DP in (38), by choosing j = 0. Both are efficient, but the latter ends up being generally the better approach, and can be seen as the ADMM-analog of Barbero and Sra (2014).
- Among the higher-order cases k ≥ 1, the case k = 1 ends up being quite special, because j = 1 simultaneously supports the DCT solver in (37) and DP in (38). When k ≥ 2, we essentially need to decide in between these highly efficient subroutines (choosing either j = 1 or j = k).

In summary, each iteration (cycle of updates over θ, z, u) of the proposed ADMM algorithm is O(n) for k = 0, 1. For $k \ge 2$, the time complexity is $O(n^{1+1/(2d)})$ when we choose j = 1, which we refer to as ADMM *Type I*. When we choose j = k, which we refer to as ADMM *Type II*, the time complexity is dominated by the sparse linear system solve in (37). This linear system should be well-conditioned for reasonable ranges of ρ , thus the standard conjugate gradient method will be able to solve it in approximately linear-time (proportional to the number of nonzero elements).

6.2 Empirical comparisons

We now compare the ADMM algorithms developed in the last subsection to Douglas-Rachford and proximal Dykstra methods applied to (7), as well as as general purpose solver—Gurobi, which is free for academic use—applied to the dual of (7).³ From the family of specialized ADMM algorithms, we pay particular attention to ADMM Types I and II, which correspond to j = 1 and j = k, respectively. We also consider ADMM Type 0, which sets j = 0, because it is closely related and should perform similarly to the Douglas-Rachford and proximal Dykstra methods. In all ADMM algorithms, we adopt an adaptive choice of ρ that balances the primal and dual suboptimality (Boyd et al., 2011).

To ensure a fair comparison between ADMM Types I and II, which we will see are generally the best performing methods (and thus the comparison between them is of particular interest), we use optimized C++ implementations for each of their prox subroutines; for Type I, this is the DP algorithm for univariate TV denoising, and for Type II, this is the PDIP algorithm for univariate trend filtering; and in both cases, we use C++ implementations from Ramdas and Tibshirani (2016). Aside from specialized subroutines, implementation of all iterative algorithms is in MATLAB.

The results are presented in Figures 3 and 4. Figure 3 compares the operator splitting algorithms for denoising the standard "Lena" method at a resolution of 256×256 . The KTF orders are taken to be k = 1, 2, 3, corresponding to the columns in the figure. For each k, the solution returned by Gurobi is used to define the optimal criterion value, which is is then used to measure the suboptimality gap of solutions returned the iterative methods. ADMM Type I is generally the winner in all cases, whether measured by iteration or especially by wall-clock time.

Figure 4 compares ADMM Type I to Gurobi for varying k, and also for varying resolutions of the underlying Lena image. In all cases, ADMM Type I obtains a moderate-quality solution in less one second—which is sometimes two orders of magnitude faster than the off-the-shelf solver provided by Gurobi. It should be further noted that Gurobi is highly-optimized, whereas our ADMM Type I implementation is not—recall, only the prox subroutine is optimized, and the outer looping is performed in MATLAB. Transporting the entire algorithm to C++ would clearly yield further improvements in efficiency. Of course, if a truly high-accuracy solution is required, then Gurobi may be the best option. However, its strong performance in this subsection suggests that ADMM Type I is an efficient and useful approach for many applications in statistics and machine learning, where moderate-accuracy solutions suffice.

7 Interpolation algorithm

In this section, we derive an algorithm to extend the KTF solution in (7), defined only at points in the lattice $Z_{n,d}$, to a function defined on all of $[0, 1]^d$. As explained in Remark 3, this is made possible by the continuous-time formulation for KTF in (12) of Proposition 2, which, recall, relates to the original discrete-time problem (7) in that at their solutions, we have $\hat{\theta}_i = \hat{f}(x_i)$, for i = 1, ..., n. Given the coefficients that define the function $\hat{f} \in \mathcal{H}_{n,d}^k$ in its expansion in terms of the tensor product of univariate falling factorial basis functions, that is, given the solution $\hat{\alpha}$ in (13), we can form the interpolated prediction $\hat{f}(x)$ at an arbitrary point $x \in [0, 1]^d$ by evaluating the basis expansion at x, as in (14).

While this is conceptually the easiest way to interpolate the fitted values $f(x_i)$, i = 1, ..., n to form the prediction at an arbitrary $x \in [0, 1]^d$, it is not the most efficient. The expression in (14) takes $O(\|\hat{\alpha}\|_0(k+1)^d)$ operations, where $\|\hat{\alpha}\|_0$ denotes the number of nonzero elements in $\hat{\alpha}$ (the number of active basis functions), because evaluating each univariate basis function $h_{i_j}(x_j)$ takes O(k+1) operations (it being a product of k+1 terms, recall (10)). In what follows, we present an interpolation algorithm that takes only $O((k+1)^{d+1})$ operations, a big savings over (14) when $\|\hat{\alpha}\|_0$ is large. Moreover, our algorithm acts directly on the solution $\hat{\theta}$ in (7), meaning that we never have to solve (13) in the first place.

7.1 Review: univariate interpolation

Our interpolation algorithm for KTF estimates in the multivariate case builds off the univariate discrete spline interpolation algorithm derived in Corollary 2 of Tibshirani (2020). For convenience, we transcribe this in Algorithm 1. Here and henceforth, we use the abbreviation $x_{a:b} = (x_a, \ldots, x_b)$ for integers $a \le b$. Also, we use $f[z_1, \ldots, z_r]$ for the divided difference of a function f at distinct points z_1, \ldots, z_r . Recall, this is defined for r = 2 by

$$f[z_1, z_2] = \frac{f(z_2) - f(z_1)}{z_2 - z_1},$$

 $^{^{3}}$ We add tiny amount of regularization to the dual problem to avoid numerical issues that cause Gurobi to fail. The solution of this regularized problem is first used to reconstruct the primal solution, but then evaluated on the objective function of the original problem, when computing the suboptimality gaps.



Figure 3: Comparison of iterative algorithms for KTF on the standard Lena image of resolution 256×256 (that is, n = 65536), when k = 1, 2, 3, corresponding to the three columns, from left to right. The top row compares the convergence of the suboptimality gap as a function of the number of iterations. The bottom row shows the same but parametrized by wall-clock time in seconds. While these methods have similar sublinear convergence rates (top row), ADMM Types I and II are clearly the fastest (bottom row) to reach a small suboptimality gap, due to their low per-iteration cost. Type I is the overall winner. (Recall that when k = 1, Types I and II coincide, so the blue curves is hidden behind the red curve).



Figure 4: Comparison of ADMM Type I to Gurobi for the same Lena problem. The left panel compares the two methods for varying k at a fixed resolution of 256×256 . The middle panel fixes k = 2 and compares them across varying resolutions. The right panel plots the time needed to achieve a certain relative error, defined by the ratio of the suboptimality gap to the objective value being less than 10^{-2} . Altogether, we see that ADMM Type I achieves a moderate-quality solution several orders of magnitude faster than Gurobi. Furthermore, the right panel shows that Gurobi appears to scale as $O(n^{1.5})$ (to be expected, if it is based on interior point methods internally) whereas ADMM Type I appears to scale closer to O(n).

Algorithm 1 INTERPOLATE-1D($x_{1:n}, \theta_{1:n}, x, k$)

Input: design points $x_{1:n}$ with entries in increasing order; values $\theta_{1:n}$ to interpolate; query point x; integer $k \ge 0$. **Output:** interpolated value f(x), where f is the unique k^{th} order discrete spline with knots in $x_{(k+1):(n-1)}$, such that $f(x_i) = \theta_i, i = 1, \dots, n.$

- 1. If $x = x_i$ for some i = 1, ..., n, then return θ_i .
- 2. Else, if $x > x_{k+1}$ and i is the smallest index such that $x_i > x$ (with i = n when $x > x_n$), then return f(x) as the unique solution of the linear system:

$$f[x_{i-k}, \dots, x_i, x] = 0.$$
 (40)

3. Else, if $x < x_{k+1}$, then return f(x) as the unique solution of the linear system:

$$f[x_1, \dots, x_{k+1}, x] = 0.$$
(41)

(Note that both (40), (41) are linear systems in just one unknown, f(x), since we interpret $f(x_i) = \theta_i$, i = 1, ..., n.)

Algorithm 2 INTERPOLATE $(\{z_{i1}\}_{i=1}^{N_1} \times \cdots \times \{z_{id}\}_{i=1}^{N_d}, \{\theta_i\}_{i \in [N_1] \times \cdots \times [N_d]}, x, k)$

Input: lattice $\{z_{i1}\}_{i=1}^{N_1} \times \cdots \times \{z_{id}\}_{i=1}^{N_d}$ where each set $\{z_{ij}\}_{i=1}^{N_j}$ in the Cartesian product is sorted in increasing order; values $\{\theta_i\}_{i\in[N_1]\times\cdots\times[N_d]}$ over the lattice to interpolate; query point x; integer $k \ge 0$.

Output: interplated value f(x), for the unique function f in the tensor product space of k^{th} degree discrete splines with knots in $\{z_{i1}\}_{i=k+1}^{N_1-1} \times \cdots \times \{z_{id}\}_{i=k+1}^{N_d-1}$, such that $f(z_{i_1,1},\ldots,z_{i_d,d}) = \theta_{i_1,\ldots,i_d}$, $(i_1,\ldots,i_d) \in [N_1] \times \cdots \times [N_d]$.

- 1. If d = 1, then return INTERPOLATE-1D $(z_{1:N_1}, \theta_{1:N_1}, x, k)$.
- 2. Else, let i_1 denote the smallest index such that $x_{i_1,1} \ge z_{i_1,1}$.
- 3. Let $\ell_1 = \min\{\max\{i_1 k, 1\}, N_1 k\}$. 4. Let $\vartheta_p = \operatorname{INTERPOLATE}(\{z_{i2}\}_{i=1}^{N_2} \times \cdots \times \{z_{id}\}_{i=1}^{N_d}, \{\theta_i\}_{i \in \{i_1 + p 1\} \times [N_2] \times \cdots \times [N_d]}, x_{2:d}, k\}$, for $p \in [k + 1]$. 5. Return INTERPOLATE-1D $(z_{\ell_1:(\ell_1 + k), 1}, \vartheta_{1:(k+1)}, x_1, k)$.

and for any $r \geq 3$ by the recursion

$$f[z_1, \dots, z_r] = \frac{f[z_2, \dots, z_r] - f[z_1, \dots, z_{r-1}]}{z_r - z_1}.$$

Note that for evenly-spaced points, this simply coincides with a scaled forward difference; in particular, recalling the notation introduced in Section 1.1, we have

$$f[z, \dots, z + (r-1)/n] = \frac{(r-1)!}{n^r} (\Delta^r f)(z).$$

For more background on divided differences, discrete splines, their connection to the falling factorial basis and to trend filtering, we refer to Tibshirani (2020).

Algorithm 1 takes $O((k+1)^2)$ operations, as (40), (41) are each linear systems in just one unknown, and forming the coefficients in either linear system can be done in $O((k+1)^2)$ operations (this uses a representation of a divided difference as an explicit linear combination of the underlying function evaluations; see, for example, Section 2.1 of Tibshirani (2020)). Note that this assumes $x_{1:n}$ are evenly-spaced design points, because in this case identifying the smallest index i such that $x_i > x$ can be done with integer division. For general design points, the identification step takes $O(\log n)$ operations via binary search, so the total cost would be $O(\log n + (k+1)^2)$.

Moreover, Corollary 2 in Tibshirani (2020) establishes that the value f(x) returned by Algorithm 1 is equal to that produced by the falling factorial basis representation in (11) (for d = 1), where α is the unique coefficient vector such that $f(x_i) = \theta_i, i = 1, ..., n$.

7.2 **Multivariate interpolation**

In the multivariate case, it turns out that we can interpolate within the space of tensor products of k^{th} degree discrete splines in $O((k+1)^{d+1})$ time, assuming a uniformly-spaced lattice. The idea is to reduce the d-dimensional problem calculation to k + 1 interpolation problems, each in dimension d - 1. Figure 5 provides the intuition for d = 2. The algorithm in described in Algorithm 2. We recall the notation from Section 5, where we abbreviate $[a] = \{1, ..., a\}$ for an integer $a \ge 1$.

Algorithm 2 assumes each side length of the lattice is at least k + 1. Its proof of correctness, as well as the running time of $O((k+1)^{d+1})$ for a uniformly-spaced lattice, follows from a straightforward inductive argument over d, whose proof we omit. We note that for an arbitrary lattice, the running time is $O(\sum_{j=1}^{d} \log N_j + (k+1)^{d+1})$. Figure 6 gives some examples of interpolation for d = 2.

8 Experiments

We explore the empirical properties of KTF through a series of experiments that compare its performance against other nonparametric methods.

8.1 Comparison of methods

We study the performance of KTF against other nonparametric estimators by analyzing their empirical mean squared error (MSE) in estimating the function f_0 defined in Figure 1, over a square lattice with $n = 256^2 = 65536$ points. In the experiments that follow, we generate data by adding Gaussian noise to evaluations of f_0 , of the same magnitude illustrated in Figure 1, yielding a signal-to-noise ratio (SNR) of 0.5. Aside from KTF and graph trend filtering (GTF) of orders k = 0, 1, 2 (for k = 0, they both reduce to TV denoising), we also consider second-order Laplacian smoothing (using the squared Laplacian L^2 , where L is the Laplacian matrix of the 2d grid graph), the eigenmaps estimator in (34) with k = 1, kernel smoothing (using a Gaussian kernel), and wavelet smoothing (using Daubechies' least asymmetric wavelets, ten vanishing moments, and hard thresholding). For the latter two estimators, we use the implementations from the R packages np and wavethresh, respectively.

Each estimator is fit over a range of tuning parameters, and its MSE as a function of the tuning parameter is shown in Figure 7. The MSE curves here are averaged over 20 repetitions (each repetition forms a data set by adding noise to f_0). Error bars are shown as well, denoting the standard deviations of the MSE over these 20 repetitions. We can see that the trend filtering estimators (with the exception of TV denoising, which returns a piecewise constant fit that is not well-suited to an underlying signal with this level of smoothness) perform quite a bit better than all competitors, and factoring in the variability over repetitions, KTF and GTF achieve essentially equivalent performance (for common values of k). The superiority of KTF over the linear estimators (Laplacian smoothing, eigenmaps projection, and kernel smoothing), should not come as a surprise—our theory prescribes that KTF should perform better in a minimax sense than any linear smoother when the underlying signal displays heterogeneous smoothness. These results thus serve as a quantitative complement to the qualitative findings in Figure 1, and the theoretical findings in Section 5.

8.2 Rates for heterogeneous signals

We now examine the empirical error rates of KTF and a linear smoother in estimating signals belonging to KTV classes, for k = 0, 1. In both cases, we choose k, d so that the effective degree of smoothness remains s = 1/2, and we use the canonical scaling in (28), so that the two classes under consideration are:

$$\mathcal{T}_{n,2}^0(\sqrt{n})$$
 and $\mathcal{T}_{n,4}^1(\sqrt{n})$. (42)

As representatives for "hard" signals within these two classes, we take the true mean θ_0 to be a "one-hot" signal when k = 0, and a linear "spike" signal when k = 1 (each scaled to the appropriate magnitude). For each sample size n, we compute the MSE of KTF, and the eigenmaps projection estimator in (34) with k = 0, 1, averaged over 20 repetitions, where each method is tuned to have the optimal average MSE over a range of tuning parameter values.

Figure 8 reports these averaged MSE curves as functions of n, where the error bars again denote standard deviations. We can see that in both cases, the KTF error decays faster than the minimax rate (suggesting that the particular signals under consideration do not achieve the worst-case rate for KTF), while the linear method exhibits a perfectly flat error curve, suggesting that it fails to be consistent entirely.



Figure 5: Illustration of multivariate interpolation from Algorithm 2, when d = 2 and k = 2. The value to be interpolated is marked by a red star. The algorithm first interpolates the k + 1 = 3 values indicated by red dots, each time using univariate interpolation along the y-axis, from Algorithm 1. As the final step, these red dots are used to interpolate the value at the red star, using univariate interpolation along the x-axis, again from Algorithm 1.



Figure 6: Top left: KTF solution, with k = 1, for a problem over a square lattice with side length N = 25 (that is, $n = 25^2 = 625$ points). Top right: the interpolated surface from Algorithm 2 (itself evaluated over a grid with 4x the resolution along each dimension; that is, N = 100). Bottom row: analogous but for k = 2. In either case the interpolated function is in the tensor space of k^{th} order discrete splines.



Figure 7: MSE of various methods for estimating the signal function f_0 in Figure 1, over a square lattice with $n = 256^2 = 65536$ points, and subject to Gaussian noise to yield an SNR of 0.5. Left panel: the average performance (over 20 repetitions) of trend filtering estimators over a common range of tuning parameters λ . Right panel: the average performance of other nonparametric methods. Since their tuning parameters lie on different scales, the tuning parameter values are scaled to fit onto a single x-axis, and we notate the MSE-optimal tuning parameter value for each method. The takeaway is that KTF and GTF, in particular for k = 2, achieve clearly the best performance.

8.3 Rates for homogeneous signals

Lastly, we examine the empirical error rates of KTF and the eigenmaps projector for estimating homogeneously smooth signals. Recall that we showed the latter is minimax rate optimal over Sobolev (and Holder) classes in Theorem 6, whereas for s > 1/2, it is not clear (from its minimax rate over the inscribing KTV class) that KTF achieves the faster rate for Sobolev (or Holder) signals. We fix k = 0 and consider two values for the dimension, d = 2 and d = 3, which correspond to s = 1/2 and s > 1/2, respectively. The true mean θ_0 was taken to be the evaluations of a linear function over the lattice, scaled appropriately (according to the canonical scaling). Figure 9 reports the MSE curves from each method, under the same general setup as in the last subsection (averaged over 20 repetitions, optimal tuning per n). Moving from s = 1/2 to s > 1/2, we do not find that the empirical performance of TV denoising becomes markedly worse than its linear competitor, and the adaptivity of KTF to the smooth signals remains an open question.

9 Discussion

9.1 General lattice structures

The results and analysis in this paper have heretofore assumed an equally spaced lattice with the same number of knots in each coordinate direction, but as alluded to in the introduction, many of our results also apply to a more general lattice structure, namely a Cartesian product $Z_{N_1,...,N_d} = \{z_{i1}\}_{i=1}^{N_1} \times \{z_{i2}\}_{i=1}^{N_2} \times \cdots \times \{z_{id}\}_{i=1}^{N_d}$, with $n = \prod_{j=1}^d N_j$. With this general lattice, we generalize the $(k+1)^{\text{th-order}}$ discrete derivative matrix (8),

$$D_{z_1,\dots,z_d}^{(k+1)} = \begin{bmatrix} D_{z_1}^{(k+1)} \otimes I_{N_2} \otimes \dots \otimes I_{N_d} \\ I_{N_1} \otimes D_{z_2}^{(k+1)} \otimes \dots \otimes I_{N_d} \\ \vdots \\ I_{N_1} \otimes I_{N_2} \otimes \dots \otimes D_{z_d}^{(k+1)} \end{bmatrix},$$
(43)



Figure 8: Empirical error rates of KTF versus the eigenmaps projection estimator, for two cases: k = 0, d = 2, and k = 1, d = 4. The true mean in each case was chosen to be representative of a "hard" signal in the appropriate KTV class in (42). KTF converges faster than the minimax rate, whereas the eigenmaps estimator fails to be consistent entirely.

where $z_j := \{z_{ij}\}_{i=1}^{N_j}$ and D_{z_j} is a univariate discrete derivative matrix posed over that set of points. When the lattice points may be embedded into Euclidean space, i.e., the data have been rotated such that the elements of z_j are increasing and vary only in the *j*th coordinate, then $D_{z_j}^{(k+1)}$ is given (Tibshirani, 2014, 2020) by

$$D_{z_j}^{(k+1)} := D^{(1)} \cdot \operatorname{diag}\left(\frac{k}{z_{(k+1)j} - z_{1j}}, \frac{k}{z_{(k+2)j} - z_{2j}}, \cdots, \frac{k}{z_{N_j j} - z_{(N_j - k)j}}\right) \cdot D^{(k)},\tag{44}$$

where we have abused notation to only consider the j^{th} coordinate in the z_{ij} differences (as they are the only nonzero differences).

The KTF estimator with (43) satisfies Proposition 1, which states that the nullspace of (43) consists of polynomials with maximum degree k. Similarly, we obtain a version of Proposition 2, with appropriate adjustments for the number of basis elements oriented in each coordinate direction, and since KTF on a general lattice is still a generalized lasso problem, we inherit results on the degrees of freedom. Finally, we expect that under mild assumptions on the spacing of the points z_j in each direction j = 1, ..., d, one also obtains the discrete Holder class embedding of Proposition 3 and the estimation error bounds of Theorem 3.

9.2 Scattered data

The continuous-time representation of KTF, elucidated in Section 2.3, suggests an approach for fitting a KTF estimate to scattered data. Recall that Proposition 2 provides that the KTF problem (7) is equivalent to a variational problem over a function space spanned by tensor product of falling factorial basis functions, penalized by a kind of anisotropic total variation norm. Moreover, this "synthesis" viewpoint of KTF has a basis regression form made explicit in (13). Given scattered data $(x_1, y_1), \ldots, (x_n, y_n) \in \mathbb{R}^d \times \mathbb{R}$ and assuming a knot set $Z_{N_1,\ldots,N_d} = \{z_{i1}\}_{i=1}^{N_1} \times \{z_{i2}\}_{i=1}^{N_2} \times \cdots \times \{z_{id}\}_{i=1}^{N_d}$, we let

$$h_{z_i,x_i} := (h_{N_i,1}(x_{ij}), h_{N_i,2}(x_{ij}), \dots, h_{N_i,N_i}(x_{ij})) \in \mathbb{R}^{N_j}$$
(45)



Figure 9: Empirical error rates of KTF versus the eigenmaps projection estimator, for two cases: k = 0, d = 2, and k = 0, d = 3. The true mean was chosen to be a linear function in either case. KTF does not appear to be outperformed (in rate) by the eigenmaps estimator, leaving its adaptivity to smooth signals when s > 1/2 an open question.

be the evaluations of the marginal falling factorial bases (with knots given by $z_j = \{z_{ij}\}_{i=1}^{N_j}$; see Tibshirani (2020) for the unevenly spaced case) in coordinate direction j for the i^{th} observation. Then we solve

$$\underset{\alpha \in \mathbb{R}^{\prod_{j=1}^{d} N_{j}}}{\operatorname{minimize}} \frac{1}{2} \left\| y - \underbrace{\left| \begin{array}{c} h_{z_{1},x_{1}} \otimes h_{z_{2},x_{1}} \otimes \cdots \otimes h_{z_{d},x_{1}} \\ h_{z_{1},x_{2}} \otimes h_{z_{2},x_{2}} \otimes \cdots \otimes h_{z_{d},x_{2}} \\ \vdots \\ h_{z_{1},x_{n}} \otimes h_{z_{2},x_{n}} \otimes \cdots \otimes h_{z_{d},x_{n}} \end{array} \right|}_{=:\tilde{H}^{(k+1)}} \alpha \left\| \right\|_{2}^{2} + \lambda \left\| D_{z_{1},\dots,z_{d}}^{(k+1)} \left[H_{z_{1}}^{(k+1)} \otimes H_{z_{2}}^{(k+1)} \otimes \cdots H_{z_{d}}^{(k+1)} \right] \alpha \right\|_{1}, \quad (46)$$

and take $\tilde{H}^{(k+1)}\hat{\alpha}$ to be fitted values for a given λ , where $D_{z_1,...,z_d}^{(k+1)}$ is defined as in (43) and $H_{z_j}^{(k+1)}$ is the $(k+1)^{\text{th}}$ order falling factorial basis with knots given by z_j . Since the scattered data do not immediately imply an appropriate lattice of knots with which to form the estimator (46), we propose a simple scheme, absent countervailing considerations. Take $N_1 = \cdots = N_d = N := \lfloor n^{1/d} \rfloor$, and project the data $\{x_i\}_{i=1}^n$ onto each of the *d* coordinate axes to obtain $\{p_{ij}\}_{i=1}^n \subset \mathbb{R}, j = 1, \ldots, d$. Then, for each *j*, let the knots $\{z_{ij}\}_{i=1}^N$ be the $1/(N+1), 2/(N+1), \ldots, N/(N+1)$ quantiles of $\{p_{ij}\}_{i=1}^n$.

9.3 Generalized linear models

The KTF problem may be further extended through the replacement of the squared loss in (7) with a generalized linear model (GLM) loss. Specifically, in lieu of (7), we solve

$$\underset{\theta \in \mathbb{R}^n}{\operatorname{minimize}} \quad -\theta^\top y + A(\theta) + \lambda \| D_{n,d}^{(k+1)} \theta \|_1,$$
(47)

where y_i is assumed to be distributed according to an exponential family distribution with canonical parameter θ and log-partition function $A(\theta)$. This reduces to the squared loss when y_i follows a Gaussian distribution with mean μ_i and variance σ^2 . A common application of the GLM loss framework is density estimation via Poisson regression. Supposing observations from a distribution supported on a hyperrectangle, one discretizes the domain using a grid and bins the counts. The centers of the bins may then be treated as lattice points and their corresponding counts modeled using a KTF GLM where the response is a Poisson mean. Specifically, letting y be a vector of binned counts and θ be the canonical parameter, we solve

$$\underset{\theta \in \mathbb{R}^n}{\operatorname{minimize}} \quad -\theta^\top y + \mathbb{1}^\top \exp \theta + \lambda \| D_{n,d}^{(k+1)} \theta \|_1.$$
(48)

This learns a spatially dependent Poisson mean that varies smoothly over the domain, with the notion of smoothness dictated by $D_{n,d}^{(k+1)}$. We note that this has previously been studied in univariate and a graph fused lasso setting in Padilla and Scott (2016); Bassett and Sharpnack (2019); this proposal would extend the approach to multiple dimensions and smoother signals. The theory for GLMs is known to be more complicated (see, e.g., Haris et al., 2019), but some simple truncation arguments may be a viable approach (e.g., Lin et al., 2017).

9.4 Mixed discrete derivatives

The KTF penalty defined via (8) applies an ℓ_1 penalty to the $(k + 1)^{\text{th}}$ discrete derivatives, marginally along each coordinate direction. One could instead penalize mixed discrete derivatives of maximum order k + 1, in which case the nullspace of the penalty operator is polynomials of *total* degree k rather than those of *maximum* degree k. The tradeoff is that this is computationally more difficult since the penalty matrix becomes more complex. In particular, the penalty for a $\theta \in \mathbb{R}^n$ is given by

$$\sum_{\alpha:\alpha^{\mathsf{T}}1=k+1,\alpha\geq 0} \sum_{x\in Z_{n,d}} \left(\Delta_{x_1^{\alpha_1}\dots x_d^{\alpha_d}}\theta\right)(x)$$

in the notation of Section 1.2. It would be interesting to verify that this penalty formulation matches with the obvious continuous equivalent, in the sense of Proposition 2 for KTF penalty. Error analysis for this method is also left to future work.

Different smoothness levels across dimensions. We can naturally extend KTF in (6) to handle differing orders of smoothness across dimensions:

$$\underset{\theta \in \mathbb{R}^n}{\text{minimize}} \quad \frac{1}{2} \sum_{i=1}^n \left(y_i - \theta(x_i) \right)^2 + \sum_{j=1}^d \lambda_j \sum_{x \in Z_{n,d}} |(\Delta_{x_j^{k_j+1}} \theta)(x)|.$$

$$\tag{49}$$

with d tuning parameters $\lambda_1, \ldots, \lambda_d$. This estimator satisfies key properties in Section 2 and the optimization and interpolation algorithms in Sections 6, 7 apply to this method with appropriate change in notation. We believe estimation error results similar in spirit to those in Section 5 will hold, but we haven't pursued the analysis. Two key pre-conditions to apply Theorem 2 are incoherence of eigenvectors of DD^{T} and tight bounds on the summation $\sum_i \xi_i^{-2}$. Incoherence is easy to verify for the difference operator in (49) but we leave bounding the summation to future work.

A Proofs

A.1 Proof of Proposition 1

Abbreviating $D = D_{n,d}^{(k+1)}$, the null space of D is the number of its nonzero singular values or equivalently, the number of nonzero eigenvalues of $D^{\mathsf{T}}D$. Following from (8), and abbreviating $Q = D_N^{(k+1)}$ and $I = I_N$,

$$D^{\mathsf{T}}D = Q^{\mathsf{T}}Q \otimes I \otimes \cdots \otimes I + I \otimes Q^{\mathsf{T}}Q \otimes \cdots \otimes I + \ldots + I \otimes I \otimes \cdots \otimes Q^{\mathsf{T}}Q,$$

the Kronecker sum of $Q^{\mathsf{T}}Q$ with itself, a total of d times. Using a standard fact about Kronecker sums, if we denote by ρ_i , i = 1, ..., N the eigenvalues of $Q^{\mathsf{T}}Q$ then

$$\rho_{i_1} + \rho_{i_2} + \dots + \rho_{i_d}, \quad i_1, \dots, i_d \in \{1, \dots, N\}$$

are the eigenvalues of $D^{\mathsf{T}}D$. By counting the multiplicity of the zero eigenvalue, we arrive at a nullity for D of $(k+1)^d$. It is straightforward to check that the vectors specified in the proposition, given by evaluations of polynomials of max degree k, are in the null space, and that these are linearly independent, which completes the proof.

A.2 **Proof of Proposition 2**

Let us define

$$B_N^{(k+1)} = \begin{bmatrix} C_N^{(k+1)} \\ D_N^{(k+1)} \end{bmatrix} \in \mathbb{R}^{N \times N}$$

where the first k + 1 rows are given by a matrix $C_N^{(k+1)} \in \mathbb{R}^{(k+1) \times N}$ that completes the row space, as in Lemma 2 of Wang et al. (2014), or Section 6.2 of Tibshirani (2020). Now, again by Lemma 2 of Wang et al. (2014), or Section 6.3 of Tibshirani (2020),

$$\left(H_N^{(k+1)}\right)^{-1} = \frac{1}{k!} B_N^{(k+1)} \tag{50}$$

where $H_N^{(k+1)} \in \mathbb{R}^{N \times N}$ is the falling factorial basis matrix of order k, which has elements

$$\left[H_N^{(k+1)}\right]_{ij} = h_{N,j}^k(i/N), \quad i, j = 1, \dots, N,$$

with $h_{N,i}^k$, i = 1, ..., N denoting the falling factorial functions in (10) with respect to design points 1/N, 2/N, ..., 1. We now transform variables in (7) by defining

$$\theta = \left(H_N^{(k+1)} \otimes \cdots \otimes H_N^{(k+1)} \right) \alpha,$$

and using (50), this turns (7) into an equivalent basis form,

$$\underset{\alpha \in \mathbb{R}^{n}}{\operatorname{minimize}} \quad \frac{1}{2} \left\| y - \left(H_{N}^{(k+1)} \otimes \cdots \otimes H_{N}^{(k+1)} \right) \alpha \right\|_{2}^{2} + \lambda k! \left\| \left(\begin{array}{c} I_{N}^{0} \otimes H_{N}^{(k+1)} \otimes \cdots \otimes H_{N}^{(k+1)} \\ H_{N}^{(k+1)} \otimes I_{N}^{0} \otimes \cdots \otimes H_{N}^{(k+1)} \\ \vdots \\ H_{N}^{(k+1)} \otimes H_{N}^{(k+1)} \otimes \cdots \otimes I_{N}^{0} \end{array} \right\|_{1}^{\alpha} \right\|_{1}$$

where $I_N^0 \in \mathbb{R}^{(N-k-1)\times N}$ denotes the last N-k-1 rows of the identity I_N . We can rewrite the problem once more by parametrizing the evaluations according to f in (11), which we claim yields (12). The equivalence between loss terms in the above problem and (12) is immediate (by definition of f); to see the equivalence between penalty terms, it can be directly checked that

$$k! \Big(I_N^0 \otimes H_N^{(k+1)} \otimes \cdots \otimes H_N^{(k+1)} \Big) \alpha$$

contains the differences of the function $\partial^k f / \partial x_1^k$ over all pairs of grid positions that are adjacent in the x_1 direction, where f is as in (11). This, combined with the fact that $\partial^k f / \partial x_1^k$ is constant in between lattice positions, means that

$$k! \left\| \left(I_N^0 \otimes H_N^{(k+1)} \otimes \cdots \otimes H_N^{(k+1)} \right) \alpha \right\|_1 = \sum_{x_{-1}} \mathrm{TV}\left(\frac{\partial^k f(\cdot, x_{-1})}{\partial x_1^k} \right),$$

the total variation of $\partial^k f / \partial x_1^k$ added up over all slices of the lattice $Z_{n,d}$ in the x_1 direction. Similar arguments apply to the penalty terms corresponding to dimensions j = 2, ..., d, and this completes the proof.

A.3 Proof of Theorem 1

Denote by $f_{\epsilon} = \eta_{\epsilon} * f$ the mollified version of f, where $\eta_{\epsilon}(x) = \epsilon^{-d} \eta(x/\epsilon)$, and $\eta : \mathbb{R}^d \to \mathbb{R}$ is the standard mollifier, defined by

$$\eta(x) = \begin{cases} c \exp\left(\frac{1}{\|x\|_2^2 - 1}\right) & \text{if } \|x\|_2 \le 1, \\ 0 & \text{else}, \end{cases}$$

and c > 0 is a normalization constant so that η integrates to 1. By construction, for any $\epsilon > 0$, we have $f_{\epsilon} \in C^{\infty}(U)$. Now from (18), simply exchanging the sum over absolute partial derivatives and the integral, we have

$$\begin{aligned} \mathrm{TV}(f_{\epsilon};U) &= \sum_{j=1}^{d} \int_{U} \left| \frac{\partial f_{\epsilon}(x)}{\partial x_{j}} \right| dx \\ &= \sum_{j=1}^{d} \int_{U_{-j}} \int_{I_{x_{-j}}} \left| \frac{\partial f_{\epsilon}(x_{j}, x_{-j})}{\partial x_{j}} \right| dx_{j} dx_{-j} \\ &= \sum_{j=1}^{d} \int_{U_{-j}} \int_{I_{x_{-j}}} \left| \frac{\partial f_{\epsilon}(x_{j}, x_{-j})}{\partial x_{j}} \right| dx_{j} dx_{-j} \\ &= \sum_{j=1}^{d} \int_{U_{-j}} \mathrm{TV}(f_{\epsilon}(\cdot, x_{-j}); I_{x_{-j}}) dx_{-j}, \end{aligned}$$

where in the last line we applied the representation of TV for smooth functions in (18), but to the univariate function $x_j \mapsto f_{\epsilon}(x_j, x_{-j})$, for fixed x_{-j} . Recalling standard results on approximation of BV functions by smooth functions (see, for example, Theorem 5.22 in Evans and Gariepy (2015)), by sending $\epsilon \to 0$, we have that the left-most and right-most sides of the previous display approach those in (20), completing the proof.

A.4 Proof of Theorem 3

Abbreviate N' = N - k - 1. Let β_i, u_i, v_i be a triplet of nonzero singular value, left singular vector, and right singular vector of $D_{N,1}^{(k+1)}$, for $i \in [N']$ and let $p_j, j \in [k+1]$ form an orthogonal basis for the null space of $D_{N,1}^{(k+1)}$. From Lemma 1 it suffices to show incoherence of $u_i, v_i, i \in [N']$, and $p_i, i \in [k+1]$. Incoherence of $u_i, i \in [N']$ and $v_i, i \in [N']$ is established in Sadhanala et al. (2017). Incoherence of $p_i, i \in [k+1]$ may be seen by choosing, e.g., these vectors to be the discrete Legendre orthogonal polynomials as in Neuman and Schonbach (1974). Applying Lemma 1, we can see that $D_{n,d}^{(k+1)}$ satisfies the incoherence property, as defined in Theorem 2.

From the incoherence property and Theorem 2, the KTF estimator $\hat{\theta}$, satisfies

$$\frac{1}{n} \|\hat{\theta} - \theta_0\|_2^2 = O_{\mathbb{P}} \left(\frac{\kappa}{n} + \frac{|I|}{n} + \frac{\mu}{n} \sqrt{\frac{\log n}{n} \sum_{i \in [N]^d \setminus (I \cup [k+1]^d)} \frac{1}{\xi_i^2}} \cdot \|\Delta \theta_0\|_1 \right),\tag{51}$$

where we abbreviate $\Delta = D_{n,d}^{(k+1)}$, $\xi_i, i \in [N]^d$ are eigenvalues of $\Delta^T \Delta$ with $\xi_i = 0$ for $i \in [k+1]^d$. We reindexed the eigenvalues so that they correspond to grid positions.

Recall the shorthand s = (k+1)/d. For $s \le 1/2$, set $I = [k+2]^d \setminus [k+1]^d$. From Lemma 6,

$$\sum_{i \in [N]^d \setminus (I \cup [k+1])^d} \frac{1}{\xi_i^2} \le c \begin{cases} n & s < 1/2\\ n \log n & s = 1/2. \end{cases}$$

Plugging this into (51) gives the desired bounds when $s \le 1/2$:

$$\frac{1}{n} \|\hat{\theta} - \theta_0\|_2^2 = \begin{cases} O_{\mathbb{P}} \left(\frac{(k+2)^d}{n} + \|\Delta\theta_0\|_1 \sqrt{\log n} \right) & s < 1/2 \\ O_{\mathbb{P}} \left(\frac{(k+2)^d}{n} + \|\Delta\theta_0\|_1 \log n \right) & s = 1/2. \end{cases}$$

Now consider the case s > 1/2. Set $I = \{i \in [N]^d : ||(i-k-2)_+||_2 < r\} \setminus [k+1]^d$ for an $r \in [1, \sqrt{dN}]$ to be chosen later. $|I| \le (r+k+1)^d$ because $I \subseteq [\lfloor r \rfloor + k + 1]^d$. Lemma 6 shows that for a constant c > 0 depending only on k, d,

$$\sum_{i \in [N]^d \setminus (I \cup [k+1]^d)} \frac{1}{\xi_i^2} \le cn^{2s} / r^{d(2s-1)}$$

Plug this bound in (51) and in order to minimize the resulting bound, choose r to balance

$$(r+k+1)^d$$
 with $\frac{C_n}{\sqrt{n}}\sqrt{n^{2s}/r^{d(2s-1)}\log n}.$

This leads us to take

$$(r+k+1)^d \simeq (C_n \sqrt{\log n})^{\frac{2}{2s+1}} n^{\frac{2s-1}{2s+1}}$$

when $C_n \sqrt{\log n}/n = O_{\mathbb{P}}(1)$ and r = 1 otherwise. With this choice (51) gives the desired bound for s > 1/2. This completes the proof.

A.5 Incoherence of Kronecker product type operators

Let

$$\Delta = \begin{bmatrix} D \otimes I \otimes \dots \otimes I \\ I \otimes D \otimes \dots \otimes I \\ \vdots \\ I \otimes I \otimes \dots \otimes D \end{bmatrix}$$
(52)

where each Kronecker product has d terms. With $D = D_{N,1}^{(k+1)}$, we get the KTF penalty operator $\Delta = D_{n,d}^{(k+1)}$.

Lemma 1. Let Δ be as defined in (52) for a matrix $D \in \mathbb{R}^{N' \times N}$ with $N' \leq N$. Let $\gamma_i, u_i, v_i, i \in [N]$ denote the singular values of D, its left and right singular vectors. Note that $\gamma_i = 0, u_i = 0, v_i \in \text{null}(D)$ for $i \in [p]$ where p = nullity(D). If these singular vectors are incoherent, that is $||v_i||_{\infty} \leq \mu/\sqrt{N}, ||u_i||_{\infty} \leq \mu/\sqrt{N'}$ for a constant $\mu \geq 1$, then the left singular vectors ν of Δ are incoherent with a constant μ^d , that is, $||\nu||_{\infty} \leq \mu^d/\sqrt{N^{d-1}N'}$.

Note that p = k + 1 when Δ is the KTF penalty operator with $D = D_{N,1}^{(k+1)}$.

Proof of Lemma 1. Abbreviate $\rho_i = \gamma_i^2$ for $i \in [N]$. We are looking for a total of $N^d - p^d$ eigenvectors for $\Delta \Delta^{\mathsf{T}}$. Assume for exposition that d = 3. For any $(i, j, k) \in [N]^d \setminus [p]^d$ (where \setminus is the set difference operator), the vectors

$$\nu_{i,j,k} := \frac{1}{\sqrt{\rho_i + \rho_j + \rho_k}} \begin{bmatrix} \gamma_i \cdot u_i \otimes v_j \otimes v_k \\ \gamma_j \cdot v_i \otimes u_j \otimes v_k \\ \gamma_k \cdot v_i \otimes v_j \otimes u_k \end{bmatrix}$$
(53)

are eigenvectors of $\Delta \Delta^{\mathsf{T}}$ as verified below.

$$\Delta \Delta^{\mathsf{T}} \begin{bmatrix} \gamma_{i} \cdot u_{i} \otimes v_{j} \otimes v_{k} \\ \gamma_{j} \cdot v_{i} \otimes u_{j} \otimes v_{k} \\ \gamma_{k} \cdot v_{i} \otimes v_{j} \otimes u_{k} \end{bmatrix} = \Delta \left(\gamma_{i}^{2} + \gamma_{j}^{2} + \gamma_{k}^{2} \right) v_{i} \otimes v_{j} \otimes v_{k}$$

$$= \left(\rho_{i} + \rho_{j} + \rho_{k} \right) \begin{bmatrix} \gamma_{i} \cdot u_{i} \otimes v_{j} \otimes v_{k} \\ \gamma_{j} \cdot v_{i} \otimes u_{j} \otimes v_{k} \\ \gamma_{k} \cdot v_{i} \otimes v_{j} \otimes u_{k} \end{bmatrix}$$
(54)

We see all $N^d - p^d$ eigenvectors of $\Delta \Delta^T$ here. Notice that $||z_{i,j,k}||_2 = 1$ and the incoherence is readily available given that the left and right singular vectors of D are incoherent.

For general d, these $N^d - p^d$ eigenvectors are given by

$$\nu_{i_1,i_2,.,i_d} = \frac{1}{\sqrt{\sum_{j=1}^d \rho_{i_j}}} \begin{bmatrix} \gamma_{i_1} \cdot u_{i_1} \otimes v_{i_2} \otimes \dots v_{i_d} \\ \gamma_{i_2} \cdot v_{i_1} \otimes u_{i_2} \otimes \dots v_{i_d} \\ \vdots \\ \gamma_{i_d} \cdot v_{i_1} \otimes v_{i_2} \otimes \dots u_{i_d} \end{bmatrix}$$
(55)

with eigenvalues $\sum_{j=1}^{d} \rho_{i_j}$ and are easily seen to be incoherent.

A.6 Upper bound for continuous KTV class

Recalling the continuous analog of KTF penalty from (12), define the class

$$\mathrm{KTV}_{n,d}^k(C) = \left\{ f: \sum_{j=1}^d \sum_{x_{-j}} \mathrm{TV}\left(\frac{\partial^k f(\cdot, x_{-j})}{\partial x_j^k}\right) \le C \right\}$$

for C > 0. If the true signal θ_0 on the grid is an evaluation of a function $f \in \operatorname{KTV}_k^d(C)$, the rates in Theorem 3 hold with C_n replaced by C, due to the following result.

Lemma 2. Let C > 0 and let $d \ge 1, k \ge 0$ be integers. For all $f \in \operatorname{KTV}_{n,d}^k(C)$, if $\theta_f \in \mathbb{R}^n$ is the evaluation of f on the grid points $Z_{n,d}$, then

$$\|D_{n,d}^{(k+1)}\theta_f\|_1 \le c_1 C$$

for a constant c_1 that depends only on k and d.

Proof of Lemma 2. Let f be an arbitrary function from $\text{KTV}_{n,d}^k(C)$. Pick a $j \in [N]$ and an x_{-j} and consider the function $\phi(\cdot) = f(\cdot, x_{-j})$ (f with all but its j argument fixed to elements of x_{-j} appropriately in order). From Theorem 1 in Mammen (1991) and its proof, there exists a spline $\tilde{\phi}$ such that

$$\tilde{\phi}(i/N) = \phi(i/N), \quad i \in [N]$$

$$\Gamma \mathcal{V}(\tilde{\phi}^{(k)}) \le \mathrm{TV}(\phi^{(k)})$$

Let t_1, \ldots, t_L be the knots of $\tilde{\phi}$, which are not necessarily in the set of input points. Because it is a spline, $\tilde{\phi}$ can be written as the sum of a polynomial and a linear combination of kth degree truncated power basis functions $g_t : x \mapsto (x - t)_+^k / k!$

$$\tilde{\phi}(u) = p(u) + \sum_{\ell=1}^{L} \beta_{\ell} g_{t_{\ell}}(u), \quad u \in [0, 1]$$

where p is a polynomial of degree $\leq k$ and $\beta_{\ell} \in \mathbb{R}, \ell \in [L]$. Let $D_{1d}^{(k+1)} = D_{N,1}^{(k+1)}$. Now

$$\begin{split} \left\| D_{1d}^{(k+1)} \begin{bmatrix} \phi(1/N) \\ \vdots \\ \phi(N/N) \end{bmatrix} \right\|_{1} &= \left\| D_{1d}^{(k+1)} \begin{bmatrix} \tilde{\phi}(1/N) \\ \vdots \\ \tilde{\phi}(N/N) \end{bmatrix} \right\|_{1} \\ &= \left\| D_{1d}^{(k+1)} \begin{bmatrix} p(1/N) \\ \vdots \\ p(N/N) \end{bmatrix} + \sum_{\ell=1}^{L} \beta_{\ell} \cdot D_{1d}^{(k+1)} \begin{bmatrix} g_{t_{\ell}}(1/N) \\ \vdots \\ g_{t_{\ell}}(N/N) \end{bmatrix} \right\|_{1} \\ &\leq \sum_{\ell=1}^{L} |\beta_{\ell}| \| D_{1d}^{(k+1)} G_{\ell}^{(k)} \|_{1} \end{split}$$
(56)

where the vector $G_{\ell}^{(k)}$ is the evaluation of $g_{t_{\ell}}$ on $1/N, \ldots, N/N$, that is $(G_{\ell}^{(k)})_i = g_{t_{\ell}}(i/N), i \in [N]$. Here we used the fact that $D_{1d}^{(k+1)}$ times the evaluations of a polynomial at the input points $1/N, \ldots, N/N$ is 0. The terms in (56) can be bound as follows. For $\ell \in [L]$, let $i_{\ell} = \max_{i \in [N]} \{i/N \leq t_{\ell}\}$, that is, let i_{ℓ}/N be the

The terms in (56) can be bound as follows. For $\ell \in [L]$, let $i_{\ell} = \max_{i \in [N]} \{i/N \leq t_{\ell}\}$, that is, let i_{ℓ}/N be the largest input point that is not greater the knot t_{ℓ} . For any vector $v \in \mathbb{R}^N$, and $i \in [N - k - 1]$, $(D_{1d}^{(k+1)}v)_i = 0$ if (v_i, \ldots, v_{i+k}) is the evaluation of a polynomial at $i/N, \ldots, (i + k + 1)/N$. $g_{t_{\ell}}$ is a polynomial on $[0, t_{\ell}]$ and on $[t_{\ell}, 1]$ for $\ell \in [L]$. Therefore, $D_{1d}^{(k+1)}G_{\ell}^{(k)}$ is nonzero in at most k + 1 elements. Letting A_i denote the *i*th row of matrix A, we can write

$$\begin{split} |D_{1d}^{(k+1)}G_{\ell}^{(k)}||_{1} &= \sum_{i=1}^{N-k-1} \left| \left(D_{1d}^{(k+1)} \right)_{i}G_{\ell}^{(k)} \right| \\ &= \sum_{i=(i_{\ell}-k)\vee 1}^{i_{\ell}} \left| \left(D_{1d}^{(k+1)} \right)_{i}G_{\ell}^{(k)} \right| \\ &\leq \sum_{i=(i_{\ell}-k)\vee 1}^{i_{\ell}} \left\| \left(D_{1d}^{(k+1)} \right)_{i} \right\|_{1}g_{t_{\ell}} \left(\frac{i_{\ell}+k+1}{N} \right) \\ &\leq \sum_{i=(i_{\ell}-k)\vee 1}^{i_{\ell}} \left\| \left(D_{1d}^{(k+1)} \right)_{i} \right\|_{1} \left(\frac{k+1}{N} \right)^{k} \frac{1}{k!} \\ &\leq (k+1) \cdot \left(\frac{k+1}{N} \right)^{k} \frac{1}{k!} \max_{i \in [N-k-1]} \left\| \left(D_{1d}^{(k+1)} \right)_{i} \right\|_{1} \\ &= (k+1) \cdot \left(\frac{k+1}{N} \right)^{k} \frac{1}{k!} \cdot 2^{k+1} N^{k} \\ &= b_{k} \end{split}$$

where b_k is a constant depending only on k. Plugging this upper bound in (56),

$$\left\| D_{1d}^{(k+1)} \begin{bmatrix} \phi(1/N) \\ \vdots \\ \phi(N/N) \end{bmatrix} \right\|_1 \le b_k \sum_{\ell=1}^L |\beta_\ell| = b_k \operatorname{TV}(\tilde{\phi}^{(k)}) \le b_k \operatorname{TV}(\phi^{(k)}).$$

This means,

$$\|D_{n,d}^{(k+1)}\theta_f\|_1 = \sum_{j=1}^d \sum_{x_{-j}} \left\|D_{1d}^{(k+1)} \begin{bmatrix} f(1/N, x_{-j}) \\ \vdots \\ f(N/N, x_{-j}) \end{bmatrix}\right\|_1$$
$$\leq \sum_{j=1}^d \sum_{x_{-j}} b_k \operatorname{TV}\left(\frac{\partial^k f(\cdot, x_{-j})}{\partial x_j^k}\right)$$
$$\leq b_k \cdot C.$$

This completes the proof.

A.7 Proof of Theorem 4

Here and henceforth, we use the notation $B_p(r) = \{x : ||x||_p \le r\}$ for the ℓ_p ball of radius r, where p, r > 0 (and the ambient dimension will be determined based on the context).

Lemma 3 (Lemma 7 in Sadhanala et al. (2016)). Let $\mathcal{T}(r) = \{\theta \in \mathbb{R}^n : \|D\theta\|_1 \leq r\}$ for a matrix D and r > 0. Recall that $\|D\|_{1,\infty} = \max_{i \in [n]} \|D_i\|_1$ where D_i is the *i*th column of D. Then for any r > 0, it holds that $B_1(r/\|D\|_{1,\infty}) \subseteq \mathcal{T}(r)$.

From Lemma 3 and the fact that $\|D_{n,d}^{(k+1)}\|_{1,\infty} = 2^{k+1}d$

$$B_1(r/(2^{k+1}d)) \subseteq \mathcal{T}_{n,d}^k(r).$$
(57)

for any r > 0, and integers $d \ge 1, k \ge 0$.

To prove Theorem 4 we will use the following result from Birge and Massart (2001), which gives a lower bound for the risk in a normal means problem, over ℓ_p balls. We state the result in our notation.

Lemma 4 (Proposition 5 of Birge and Massart (2001)). Assume i.i.d. observations $y_i \sim N(\theta_{0,i}, \sigma^2)$, i = 1, ..., n, and $n \geq 2$. Then the minimax risk over the ℓ_p ball $B_p(r_n)$, where 0 , satisfies

$$n \cdot R(B_p(r_n)) \ge c \cdot \begin{cases} \sigma^{2-p} r_n^p \left[1 + \log\left(\frac{\sigma^p n}{r_n^p}\right) \right]^{1-p/2} & \text{if } \sigma \sqrt{\log n} \le r_n \le \sigma n^{1/p} / \sqrt{\rho_p} \\ r_n^2 & \text{if } r_n < \sigma \sqrt{\log n} \\ \sigma^2 n / \rho_p & \text{if } r_n > \sigma n^{1/p} / \sqrt{\rho} \end{cases}$$

Here c > 0 *is a universal constant, and* $\rho_p > 1.76$ *is the unique solution of* $\rho_p \log \rho_p = 2/p$ *.*

Proof of Theorem 4. It suffices to show that the minimax optimal risk $R(\mathcal{T}_{n,d}^k(C_n))$ is lower bounded by the three terms present in the statement's lower bound separately:

$$R(\mathcal{T}_{n,d}^{k}(C_{n})) = \Omega\left(\frac{\kappa\sigma^{2}}{n}\right),$$

$$R(\mathcal{T}_{n,d}^{k}(C_{n})) = \Omega\left(\frac{\sigma C_{n}}{n} \wedge \sigma^{2}\right),$$

$$R(\mathcal{T}_{n,d}^{k}(C_{n})) = \Omega\left(\left(\frac{C_{n}}{n}\right)^{\frac{2}{2s+1}} \sigma^{\frac{4s}{2s+1}} \wedge \sigma^{2}\right),$$
(58)

where $\kappa = \text{nullity}(D_{n,d}^{(k+1)}) = (k+1)^d$. First, as the null space of $D_{n,d}^{(k+1)}$ has dimension κ , we get the first lower bound:

$$\inf_{\hat{\theta}} \sup_{\theta_0 \in \mathcal{T}_{n,d}^k(C_n)} \frac{1}{n} \mathbb{E} \| \hat{\theta} - \theta_0 \|_2^2 \geq \inf_{\hat{\theta}} \sup_{\theta_0 \in \operatorname{null}\left(D_{n,d}^{(k+1)}\right)} \frac{1}{n} \mathbb{E} \| \hat{\theta} - \theta_0 \|_2^2 \geq \frac{\kappa \sigma^2}{n}$$

We get the second lower bound in (58) by using the ℓ_1 -ball embedding

$$B_1(C_n/d_{\max}) \subset \mathcal{T}_{n,d}^k(C_n)$$

from (57) and then using Lemma 4. Finally, from Theorem 4 in Sadhanala et al. (2017), it follows that

$$R(\mathcal{C}_{n,d}^k(L_n)) = \Omega\left(\left(\frac{\sigma^2}{n}\right)^{\frac{2s}{2s+1}} L_n^{\frac{2}{2s+1}} \wedge \sigma^2\right)$$
(59)

with additional tracking for σ^2 . Taking $L_n = C_n/n^{1-s}$ and applying Proposition 3 would then give the third lower bound in (58). This completes the proof.

A.8 **Proof of Theorem 5 (Minimax linear rate)**

We use the following shorthand for the risk of an estimator $\hat{\theta}$ over a class \mathcal{K} :

$$\operatorname{Risk}(\hat{\theta}) = \sup_{\theta_0 \in \mathcal{K}} \frac{1}{n} \mathbb{E} \|\hat{\theta} - \theta_0\|_2^2.$$

For a matrix $S \in \mathbb{R}^{n \times n}$ let $\operatorname{Risk}(S)$ also denote the risk of the linear smoother $\hat{\theta} = Sy$.

Proof of Theorem 5. For brevity, denote $D = D_{n,d}^{(k+1)}$ and let S stand for a linear smoother in the context of this proof. The minimax linear risk for the class $\mathcal{T}_{n,d}^k(C_n)$ is

$$R_L(\mathcal{T}_{n,d}^k(C_n)) = \inf_{S \in \mathbb{R}^{n \times n}} \sup_{\theta_0 \in \mathcal{T}_{n,d}^k(C_n)} \frac{1}{n} \mathbb{E} \|Sy - \theta_0\|_2^2$$
$$= \inf_{S} \sup_{\theta_0 \in \mathcal{T}_{n,d}^k(C_n)} \frac{1}{n} \mathbb{E} \|S(\theta_0 + \epsilon) - \theta_0\|_2^2$$
$$= \frac{1}{n} \inf_{S} \sup_{\theta_0 \in \mathcal{T}_{n,d}^k(C_n)} \sigma^2 \|S\|_F^2 + \|(S - I)\theta_0\|_2^2$$

where in the last line we used the assumption that $\epsilon_i, i \in [n]$ are i.i.d. with mean zero and variance σ^2 and used the notation $||A||_F$ for the Frobenius norm of a matrix A. The infimum can be restricted to the set of linear smoothers

$$\mathbb{S} = \{S : \operatorname{null}(S - I) \supseteq \operatorname{null}(D)\}$$

because if for a linear smoother S, if there exists $\eta \in \text{null}(D)$ such that $(S - I)\eta \neq 0$, then the inner supremum above will be ∞ , that is, its risk will be ∞ . If the outer infimum is over S, then the supremum can be restricted to $\{\theta_0 \in \text{row}(D) : \theta \in \mathcal{T}_{n,d}^k(C_n)\}$. We continue to lower bound minimax linear risk as follows:

$$R_{L}(\mathcal{T}_{n,d}^{k}(C_{n})) = \frac{1}{n} \inf_{S \in \mathbb{S}} \sigma^{2} \|S\|_{F}^{2} + \sup_{\theta_{0} \in \operatorname{row}(D): \|D\theta_{0}\|_{1} \leq C_{n}} \|(S-I)\theta_{0}\|_{2}^{2}$$

$$= \frac{1}{n} \inf_{S \in \mathbb{S}} \sigma^{2} \|S\|_{F}^{2} + \sup_{z: \|z\|_{1} \leq C_{n}} \|(S-I)D^{+}z\|_{2}^{2}$$

$$= \frac{1}{n} \inf_{S \in \mathbb{S}} \sigma^{2} \|S\|_{F}^{2} + C_{n}^{2} \max_{i \in [m]} \|((S-I)D^{+})_{i}\|_{2}^{2}$$

$$\geq \frac{1}{n} \inf_{S \in \mathbb{S}} \sigma^{2} \|S\|_{F}^{2} + \frac{C_{n}^{2}}{m} \sum_{i=1}^{m} \|((S-I)D^{+})_{i}\|_{2}^{2}$$

$$= \inf_{S \in \mathbb{S}} \underbrace{\frac{\sigma^{2}}{n} \|S\|_{F}^{2} + \frac{C_{n}^{2}}{mn} \|(S-I)D^{+}\|_{F}^{2}}_{=:r(S)}$$
(61)

In the third line, $(A)_i$ denotes the *i*th column of matrix A and m denotes the number of rows in D. In the fourth line, we used the fact that the maximum of a set is at least as much as their average. In the last line — within the context of this proof — we define the quantity r(S) which is a lower bound on the risk of a linear smoother $S \in S$.

Notice that $r(\cdot)$ is a quadratic in the entries of S and the constraint $S \in S$ translates to linear constraints on the entries of S. Writing the KKT conditions, after some work, we see that $r(\cdot)$ is minimized at

$$S_0 = a_n \left(\sigma^2 L^{(k+1)} + a_n I \right)^{-1}$$
(62)

where we denote $a_n = \frac{C_n^2}{m}$ and $L^{(k+1)} = D^{\mathsf{T}}D$ (the inverse is well defined because $a_n > 0$). Further, $S_0 \in \mathbb{S}$. Therefore,

$$R_L(\mathcal{T}_{n,d}^k(C_n)) \ge r(S_0). \tag{63}$$

We simplify the expression for $r(S_0)$ now. Let $\lambda_i, i \in [n]$ be the eigenvalues of $L^{(k+1)}$. Then the eigenvalues of S_0 are

$$\frac{a_n}{\sigma^2 \xi_i + a_n}, i \in [n]$$

and the non-zero squared singular values of $(S_0 - I)D^+$ are given by

$$\frac{\sigma^4 \xi_i}{(\sigma^2 \xi_i + a_n)^2}, \quad \kappa < i \le n.$$

Using the fact that the squared Frobenius norm of a matrix is the sum of squares of its singular values, substituting the above eigenvalues and singular values in (61), we have

$$r(S_0) = \frac{\sigma^2}{n} \sum_{i=1}^n \left(\frac{a_n}{\sigma^2 \xi_i + a_n}\right)^2 + \frac{a_n}{n} \sum_{i=1}^n \frac{\sigma^4 \xi_i}{(\sigma^2 \xi_i + a_n)^2} \\ = \frac{1}{n} \sum_{i=1}^n \frac{\sigma^2 a_n}{\sigma^2 \xi_i + a_n}.$$
(64)

Now we upper bound the risk $Risk(S_0)$ of the linear smoother defined by S_0 . From (60), we can write

$$\operatorname{Risk}(S_0) = \frac{\sigma^2}{n} \|S_0\|_F^2 + \frac{C_n^2}{n} \max_{i \in [m]} \left\| \left((S_0 - I)D^+ \right)_i \right\|_2^2.$$

Let $D = U\Sigma V^{\mathsf{T}}$ be the singular value decomposition of D. Also let the eigen-decomposition of $S_0 - I = V\Lambda V^{\mathsf{T}}$. Then using incoherence of columns of U, that is, the fact that there exists a constant c > 1 that depends only on k, d such that $U_{ij}^2 \leq \frac{c}{m}$ for all $i \in [m], j \in [n]$, we can write

$$\max_{i \in [m]} \left\| \left((S_0 - I)D^+ \right)_i \right\|_2^2 = \max_{i \in [m]} \left\| V\Lambda V^\mathsf{T} V\Sigma^+ (U^\mathsf{T})_i \right\|_2^2$$
$$= \max_{i \in [m]} (U^\mathsf{T})_i^\mathsf{T} (\Lambda \Sigma^+)^2 (U^\mathsf{T})_i$$
$$\leq \frac{c}{m} \operatorname{tr} \left((\Lambda \Sigma^+)^2 \right)$$
$$= \frac{c}{m} \sum_{i=1}^n \frac{\sigma^4 \xi_i}{(\sigma^2 \xi_i + a_n)^2}.$$

Plugging this back in the previous display and also using the fact that the squared Frobenius norm of a matrix is equal to the sum of the squares of its eigenvalues,

$$\operatorname{Risk}(S_0) = \frac{\sigma^2}{n} \sum_{i=1}^n \left(\frac{a_n}{\sigma^2 \xi_i + a_n}\right)^2 + \frac{c \cdot a_n}{n} \sum_{i=1}^n \frac{\sigma^4 \xi_i}{(\sigma^2 \xi_i + a_n)^2} \le c \cdot r(S_0)$$

Combining this with the lower bound in (63), we have

$$r(S_0) \le R_L(\mathcal{T}_{n,d}^k(C_n)) \le \min\left\{\sigma^2, \operatorname{Risk}(S_0)\right\} \le \min\left\{\sigma^2, \ c \cdot r(S_0)\right\}.$$
(65)

In other words, the minimax linear rate is essentially $r(S_0)$ up to a constant factor. Further, one of the estimators $\hat{y} = S_0 y$, $\hat{y} = y$ achieves the minimax linear rate up to a constant factor.

Now we bound $r(S_0)$. Let $\kappa = (k+1)^d$ denote the nullity of D. Recall from (64)

$$r(S_0) = \frac{1}{n} \sum_{i=1}^n \frac{\sigma^2 a_n}{\sigma^2 \xi_i + a_n} = \frac{\kappa \sigma^2}{n} + \frac{1}{n} \sum_{i=\kappa+1}^n \frac{\sigma^2 a_n}{\sigma^2 \xi_i + a_n}.$$
 (66)

Lower bounding $r(S_0)$. We give three lower bounds on $r(S_0)$. By using the fact that arithmetic mean of positive

numbers is at least as large as their harmonic mean, we have

$$r(S_{0}) = \frac{1}{n} \sum_{i=1}^{n} \frac{\sigma^{2} a_{n}}{\sigma^{2} \xi_{i} + a_{n}}$$

$$\geq \frac{n \sigma^{2} a_{n}}{\sum_{i=1}^{n} (\sigma^{2} \xi_{i} + a_{n})}$$

$$= \frac{n \sigma^{2} a_{n}}{n a_{n} + \sigma^{2} \|D\|_{F}^{2}}$$

$$= \frac{n \sigma^{2} a_{n}}{n a_{n} + \sigma^{2} d n^{1-1/d} \|D_{1d}^{(k+1)}\|_{F}^{2}}$$

$$= \frac{\sigma^{2} a_{n}}{a_{n} + \sigma^{2} d n^{-1/d} (n^{1/d} - k - 1) {2k+2 \choose k+1}}$$

$$\geq \frac{\sigma^{2} a_{n}}{a_{n} + \sigma^{2} d 4^{k+1}}$$
(67)

Now we bound in $r(S_0)$ in a second way. Let n_1 be the cardinality of $\{i \in [n] : \sigma^2 \xi_i \leq a_n\}$. Then

$$r(S_0) = \frac{1}{n} \sum_{i=1}^n \frac{\sigma^2 a_n}{\sigma^2 \xi_i + a_n} \ge \frac{1}{n} \sum_{i=1}^{n_1} \frac{\sigma^2 a_n}{a_n + a_n} = \frac{n_1 \sigma^2}{2n}.$$

Note that $n_1 = \lfloor nF(a_n/\sigma^2) \rfloor$ where F is the spectral distribution of $(D_{n,d}^{(k+1)})^{\mathsf{T}} D_{n,d}^{(k+1)}$ defined in Lemma 10. Applying Lemma 10, we get

$$r(S_0) \ge \frac{\sigma^2}{2} \left(F\left(\frac{a_n}{\sigma^2}\right) - \frac{1}{n} \right)$$

$$\ge c\sigma^2 \min\left\{ 1, (a_n/\sigma^2)^{\frac{1}{2s}} \right\} - \sigma^2/2n$$

$$= \min\left\{ c\sigma^2, c\sigma^{2-\frac{1}{s}} a_n^{\frac{1}{2s}} - \sigma^2/2n \right\}$$
(68)

In the special case s = 1/2, from Lemma 9 we get a third bound:

$$r(S_0) = \frac{1}{n} \sum_{i=1}^n \frac{\sigma^2 a_n}{\sigma^2 \xi_i + a_n} \ge c_1 a_n \log\left(1 + c_2/a_n\right)$$
(69)

where c_1, c_2 constants that depend only on k, d.

From (66),(67), (68) and (69) we have the lower bound

$$r(S_0) \ge \max\left\{\frac{\kappa\sigma^2}{n}, \frac{\sigma^2 a_n}{a_n + \sigma^2 d2^{2k+2}}, \sigma^2 \wedge c\sigma^{2-\frac{1}{s}} a_n^{\frac{1}{2s}} - \frac{\sigma^2}{2n}\right\}.$$
(70)

and an additional lower bound of $c_1 a_n \log (1 + c_2/a_n)$ in the case s = 1/2. Substituting $a_n = C_n^2/m$, using the assumption that $C_n^2/n \le 1$ and treating k, d, σ as constants, we get the stated lower bound. Upper bounding $r(S_0)$. If s < 1/2, then

$$r(S_0) = \frac{1}{n} \sum_{i=1}^n \frac{\sigma^2 a_n}{\sigma^2 \xi_i + a_n}$$

$$\leq \frac{\kappa \sigma^2}{n} + \frac{1}{n} \sum_{i=\kappa+1}^n \frac{\sigma^2 a_n}{\sigma^2 \xi_i}$$

$$= \frac{\kappa \sigma^2}{n} + \frac{a_n}{n} \sum_{i=1}^{\kappa+1} \frac{1}{\xi_i}$$

$$\leq \frac{\kappa \sigma^2}{n} + \frac{a_n}{n} (c_3 n)$$

$$= \frac{\kappa \sigma^2}{n} + c_3 a_n$$
(71)

We used Lemma 6 to control the second term in the third line. Similarly, if s = 1/2, $r(S_0) \le \kappa \sigma^2/n + c_3 a_n \log n$. For the case s > 1/2, we can write

$$r(S_{0}) = \frac{1}{n} \sum_{i=1}^{n} \frac{\sigma^{2} a_{n}}{\sigma^{2} \xi_{i} + a_{n}}$$

$$\leq \frac{1}{n} \sum_{i=1}^{n_{1}} \frac{\sigma^{2} a_{n}}{a_{n}} + \frac{1}{n} \sum_{i=n_{1}+1}^{n} \frac{\sigma^{2} a_{n}}{2\sigma^{2} \xi_{i}}$$

$$= \frac{n_{1} \sigma^{2}}{n} + \frac{a_{n}}{2n} \sum_{i=n_{1}+1}^{n} \frac{1}{\xi_{i}}$$

$$\leq c \frac{\sigma^{2}}{n} + c \sigma^{2} \left(\frac{a_{n}}{\sigma^{2}}\right)^{\frac{1}{2s}} + c \frac{a_{n}}{2n} n^{2s} \left(n(a_{n}/\sigma^{2})^{\frac{1}{2s}}\right)^{1-2s}$$

$$\leq c \frac{\sigma^{2}}{n} + c \sigma^{2-\frac{1}{s}} a_{n}^{\frac{1}{2s}}$$
(72)

To get the fourth line, we used Lemma 10 to bound n_1 and Lemma 6 to bound the summation.

Upper bound with the polynomial projection estimator $\hat{\theta}^{\text{poly}}$. For brevity, let Π denote the matrix that projects on to the null space of D. Note that $(I - \Pi)D^+ = D^+$. From bias variance decomposition similar to that in (60),

$$\frac{1}{n} \sup_{\theta_0 \in \mathcal{T}_{n,d}^k(C_n)} \mathbb{E}\left[\|\hat{\theta}^{\text{poly}} - \theta_0\|_2^2 \right] = \frac{\sigma^2}{n} \|\Pi\|_F^2 + \max_{i \in [m]} \|\left((\Pi - I)D^+\right)_i\|_2^2$$
$$= \frac{\kappa \sigma^2}{n} + \max_{i \in [m]} \|D_i^+\|_2^2$$

Then using incoherence of columns of U, that is, the fact that there exists a constant c > 1 that depends only on k, d such that $U_{ij}^2 \leq \frac{c}{m}$ for all $i \in [m], j \in [n]$, we can write

$$\max_{i \in [m]} \|D_i^+\|_2^2 = \max_{i \in [m]} \|V\Sigma^+(U^{\mathsf{T}})_i\|_2^2$$
$$= \max_{i \in [m]} (U^{\mathsf{T}})_i^{\mathsf{T}} (\Sigma^+)^2 (U^{\mathsf{T}})_i$$
$$\leq \frac{c}{m} \operatorname{tr} \left((\Sigma^+)^2 \right)$$
$$= \frac{c}{m} \sum_{i=\kappa+1}^n \frac{1}{\xi_i}$$

Plugging this back in the above display and using the bound on $\sum_{i=\kappa+1}^{n} \frac{1}{\xi_i}$ from Lemma 6, we get the desired result.

Upper bound with the projection estimator (34) when s > 1/2. From (60), for the projection estimator $\hat{\theta} = S_Q y$ in (34),

$$\operatorname{Risk}(\hat{\theta}) = \frac{\sigma^2}{n} |Q| + \frac{C_n^2}{n} \max_{i \in [m]} \left\| \left((S_Q - I)D^+ \right)_i \right\|_2^2.$$
(73)

Set $Q = [\tau]^d$ for a τ to be chosen later from (k + 2, N]. Also write $S_Q - I = V \Lambda_Q V^{\mathsf{T}}$. Again using incoherence of columns of U, we can write

$$\max_{i \in [m]} \left\| \left((S_Q - I)D^+ \right)_i \right\|_2^2 = \max_{i \in [m]} \left\| V \Lambda_Q V^\mathsf{T} V \Sigma^+ (U^\mathsf{T})_i \right\|_2^2$$
$$= \max_{i \in [m]} (U^\mathsf{T})_i^\mathsf{T} (\Lambda_Q \Sigma^+)^2 (U^\mathsf{T})_i$$
$$\leq \frac{c}{m} \operatorname{tr} \left((\Lambda_Q \Sigma^+)^2 \right)$$
$$= \frac{c}{m} \sum_{i \in [N]^d \setminus Q} \frac{1}{\xi_i}$$

The summation in the last line can be bound using Lemma 6 (recall s > 1/2 here):

$$\sum_{i \in [N]^d \setminus Q} \frac{1}{\xi_i} \le \sum_{\|(i-k-2)_+\|_2 \ge \tau - k - 2} \frac{1}{\xi_i} \le cn(n/(\tau - k - 2)^d)^{2s - 1}$$

Tracing this back to (73),

$$\operatorname{Risk}(\hat{\theta}) \le \frac{\sigma^2}{n} \tau^d + \frac{cC_n^2}{m} \cdot (n/(\tau - k - 2)^d)^{2s - 1}$$

Minimize this bound by setting τ such that $\tau^d \simeq (C_n/\sigma)^{\frac{1}{s}} n^{1-\frac{1}{2s}}$ to get the desired bound.

Remark 14. In Theorem 5, when $s \le 1/2$, the lower bound may also be obtained by embedding the ℓ_1 -ball $B_1(C_n/(2^{k+1}d))$ into $\mathcal{T}^k_{n,d}(C_n)$.

A.9 **Proof of Theorem 6 (Discrete Sobolev classes)**

Proof of upper bound. Like in the proof of minimax linear rates for KTV class in Theorem 5, for the projection estimator $\hat{\theta} = S_Q y$ where $S_Q = V_Q V_Q^T$, we can derive

$$\frac{1}{n} \sup_{\theta_0 \in \mathcal{W}_{n,d}^{k+1}(B_n)} \mathbb{E}\left[\|\hat{\theta} - \theta_0\|_2^2\right] = \frac{\sigma^2}{n} |Q| + \frac{1}{n} \sup_{\theta_0 \in \mathcal{W}_{n,d}^{k+1}(B_n)} \|(I - S_Q)\theta_0\|_2^2.$$

Denote $D = D_{n,d}^{(k+1)}$ for brevity. Set $Q = [\tau]^d$, where $\tau \in (k+2, N]$ is an integer (recall $N = n^{1/d}$) and analyze the maximum of the second term:

$$\begin{split} \sup_{\theta_0: \|D\theta_0\|_2 \le B_n} \frac{1}{n} \| (I - S_q) \theta_0 \|_2^2 &= \sup_{z: \|z\|_2 \le C_n} \frac{1}{n} \| (I - S_q) D^{\dagger} z \|_2^2 \\ &= \frac{B_n^2}{n} \sigma_{\max}^2 \left((I - S_q) D^{\dagger} \right) \\ &\le \frac{B_n^2}{n} \frac{1}{4^{k+1} \sin^{2k+2} (\pi (\tau - k - 2)/(2N))} \\ &\le \frac{B_n^2}{n} \frac{N^{2k+2}}{(\pi (\tau - k - 2))^{2k+2}}. \end{split}$$

Here we denote by $\sigma_{\max}(A)$ the maximum singular value of a matrix A. The last inequality above used the inequality $\sin(x) \ge x/2$ for $x \in [0, \pi/2]$. The earlier inequality used that $\sigma_{\max}^2((I - S_Q)D^{\dagger})$ is the reciprocal of the smallest eigenvalue ρ_Q of $M = D^{\mathsf{T}}D$ with index in $[N]^d \setminus Q$. That is,

$$\rho_Q = \rho_{\tau+1,1,\dots,1} \ge \left(4\sin^2(\pi(\tau-k-2)/(2N))\right)^{k+1},$$

where the last inequality is due to the relation in (86). Hence, we have established

$$\sup_{\theta_0: \|D\theta_0\|_2 \le B_n} \frac{1}{n} \mathbb{E} \left[\|\hat{\theta} - \theta_0\|_2^2 \right] \le \frac{\sigma^2}{n} \tau^d + \frac{B_n^2}{n} \frac{N^{2k+2}}{(\pi(\tau - k - 2))^{2k+2}}$$

Choosing τ to balance the two terms on the right-hand side above results in $\tau^d \simeq (k+2)^d + (B_n^2 n^2 s / \sigma^2)^{\frac{1}{2s+1}}$. Also, in the edge case where $Q = [N]^d$, the risk is σ^2 . Plugging this choice of τ gives the upper bound result.

Proof of lower bound. Similar to argument in the proof of Theorem 4, the nullity of $D_{n,d}^{(k+1)}$ implies the lower bound

$$R(\mathcal{W}_{n,d}^{k+1}(C_n)) = \Omega\left(\frac{\kappa\sigma^2}{n}\right).$$
(74)

The Holder ball embedding

$$\mathcal{W}_{n,d}^{k+1}(C_n) \supseteq \mathcal{C}_{n,d}^k(cC_n n^{s-\frac{1}{2}})$$

implies that

$$R(\mathcal{W}_{n,d}^{k+1}(C_n)) \ge R\left(\mathcal{C}_{n,d}^k(cC_n n^{s-\frac{1}{2}})\right) \gtrsim \left(\frac{C_n^2}{n}\right)^{\frac{1}{2s+1}} \sigma^{\frac{4s}{2s+1}} \wedge \sigma^2,$$

where the second inequality follows from (59). Putting these two bounds together, we get the desired lower bound.

B Estimation theory for Graph trend filtering on grids

We recall the GTF operator from Wang et al. (2016) for convenience. Let G(V, E) be a graph with n vertices and m edges $(u_1, v_1), \ldots, (u_m, v_m) \in [n] \times [n]$. Let $D \in \mathbb{R}^{m \times n}$ be the incidence matrix of G satisfying

$$|(Dx)_j| = |x_{u_j} - x_{v_j}|$$
 for all $x \in \mathbb{R}^n$

for all edges (u_j, v_j) for $j \in [m]$. The graph Laplacian is $L = D^{\mathsf{T}}D$. The GTF operators of all orders are defined by

$$S_{n,d}^{(1)} = D, \quad S_{n,d}^{(2)} = L,$$

$$S_{n,d}^{(2k+1)} = DL^k, \quad S_{n,d}^{(2k)} = L^k \text{ for } k \ge 0, k \in \mathbb{Z}.$$
(75)

B.1 Upper bounds for GTF

Wang et al. (2016) used Theorem 2 (their Theorem 6) in order to derive error rates for GTF on 2d grids already; see their Corollary 8. Sadhanala et al. (2017) refine this result using a tighter upper bound for the partial sum of inverse eigenvalues. Here, we give a more general result that applies to not just 2d grids, but all $d \ge 2$ and $k \ge 0$. We further show that these rates are optimal by deriving a matching lower bound. Recall the abbreviation s = (k + 1)/d.

Theorem 7. Assume that $d \ge 1$ and $k \ge 0$. Denote $C_n = \|S_{n,d}^{(k+1)}\theta_0\|_1$. Then GTF defined by the estimator in (30) with $D = S_{n,d}^{(k+1)}$ in (75) satisfies

$$\frac{1}{n} \|\hat{\theta} - \theta_0\|_2^2 = O_{\mathbb{P}} \left(\frac{1}{n} + \frac{\lambda}{n} C_n \right)$$

with

$$\lambda \asymp \begin{cases} \sqrt{\log n} & s < 1/2\\ \log n & s = 1/2\\ (\log n)^{\frac{1}{2s+1}} \left(\frac{n}{C_n}\right)^{\frac{2s-1}{2s+1}} & s > 1/2. \end{cases}$$

With canonical scaling of C_n , we see the following error bound.

Corollary 2. With canonical scaling $C_n = C_n^* = n^{1-s}$, the GTF estimator with λ scaling as in Theorem 7 satisfies

$$\sup_{\theta_0 \in \mathcal{S}_{n,d}^k(C_n)} \frac{1}{n} \| \hat{\theta} - \theta_0 \|_2^2 = \begin{cases} O_{\mathbb{P}} \left(n^{-s} \sqrt{\log n} \right) & s < 1/2 \\ O_{\mathbb{P}} \left(n^{-s} \log n \right) & s = 1/2 \\ O_{\mathbb{P}} \left(n^{-\frac{2s}{2s+1}} (\log n)^{\frac{1}{2s+1}} \right) & s > 1/2. \end{cases}$$

Remarks following Theorem 3 for KTF apply for GTF as well. The proof is in Appendix B.4.

B.2 Lower bound

Similar to the lower bound in Theorem 4 for KTV class, we give a bound for the Graph total variation (GTV) class

$$S_{n,d}^k(C_n) = \{ \theta \in \mathbb{R}^n : \|S_{n,d}^{(k+1)}\theta\|_1 \le C_n \}.$$
(76)

Due to the lower order discrete derivatives on the boundary of the grid $Z_{n,d}$, the GTV class $S_{n,d}^k(C_n)$ cannot contain the discrete Holder class with appropriate scaling $C_{n,d}^k(C_n n^{s-1})$; see Lemma 4 in Sadhanala et al. (2017). However, by an alternative route Sadhanala et al. (2017) show a lower bound for $S_{n,d}^k(C_n)$ that matches with the lower bound for the Holder class $C_{n,d}^k(C_n n^{s-1})$. We further tighten their result by embedding an ℓ_1 ball of appropriate size.

Theorem 8. For any integers $k \ge 0$, $d \ge 1$, the minimax estimation error for GTV class defined in (76) satisfies

$$R\left(\mathcal{S}_{n,d}^k(C_n)\right) = \Omega\left(\frac{\sigma^2}{n} + \frac{\sigma C_n}{n} + \left(\frac{C_n}{n}\right)^{\frac{2s}{2s+1}} \sigma^{\frac{4s}{2s+1}} \wedge \sigma^2\right).$$

Proof of Theorem 8. Similar to the proof of Theorem 4, it is sufficient to show three lower bounds separately. We get the first two lower bounds just as in the proof of Theorem 4 by using the fact that $\operatorname{nullity}(S_{n,d}^{(k+1)}) = 1$ and the ℓ_1 -ball embedding

$$B_1(C_n/(2^{k+1}d)) \subseteq \mathcal{S}_{n,d}^k(C_n)$$

from Lemma 3 and the fact that $\|S_{n,d}^{(k+1)}\|_{1,\infty} \le 2^{k+1}d$. The third term is from Theorem 5 in Sadhanala et al. (2017).

Minimax linear rate for GTV class **B.3**

The minimax linear rate analysis for GTV class is very similar to that for KTV class. So we simply state the result and skip the proof.

Theorem 9. The minimax linear risk over the KTV class in (23) satisfies, for any sequence $C_n \leq \sqrt{n}$,

$$R_L(\mathcal{T}_{n,d}^k(C_n)) = \begin{cases} \Omega(1/n + C_n^2/n) & \text{if } s < 1/2, \\ \Omega(1/n + C_n^2/n\log(1 + n/C_n^2)) & \text{if } s = 1/2, \\ \Omega(1/n + (C_n^2/n)^{\frac{1}{2s}}) & \text{if } s > 1/2. \end{cases}$$
(77)

This is achieved in rate by the projection estimator in (34), by setting $Q = [\tau]^d$ for $\tau^d \simeq (C_n n^{s-1/2})^{1/s}$, in the case s > 1/2. When s < 1/2, the simple mean estimator, $\hat{\theta}^{\text{mean}} = \bar{y}\mathbf{1}$, achieves the rate in (77). When s = 1/2, this estimator achieves the rate in (77) up to a log factor. Lastly, if $C_n^2 = O(n^{\alpha})$ for $\alpha < 1$, and still s = 1/2, then the mean estimator achieves the rate in (77) without the additional log factor.

B.4 Proof of Theorem 7

For d = 2, it is shown in the proof of Corollary 8 in Wang et al. (2016) that the GTF operator $S_{n,d}^{(k+1)}$ satisfies the incoherence property, as defined in Theorem 2, with a constant $\mu = 4$ when k is even and $\mu = 2$ when k is odd. Here we extend this incoherence property for d > 2 using Lemma 1. We treat the cases where k is odd and even separately.

If k is odd we can extend the argument from Corollary 8 in Wang et al. (2016) in a straightforward manner. The GTF operator is $S_{n,d}^{(k+1)} = L^{(k+1)/2}$ where L is the Laplacian of the d-dimensional grid graph. Denoting the Laplacian of the chain graph of length N by L_{1d} , L is given by

$$L = L_{1d} \otimes I \otimes I + I \otimes L_{1d} \otimes I + I \otimes I \otimes L_{1d}$$

for d = 3 and

$$L = L_{1d} \otimes I \cdots \otimes I + I \otimes L_{1d} \cdots \otimes I + \cdots + I \otimes \ldots I \otimes L_{1d}$$

for general d where each term in the summation is a Kronecker product of d matrices. Let $\alpha_i, u_i, i \in [N]$ be the eigenvalues and eigenvectors of L_{1d} . As shown in Wang et al. (2016), in 1d, we have the incoherence property $||u_i||_{\infty} \leq \sqrt{2/N}$ for all $i \in [N]$. The eigenvalues of L are $\sum_{j=1}^d \alpha_{i_j}$ and the corresponding eigenvectors are $u_{i_1} \otimes \cdots \otimes u_{i_d}$ for $i_1, \ldots, i_d \in [N]$. Clearly, incoherence holds for the eigenvectors of L with constant $\mu = 2^{d/2}$.

If k is even, then the left singular vectors of $S_{n,d}^{(k+1)}$ are the same as those of $S_{n,d}^{(1)}$. We know that both the left and right singular vectors of $D_{1d}^{(1)}$ satisfy the incoherence property with constant $\mu = \sqrt{2}$ (see the proof of Corollary 7 in Wang et al. (2016)). Setting $D = D_{1d}^{(1)}$ in Lemma 1, we see that the left singular vectors of $S_{n,d}^{(1)}$ and hence those of $S_{n,d}^{(k+1)}$ satisfy incoherence property with constant $2^{d/2}$. Therefore, for all integers $k \ge 0$, the left singular vectors of $S_{n,d}^{(1)}$ are incoherent with constant $2^{d/2}$.

From the incoherence property and Theorem 2, the GTF estimator $\hat{\theta}$, satisfies

$$\frac{1}{n} \|\hat{\theta} - \theta_0\|_2^2 = O_{\mathbb{P}} \left(\frac{1}{n} + \frac{|I|}{n} + \frac{\mu}{n} \sqrt{\frac{\log n}{n} \sum_{i \in [N]^d \setminus (I \cup \{1\}^d)} \frac{1}{\rho_i^2}} \cdot \|\Delta \theta_0\|_1 \right),\tag{78}$$

where $\rho_i, i \in [N]^d$ are the eigenvalues $S_{n,d}^{(k+1)^{\mathsf{T}}} S_{n,d}^{(k+1)}$ and $\mu = 2^{d/2}$.

Consider the set $I = \{i \in [N]^d : ||i - 1||_2 < r\} \setminus \{1\}^d$ for an $r \in [1, \sqrt{dN}]$ chosen later. Lemma 5 gives the key calculation where it is shown that for large enough n,

$$\sum_{\|i-1\| \ge r} \frac{1}{\rho_i^2} = \sum_{\|i-1\| \ge r} \frac{1}{\lambda_i^{k+1}} \le c \begin{cases} n & s < 1/2 \\ n \log(2\sqrt{d}N/r) & s = 1/2 \\ n(n/r^d)^{2s-1} & s > 1/2 \end{cases}$$

where $\lambda_i, i \in [N]^d$ are eigenvalues of the Laplacian L and c > 0 is a constant that depends only on k, d.

For $s \le 1/2$, to minimize the upper bound in (78), set r = 1 so that I is empty and apply the above inequality. This gives the desired bound. Now consider s > 1/2. Note that $|I| \le r^d$ because $I \subseteq [r]^d$. Therefore (78) reduces to

$$\frac{1}{n} \|\hat{\theta} - \theta_0\|_2^2 = O_{\mathbb{P}} \left(\frac{r^d}{n} + \frac{\mu}{n} \sqrt{\log n (n/r^d)^{2s-1}} \|\Delta\theta_0\|_1 \right)$$
(79)

To minimize the upper bound in (79) balance

$$r^d$$
 with $\frac{C_n}{\sqrt{n}}\sqrt{n(n/r^d)^{2s-1}\log n}$.

This leads us to take

$$r^d \asymp (C_n \sqrt{\log n})^{\frac{2}{2s+1}} n^{\frac{2s-1}{2s+1}}$$

and plugging this in (79) gives the desired bound for s > 1/2. This completes the proof.

C Technical lemmas

Lemma 5. Consider the eigenvalues $\{\lambda_i : i = (i_1, \dots, i_d) \in [N]^d\}$ of the d-dimensional grid graph Laplacian with $n = N^d$ nodes. Let k be a non-negative integer and $r_0 \in [1, \sqrt{dN}]$. Then,

$$\sum_{i \in [N]^d: \|i-1\|_2^2 \ge r_0^2} \frac{1}{\lambda_i^k} \le c \begin{cases} n & 2k < d \\ n \log(2\sqrt{d}N/r_0) & 2k = d \\ N^{2k} r_0^{d-2k} & 2k > d \end{cases}$$

for a constant c > 0 that depends on k, d but not on N, r_0 .

Proof of Lemma 5. Let I denote the summation on the left. Then

$$I = \sum_{i \in [N]^d: ||i-1||_2 \ge r_0} \frac{1}{\lambda_i^k} = \sum_{||i-1||_2 \ge r_0} \left(\sum_{j=1}^d 4 \sin^2 \frac{\pi(i_j - 1)}{2N} \right)^{-k}$$

$$\leq \sum_{||i-1||_2 \ge r_0} \left(\sum_{j=1}^d \frac{\pi^2(i_j - 1)^2}{4N^2} \right)^{-k}$$

$$= cN^{2k} \sum_{||i-1||_2 \ge r_0} \left(\sum_{j=1}^d (i_j - 1)^2 \right)^{-k}$$

$$\leq cN^{2k} \sum_{i \in \{0, 1, \dots, N-1\}^d: ||i||_2 \ge r_0} ||i||_2^{-2k}$$
(80)

In the second line, we used the fact that $\sin x \ge x/2$ for $x \in [0, \pi/2]$.

Case $r_0 \ge 2\sqrt{d}$. In the last expression, upper bound $||i||_2^{-2k}$ with the integral of $f(x) = ||x||_2^{-2k}$, $f : \mathbb{R}^d \to \mathbb{R}$ over the unit length cube whose top right corner is at *i*. Note that, the norm of any point in this cube is at least $||i-1||_2 \ge 1$

 $||i||_2 - ||\mathbf{1}||_2 = ||i||_2 - \sqrt{d} \ge r_0 - \sqrt{d} \ge r_0/2$. Therefore, we can continue to bound

$$I \le cN^{2k} \int_{r_0/2 \le \|x\|_2 \le \sqrt{d}N} \|x\|_2^{-2k} dx$$
$$\le cN^{2k} \int_{r_0/2 \le r \le \sqrt{d}N} (r^2)^{-k} r^{d-1} dr$$

The last line is obtained by changing to polar coordinates and integrating out the angles. Recall that the constants c may change from line to line and they may depend on k, d, but not on N, r_0 . If d = 2k, then the integral

$$I \le cN^{2k} \log(2\sqrt{d}N/r_0) = cn \log(2\sqrt{d}N/r_0).$$

If 2k < d, then

$$I \le cN^{2k} \left((N\sqrt{d})^{d-2k} - (r_0/2)^{d-2k} \right) \le cN^d.$$

If 2k > d, then

$$I \le cN^{2k} \left((r_0/2)^{d-2k} - (N\sqrt{d})^{d-2k} \right).$$

Treating d, k as constants, we write

$$I \le cN^{2k}r_0^{d-2k}.$$

Case $r_0 < 2\sqrt{d}$. Continuing from (80), write

$$I \le cN^{2k} \sum_{i \in \{0,1,\dots,N-1\}^d: \|i\|_2 \in [r_0, 2\sqrt{d})} \|i\|_2^{-2k} + cN^{2k} \sum_{i \in \{0,1,\dots,N-1\}^d: \|i\|_2 \ge 2\sqrt{d}} \|i\|_2^{-2k}$$
(81)

From the previous case, the second summation can be upper bound with cn if 2k < d, $cn \log n$ if 2k = d and cN^{2k} if 2k > d. In the first summation (in the above display), the number of entries i is at most $(2\sqrt{d})^d$ and each entry is at most r_0^{-2k} . Therefore the first term is at most $cN^{2k}(2\sqrt{d})^d r_0^{-2k}$. Putting the two sums together, we can verify the stated bounds.

Following lemma provides a result analogous to Lemma 5 for KTF.

Lemma 6. Let $\{\xi_i : i = (i_1, \dots, i_d) \in [N]^d\}$ be the eigenvalues of $D_{n,d}^{(k+1)^{\mathsf{T}}} D_{n,d}^{(k+1)}$ and suppose $r_0 \in [1, \sqrt{dN}]$. Then,

$$\sum_{i \in [N]^d \setminus [k+2]^d} \frac{1}{\xi_i^2} \le c \begin{cases} n & 2(k+1) < d \\ n \log n & 2(k+1) = d. \end{cases}$$

In the case 2k + 2 > d*,*

$$\sum_{i \in [N]^d : \|(i-k-2)_+\|_2 \ge r_0} \frac{1}{\xi_i^2} \le cN^{2k+2} r_0^{d-2k-2}$$

Here c > 0 *is a constant that depends on* k, d *but not on* N, r_0 *.*

Proof. Using Lemma 7, we can write

$$\sum_{i \in [N]^d : \|(i-k-2)_+\|_2 \ge r_0} \frac{1}{\xi_i^2} \le d^{2k} \sum_{i \in [N]^d : \|(i-k-2)_+\|_2 \ge r_0} \frac{1}{\lambda_{i-k-1}^{2k+2}} \le d^{2k} \sum_{i \in [N]^d : \|i-1\|_2 \ge r_0} \frac{1}{\lambda_i^{2k+2}}.$$
(82)

Applying Lemma 5 directly gives the desired result in the case 2k + 2 > d. In the case $2k + 2 \le d$, we get the bound by setting $r_0 = 1$ in (82) and then applying Lemma 5.

Lemma 7. Let $\{\xi_i : i = (i_1, \dots, i_d) \in [N]^d\}$ be the eigenvalues of $D_{n,d}^{(k+1)\mathsf{T}} D_{n,d}^{(k+1)}$ for $k \ge 0, d \ge 1, N \ge 1, n = N^d$. Let $\alpha_i, i \in [N]$ be the eigenvalues of L, the Laplacian of chain graph of length N. Let $\lambda_{i_1,\dots,i_d} = \sum_{j=1}^d \alpha_{i_j}, i \le N$ elementwise with the convention that $\alpha_\ell = 0$ for $\ell \le 0$. Then

$$\xi_i \ge d^{-k} \lambda_{i-k-1}^{k+1} \text{ for } i \in [N]^d.$$

Proof. Abbreviate $D = D_{N,1}^{(k+1)}$, and let G be the kth order GTF operator defined over a 1d chain of length N. Also let N' = N - k - 1, and $k' = \lfloor (k+1)/2 \rfloor$. Let

- β_ℓ, ℓ ∈ [N'] be the eigenvalues of DD^T
 γ_ℓ, ℓ ∈ [N''] be the eigenvalues of GG^T where N'' = N − 1{k is even}

 GG^{T} and $G^{\mathsf{T}}G$ should have the same nonzero eigenvalues. From the definition of $G, G^{\mathsf{T}}G = L^{k+1}$. The first eigenvalue of L is 0 and the rest are nonzero. Putting these facts together, we see that

$$\gamma_{\ell} = \alpha_{\ell+N-N''}^{k+1} \text{ and } \alpha_{\ell}^{k+1} \le \gamma_{\ell} \quad \text{ for } \ell \in [N''].$$
(83)

Removing the top k' and bottom k' rows of G yields D, i.e.,

$$D = PG$$
, where $P = \begin{bmatrix} 0_{N' \times k'} & I_{N'} & 0_{N' \times k'} \end{bmatrix}$.

As $DD^{\mathsf{T}} = PGG^{\mathsf{T}}P^{\mathsf{T}}$ and $PP^{\mathsf{T}} = I_{N'}$, Cauchy interlacing theorem (Lemma 8) tells us that

$$\gamma_i \le \beta_i \le \gamma_{i+N''-N'}, \quad \text{for } i \in [N'].$$
(84)

Thanks to the Kronecker sum structure, the eigenvalues of $(D_{n,d}^{(k+1)})^{\mathsf{T}} D_{n,d}^{(k+1)}$ are

$$\xi_{i_1,...,i_d} = \sum_{j=1}^d \rho_{i_j}, \quad i \in [N]^d,$$

where ρ_1, \ldots, ρ_N denote the eigenvalues of $D^{\mathsf{T}}D$, i.e., $\rho_1 = \cdots = \rho_{k+1} = 0$ and $\rho_{\ell+k+1} = \beta_{\ell}\ell \in [N']$. Similarly, we can write the eigenvalues of the Laplacian of the d-dimensional grid graph as

$$\lambda_{i_1,\dots,i_d} = \sum_{j=1}^d \alpha_{i_j}, \quad i \in [N]^d$$

For arbitrary $i \in [N]^d$, we can write

$$\xi_{i_1,\dots,i_d} = \sum_{j=1}^d \beta_{i_j-k-1} \ge \sum_{j=1}^d \gamma_{i_j-k-1} \ge \sum_{j=1}^d \alpha_{i_j-k-1}^{k+1} \ge d^{-k} \lambda_{i_1-k-1,\dots,i_d-k-1}^{k+1}$$

with the convention $\alpha_{\ell} = \beta_{\ell} = \gamma_{\ell} = 0$ for $\ell \leq 0$. The first inequality is due to (84), the second is due to (83), and the third is due to a simple application of Jensen's inequality: $(\frac{1}{d}\sum_{j=1}^{d}a_{i})^{k} \leq \frac{1}{d}\sum_{j=1}^{d}a_{i}^{k}$ if $k \geq 1$ and $a \geq 0$ elementwise.

Lemma 8 (Cauchy Interlacing theorem). Let A be an $n \times n$ symmetric matrix, $P \in \mathbb{R}^{m \times n}$ be an orthogonal projection matrix (satisfying $PP^{\mathsf{T}} = I_m$) with $m \leq n$ and define $B = PAP^{\mathsf{T}}$. Let $\alpha_1 \leq \alpha_2 \leq \cdots \leq \alpha_n$ be the eigenvalues of A and $\beta_1 \leq \beta_2 \leq \cdots \leq \beta_m$ be the eigenvalues of B. Then

$$\alpha_i \leq \beta_i \leq \alpha_{i+n-m}, \quad \text{for } i \in [m].$$

Lemma 9. Let $\{\xi_i : i = (i_1, \dots, i_d) \in [N]^d\}$ be the eigenvalues of $D_{n,d}^{(k+1)^{\mathsf{T}}} D_{n,d}^{(k+1)}$ for $k \ge 0, d \ge 1, N \ge 1, n = N^d$. Suppose s = 1/2 and a > 0. Then

$$\sum_{i \in [N]^d} \frac{1}{\xi_i + a} \ge cn \log \left(1 + \pi^{2k+2} a^{-1} \right)$$

for a constant c that depends only on k, d.

Proof of Lemma 9. From (86) and the inequality $\sin x \le x$ for $x \ge 0$, for any $i \in [N]^d$,

$$\xi_i = \sum_{j=1}^d \rho_{i_j} \le \sum_{j=1}^d 4^{k+1} \sin^{2k+2} \frac{\pi(i_j-1)}{2N} \le \pi^{2k+2} n^{-2s} \|i-1\|_{2k+2}^{2k+2} \le \pi^{2k+2} n^{-2s} \|i-1\|_2^{2k+2}.$$

With this inequality,

$$\sum_{i \in [N]^d} \frac{1}{\xi_i + a} \ge \sum_{i \in [N]^d} \frac{1}{\pi^{2k+2} n^{-2s} \|i - 1\|_2^{2k+2} + a}$$
$$\ge c \int_{r=0}^N \frac{1}{\pi^{2k+2} n^{-2s} r^{2k+2} + a} r^{d-1} dr.$$
(85)

In the second inequality is obtained as follows. Consider axis-parallel unit cubes with corners located at integer coordinates. Let $A_i \subset \mathbb{R}^d$ be the cube with its farthest corner from origin located at *i*, for $i \in [N]^d$. Clearly,

$$\frac{1}{\pi^{2k+2}n^{-2s}\|i-1\|_2^{2k+2}+a} \ge \int_{A_i} \frac{1}{\pi^{2k+2}n^{-2s}\|x\|_2^{2k+2}+a} \, dx.$$

Next observe that the set $\{x \in \mathbb{R}^d : ||x||_2 \le N, x \ge 0\}$ is contained in the cube $\{x \in \mathbb{R}^d : ||x||_\infty \le N, x \ge 0\}$ The former set is the non-negative orthant of the ℓ_2 ball of radius N in \mathbb{R}^d . So, for radially symmetric functions f, integral of f over this set is 2^{-d} times its integral over the ℓ_2 ball. This justifies (85) after a change to polar coordinates. In the integral (85), noting that s = 1/2, 2k + 2 = d, put $u = r^d$ to get

$$\sum_{\in [N]^d} \frac{1}{\xi_i + a} \ge c \int_0^n \frac{1}{\pi^{2k+2} u/n + a} \, du = cn \log\left(1 + \pi^{2k+2} a^{-1}\right).$$

Lemma 10. Let $\{\xi_i : i = (i_1, \dots, i_d) \in [N]^d\}$ be the eigenvalues of $D_{n,d}^{(k+1)^{\mathsf{T}}} D_{n,d}^{(k+1)}$ for $k \ge 0, d \ge 1, N \ge 1, n = N^d$. Let $\alpha_i, i \in [N]$ be the eigenvalues of L, the Laplacian of chain graph of length N. Define

$$F(t) = \frac{1}{n} \sum_{i \in [N]^d} 1\{\lambda_i \le t\}, \quad \text{for } t \in [0, \lambda_n].$$

Then there exist constants $c_1, c_2, c_3 > 0$ independent of n, t such that

$$c_1 t^{\frac{d}{2k+2}} \le F(t) \le c_2 + c_3 t^{\frac{d}{2k+2}}$$

for all $t \in [0, \lambda_n]$.

Proof of Lemma 10. Using the notation in the proof of Lemma 6,

$$\lambda_{i_1,\dots,i_d} = \rho_{i_1} + \dots + \rho_{i_d}, \quad \text{for } (i_1,\dots,i_d) \in [N]^d$$

where $\rho_{\ell} = \beta_{\ell-k-1}, \ell \in [N]$ with the convention that $\beta_{\ell} = 0$ for $\ell \leq 0$. From (84), (83) and the fact that the eigenvalues of chain Laplacian are given by $4 \sin^2 \frac{\pi(\ell-1)}{2N}$ for $\ell \in [N]$, we have

$$\left(4\sin^2\frac{\pi(\ell-k-2)_+}{2N}\right)^{k+1} \le \rho_i \le \left(4\sin^2\frac{\pi(\ell-1)}{2N}\right)^{k+1}, \quad \text{for } \ell \in [N]$$
(86)

where $(x)_{+} = \max\{x, 0\}$ for $x \in \mathbb{R}$. The upper bound can be argued as follows.

$$nF(t) = \sum_{i \in [N]^d} 1\left\{\lambda_{i_1,\dots,i_d} \le t\right\}$$

= $\sum_{i \in [N]^d} 1\left\{\sum_{j=1}^d \rho_{i_j} \le t\right\}$
 $\le \sum_{i \in [N]^d} 1\left\{\sum_{j=1}^d 4^{k+1} \sin^{2k+2} \frac{\pi(i_j - k - 2)_+}{2N} \le t\right\}$
 $\le \sum_{i \in [N]^d} 1\left\{\sum_{j=1}^d \left(\frac{\pi}{2}\right)^{2k+2} (i_j - k - 2)_+^{2k+2} \le tN^{2k+2}\right\}$

In the third line, we use (86) and in the fourth line, we use the fact that $\sin x \ge x/2$ for $x \in [0, \pi/2]$. Observe that

$$\left(\frac{\pi}{2}\right)^{2k+2} \sum_{j=1}^{d} (i_j - k - 2)^{2k+2}_+ \le t N^{2k+2} \Rightarrow ||(i - k - 2)_+||_{\infty} \le \frac{2}{\pi} N t^{\frac{1}{2k+2}}$$
$$\Rightarrow ||i||_{\infty} \le k + 2 + \frac{2}{\pi} N t^{\frac{1}{2k+2}}$$

Applying this fact to the previous bound on nF(t),

$$nF(t) \le \sum_{i \in [N]^d} 1\Big\{ \|i\|_{\infty} \le k+2 + \frac{2}{\pi} N t^{\frac{1}{2k+2}} \Big\} \le \Big(k+2 + \frac{2}{\pi} N t^{\frac{1}{2k+2}}\Big)^d \le 2^{d-1} (k+2)^d + 2^{2d-1} \pi^{-d} n t^{\frac{d}{2k+2}}.$$

where in the last inequality we used the fact that $(a + b)^d \le 2^{d-1}(a^d + b^d)$ for $a, b \ge 0, d \ge 1$.

The lower bound can be derived as follows. Certainly, $F(t) \ge F(0) = \kappa/n$ for $t \ge 0$. We can write

$$nF(t) = \sum_{i \in [N]^d} 1\{\lambda_{i_1,\dots,i_d} \le t\}$$

= $\sum_{i \in [N]^d} 1\{\sum_{j=1}^d \rho_{i_j} \le t\}$
= $\sum_{i \in [N]^d} 1\{\sum_{j=1}^d 4^{k+1} \sin^{2k+2} \frac{\pi(i_j-1)}{2N} \le t\}$
 $\ge \sum_{i \in [N]^d} 1\{\sum_{j=1}^d \pi^{2k+2}(i_j-1)^{2k+2} \le tN^{2k+2}\}$
= $\sum_{i \in [N]^d} 1\{\sum_{j=1}^d \|i-1\|_{2k+2} \le r\}$

where $r = \frac{1}{\pi}Nt^{\frac{1}{2k+2}}$. In the third line, we used (86) and in the fourth line, we used the fact that $\sin x \le x$ for $x \ge 0$. Note that, we can inscribe a cube $\{i : ||i-1||_{\infty} \le rd^{\frac{-1}{2k+2}}\}$ in the ℓ_{2k+2} body $\{i : ||i-1||_{2k+2} \le r\}$ and the cube contains $(1 + \lfloor rd^{\frac{-1}{2k+2}} \rfloor)^d$ lattice points in $[N]^d$. Therefore, continuing to bound from the previous display,

$$nF(t) \ge \left(1 + \lfloor \frac{1}{\pi} d^{\frac{-1}{2k+2}} N t^{\frac{1}{2k+2}} \rfloor \right)^d \ge \frac{1}{\pi^d} d^{\frac{-d}{2k+2}} n t^{\frac{d}{2k+2}}.$$

where in the last inequality we used the fact $(1 + \lfloor x \rfloor)^d \ge x^d$ for $x \ge 0$.

D Fast algorithm for degrees of freedom

From (16), given a KTF estimate $\hat{\theta}$, nullity (D_{-A}) is an unbiased estimate of degrees of freedom of KTF where A denotes the set of rows r in D for which $(D\hat{\theta})_r \neq 0$. We give an algorithm to compute nullity (D_{-A}) in O(ndk) time.

Algorithm 3 Degrees-of-freedom (θ, k)		
Input: fitted values $\theta \in \mathbb{R}^n$, trend filtering order $k \ge 0$ Output: estimate of degrees of freedom	$\overline{\frac{\mathbf{SPREAD}(p: Piece)}{p.set = true}}$	
 struct Piece: set = false, knowns = 0, start, end ∈ [N]^d pieces : Piece[], pieces-containing[i]: Piece[], set[i] : boo for i ∈ [N]^d for d' ∈ [d] for i' ∈ [N]^{d-1}: (a) i←i' with an extra 1 inserted before d'th entry: 	for vertex <i>i</i> on the line [<i>p.start</i> , <i>p.end</i>]: 1. if set[<i>i</i>] : continue 2. set[<i>i</i>] \leftarrow true 3. SPREAD-VERTEX(<i>i</i>) MAKE-POLYNOMIAL-PIECES(θ $i \in [N]^d$ $d' \in [d]$)	
$i_j = i'_j$ for $j < d'$, $i_j = 1$ for $j = d'$, $i_j = i'_{j-1}$ for $j > d'$ (b) MAKE-POLYNOMIAL-PIECES (θ, i, d')	makes polynomial pieces on the line containing <i>i</i> along axis d'	
4. $df = 0$		
5. for p in pieces: (a) if p set : continue	1. $a_j \leftarrow \theta[i \text{ with } i_{d'} = j] \text{ for } j \in [N]$ 2. while $j \leq N$	
(b) df $+= \max\{0, \min\{p.length, k+1\} - p.knowns\}$ (c) SPREAD (p)	(a) start = j, end = j (b) while $(j \le N' \text{ and } \langle w, a[j:j+k+1] \rangle = 0)$	
6. return df	(c) if $j \neq$ start: end $\leftarrow j + k$	
SPREAD-VERTEX (i)	(d) Piece p(i with $i_{d'}$ = start, i with $i_{d'}$ = end)	
for piece q in pieces-containing $[i]$: 1. if $q.set$ continue	(e) for j' in [start, end]: pieces-containing[i with $i_{d'} = j'$].add(p)	
2. q .knowns++ 3. if q .knowns> k : SPREAD (q)	(f) pieces.add(p)	

Notation: N' = N - k - 1, $w \in \mathbb{R}^{k+2}$ is the (k+1)th order difference vector.

D.1 Time complexity and correctness of the algorithm

Let A denote the set of rows r in D for which $(D\hat{\theta})_r \neq 0$. Denote the null space of D_{-A} with \mathcal{N} .

In step 3 of DEGREES-OF-FREEDOM, we find line segments on the lattice where θ is a *k*th degree polynomial. In a bit more detail, for each straight line in the lattice between opposing faces, we find segments along the line where θ is a *k*th degree polynomial. We call these line segments (polynomial) *pieces* in the algorithm. In 2d, MAKE-POLYNOMIAL-PIECES is called on rows and columns separately, and a piece is a part of a row or a column. An important characterization of the null space \mathcal{N} is the following:

A $\theta \in \mathcal{N}$ iff it is a *k*th degree polynomial on all the pieces found in step 3.

If a piece has fewer than k + 2 elements, then any θ is trivially a kth degree polynomial on the piece. Otherwise, k + 1 values on the piece determine a polynomial on the piece.

We pretend to build a vector $\eta \in \mathcal{N}$ by making sure that η is a *k*th degree polynomial on the pieces. In step 5, we pretend to set the values of η in a piece p. The number of new entries in η required to determine a polynomial on piece p is shown in 5(b). Once the new entries are picked arbitrarily, all the values on the piece are determined via the constraints in $D_{-A}\eta = 0$. Then we propagate the values from this piece to other adjoining pieces in a depth-first fashion. By the end of the procedure, df accumulates the total number of free parameters that we can use to build such a η . The dimension of \mathcal{N} is equal to the number of free parameters in the algorithm.

Time Complexity. It takes O(nd(k+1)) time to make the polynomial pieces in line 3 of DEGREES-OF-FREEDOM. The number of pieces is at most nd(k+1). Therefore the for loop in line 5 is run at most as many times. SPREAD is called on a piece exactly once and SPREAD-VERTEX is called on a node exactly once. A node is contained in a maximum of (k+2)d pieces. Therefore, the total time complexity is O(nd(k+2)).

Correctness. Suppose the number of free parameters returned by the algorithm is f. Given the values at the f free nodes $F \subset [n]$, the values are determined at all n nodes. Further this mapping from $\mathbb{R}^f \mapsto \mathbb{R}^n$ is linear. Therefore there

exists a matrix C with size $n \times f$ such that $Cb \in \mathcal{N}$ for any $b \in \mathbb{R}^f$. Further, $(Cb)_F$ is a permutation of b, because the values at free nodes are not modified by C. Therefore, there are f rows in C corresponding to the free nodes F, which when vertically stacked together form a permutation of $f \times f$ identity matrix. Therefore, the column span of C has dimension f. Hence $f \leq \dim(\mathcal{N})$. Conversely, consider any $\eta \in \mathcal{N}$. Given the entries b of η at the locations of free parameters, then the rest of the entries of η are determined by $\eta = Cb$. Therefore η must lie in the column span of C. Therefore $f = \dim(\mathcal{N})$.

E More details on optimization algorithms

Generic Quadratic Programming on the Dual Recall that KTF solves the following convex optimization problem:

$$\hat{\theta} = \underset{\theta \in \mathbb{R}^n}{\operatorname{argmin}} \ \frac{1}{2} \|y - \theta\|_2^2 + \lambda \|D\theta\|_1.$$
(87)

with $D = D_{n,d}^{(k+1)}$. The corresponding Lagrange dual problem is

$$\max_{u} -\frac{1}{2}uDD^{\mathsf{T}}u + y^{\mathsf{T}}D^{\mathsf{T}}u$$
subject to $-\lambda \le u \le \lambda$.
(88)

Note that the dual problem is a standard quadratic program (QP) and can be solved using the interior point method (IPM) to high precision. Then the primal solution can be constructed using $\hat{\theta} = y - D^{\mathsf{T}}u^*$ using the optimal solution u^* . In practice, IPM takes only a few iterations to converge⁴, but each iteration involves solving a linear system. This linear system is sparse since D is sparse — it has only O(dkn) non-zero elements. However, the condition number of the linear system grows as the weights on the barrier function increase, which makes it difficult to exploit the sparsity using methods such as preconditioned conjugates gradient method. On the other hand, direct solvers such as Gaussian elimination and Cholesky decomposition can take up to $O(n^3)$. Sometimes this can be improved by exploiting the banded structure of the linear system, we will describe a particular version of the interior point method using logarithmic-barrier function.

Primal-Dual Interior Point method The primal-dual version of the interior point solver proposed by (Kim et al., 2009) for ℓ_1 trend filtering can be straightforwardly applied to any generalized lasso problem, including KTF. The main idea is to trace a "central path" using Newton's method with an increasing weights *t* on the logarithmic barrier functions. The computation is dominated by computing the search direction of the Newton step, which boils down to solving the following system of linear equations

$$\begin{bmatrix} DD^{\mathsf{T}} & I & -I \\ I & J_1 & 0 \\ -I & 0 & J_2 \end{bmatrix} \begin{bmatrix} \Delta u \\ \Delta \mu_1 \\ \Delta \mu_2 \end{bmatrix} = -\begin{bmatrix} DD^T u - Dy + \mu_1 - \mu_2 \\ f_1 + (1/t)\mu_1^{-1} \\ f_2 + (1/t)\mu_2^{-1} \end{bmatrix}$$
(89)

where $\mu_1, \mu_2 \in \mathbb{R}^m$ are the dual variables of the dual problem (88), $f_1 = u - \lambda \mathbf{1}$, $f_2 = -u - \lambda \mathbf{1}$, $J_i = \text{diag}\mu_i^{-1}\text{diag}(f_i)$ are diagonal matrices and μ_i^{-1} denotes entrywise inversion. Following the derivation of (Kim et al., 2009), we can further eliminate $\Delta \mu_1$ and $\Delta \mu_2$ and solve a linear system of the form

$$(DD^{\mathsf{T}} - J_1^{-1}J_2^{-1})\Delta u = -(DD^{\mathsf{T}}u - D^{(k+1)}y - (1/t)f_1^{-1} + (1/t)f_2^{-1}).$$
(90)

and then construct the remainder of the solutions using

$$\Delta \mu_1 = -(\mu_1 + (1/t)f_1^{-1} + J_1^{-1}\Delta u),$$

$$\Delta \mu_2 = -(\mu_2 + (1/t)f_2^{-1} - J_2^{-1}\Delta u).$$

⁴In theory it could take up to $O(n^{1/2})$ iterations to converge.

Unlike in the trend filtering problems in 1D where (90) is a banded linear system with a bandwidth at most 2k + 3, in a *d*-dimensional grid, the linear system is the following:

$$\begin{bmatrix} \left(D_{1d}D_{1d}^{\mathsf{T}}\right)\otimes I\otimes\ldots\otimes I, & D_{1d}\otimes D_{1d}^{\mathsf{T}}\otimes I\otimes\ldots\otimes I, & \cdots, & D_{1d}\otimes I\otimes\ldots\otimes I\otimes D_{1d}^{\mathsf{T}}, \\ D_{1d}^{\mathsf{T}}\otimes D_{1d}\otimes I\otimes\ldots\otimes I, & I\otimes\left(D_{1d}D_{1d}^{\mathsf{T}}\right)\otimes I\otimes\ldots\otimes I, & \cdots, & I\otimes D_{1d}\otimes I\otimes\ldots\otimes I\otimes D_{1d}^{\mathsf{T}}, \\ \vdots & \vdots & \ddots & \vdots \\ D_{1d}^{\mathsf{T}}\otimes I\otimes\ldots\otimes I\otimes D_{1d} & I\otimes D_{1d}^{\mathsf{T}}\otimes I\otimes\ldots\otimes I\otimes D_{1d} & \cdots, & I\otimes\ldots\otimes I\otimes\left(D_{1d}D_{1d}^{\mathsf{T}}\right) \end{bmatrix} - J_{1}^{-1}J_{2}^{-1}$$

where $D_{1d} = D_{N,1}^{(k+1)}$. This is still sparse, structured, but the bandedness is on the order of $O(n^{1-1/d} + k^2)$. Moreover, the above matrix is not full rank, and the condition number of the linear system blows up as the dual variables μ_1, μ_2 converges to 0 with $t \to \infty$.

Proximal Dykstra's algorithm Proximal Dykstra's algorithm is an operator-splitting method for solving problems of the form

$$\underset{\theta \in \mathbb{R}^n}{\text{minimize}} \ \frac{1}{2} \|y - \theta\|_2^2 + r_1(\theta) + r_2(\theta) + \dots + r_d(\theta)$$
(91)

where $r_1, ..., r_d$ are convex but possibly non-smooth functions. We can clearly see that the regularizer in KTF decomposes into this form with

$$r_i(\theta) = \sum_{j_1, j_2, \dots, j_{i-1}, j_{i+1}, \dots, j_d} \|\tilde{D}_{1d}^{(k+1)} \theta[j_1, \dots, j_{i-1}, \vdots, j_{i+1}, \dots, j_d]\|_1$$

where we reshape θ into the *d*-dimensional tensor format and use the "numpy.ndarray" notation.

The proximal Dykstra algorithm (see, e.g., Tibshirani, 2017) initializes $\theta^{(0)} = y, z^{(-d+1)} = ... = z^{(0)} = 0$ and then iteratively applies the following update rule for t = 1, 2, 3, ...:

$$\begin{split} \theta^{(t)} &= \mathsf{prox}_{r_t \bmod d+1} (\theta^{(t-1)} + z^{(t-d)}) \\ z^{(t)} &= \theta^{(t-1)} + z^{(t-d)} - \theta^{(t)}. \end{split}$$

where $\cdot \mod \cdot$ is the modulo operator, and the proximal operator

$$\operatorname{prox}_{r}(u) = \operatorname{argmin}_{\theta} \frac{1}{2} \|u - \theta\|^{2} + r(\theta).$$

Note that this is equivalent to a cyclic block coordinate descent in the dual.

For KTF, each proximal problem can be parallelized (Barbero and Sra, 2014). Specifically, on a *d*-dim regular grid, the proximal operator of r_i further splits into solving $O(n^{1-1/d})$ 1D-trend filters of size $n^{1/d}$ in parallel. Each subproblem can be solved efficiently in $O(n^{1.5/d})$ time with the primal-dual interior point method for $k \ge 1$ (Tibshirani, 2014) and in linear time when k = 0 using dynamic programming (Johnson, 2013).

Douglas-Rachford Splitting Another operator-splitting method for solving KTF is through the Douglas-Rachford (DR) algorithm (Eckstein and Bertsekas, 1992). For simplicity, we will focus our discussion on the case of 2D grids. The DR algorithm generically solves the following unconstrained problem:

$$\min_{\theta} \inf_{\theta} f(\theta) + g(\theta) \tag{92}$$

for convex functions f, g. The update rules include initializing an auxiliary variable $z^{(0)} = y$ and applying the following for t = 0, 1, 2, ...:

$$\begin{split} \theta^{(t+1)} &= \mathrm{prox}_f(z^{(t)}) \\ z^{(t+1)} &= z^{(t)} + \mathrm{prox}_a(2\theta^{(t+1)} - z^{(t)}) - \theta^{(t+1)}. \end{split}$$

There are multiple ways of applying this to our problem. We apply the DR algorithm to the dual of the following reformulation according to (Barbero and Sra, 2014, Algorithm 9):

$$\underset{\theta \in \mathbb{R}^n}{\text{minimize}} \ \frac{1}{2} \|\theta\|_2^2 + \left(\lambda \|\tilde{D}_{1d}^{(k+1)} \otimes I\theta\|_1 - \langle\theta, y\rangle\right) + \left(\lambda \|I \otimes \tilde{D}_{1d}^{(k+1)}\theta\|_1\right).$$
(93)

We refer interested readers to (Barbero and Sra, 2014) for the derivation of the dual and the conversion of the problem into one that resembles (92). Ultimately, the proximal operator of the conjugate function (an indicator on a certain polytope) can be evaluated using the proximal operator of the r_1 and r_2 as in the proximal Dykstra updates via the Moreau decomposition:

$$\operatorname{prox}_{r_i}(u) + \operatorname{prox}_{r_i^*}(u) = u.$$

In other words, the Douglas-Rachford algorithm enjoys the same computational benefits of the proximal Dykstra's algorithm as each proximal operator evaluation involves only solving 1D trend filtering problems in parallel.

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