# INTRINSIC RIEMANNIAN FUNCTIONAL DATA ANALYSIS 

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#### Abstract

In this work we develop a novel and foundational framework for analyzing general Riemannian functional data, in particular a new development of tensor Hilbert spaces along curves on a manifold. Such spaces enable us to derive Karhunen-Loève expansion for Riemannian random processes. This framework also features an approach to compare objects from different tensor Hilbert spaces, which paves the way for asymptotic analysis in Riemannian functional data analysis. Built upon intrinsic geometric concepts such as vector field, Levi-Civita connection and parallel transport on Riemannian manifolds, the developed framework applies to not only Euclidean submanifolds but also manifolds without a natural ambient space. As applications of this framework, we develop intrinsic Riemannian functional principal component analysis (iRFPCA) and intrinsic Riemannian functional linear regression (iRFLR) that are distinct from their traditional and ambient counterparts. We also provide estimation procedures for iRFPCA and iRFLR, and investigate their asymptotic properties within the intrinsic geometry. Numerical performance is illustrated by simulated and real examples.


1. Introduction. Functional data analysis (FDA) advances substantially in the past two decades, as the rapid development of modern technology enables collecting more and more data continuously over time. There is rich literature spanning more than seventy years on this topic, including development on functional principal component analysis such as Rao (1958); Kleffe (1973); Dauxois, Pousse and Romain (1982); Silverman (1996); Yao, Müller and Wang (2005a); Hall and Hosseini-Nasab (2006); Zhang and Wang (2016), and advances on functional linear regression such as Yao, Müller and Wang (2005b); Hall and Horowitz (2007); Yuan and Cai (2010); Kong et al. (2016), among many others. For a thorough review of the topic, we refer readers to the review article Wang, Chiou and Müller (2016) and monographs Ramsay and Silverman (2005); Ferraty and Vieu (2006); Hsing and Eubank (2015); Kokoszka and Reimherr (2017) for comprehensive treatments on classic functional data analysis. Although traditionally functional data take values in a vector space, more data of nonlinear structure arise and should be properly handled in a nonlinear space. For instance, trajectories of bird migration are naturally regarded as curves on a sphere which is a nonlinear Riemannian manifold, rather than the three-dimensional vector space $\mathbb{R}^{3}$. Another example is the dynamics of brain functional connectivity. The functional connectivity at a time point is represented by a symmetric positive-definite matrix (SPD). Then the dynamics shall be modelled as a curve in the space of SPDs that is endowed with either the affine-invariant metric (Moakher, 2005) or the Log-Euclidean metric (Arsigny et al., 2007) to avoid the "swelling" effect (Arsigny et al., 2007). Both metrics turn SPD into a nonlinear Riemannian manifold. In this paper, we refer

[^0]this type of functional data as Riemannian functional data, which are functions taking values on a Riemannian manifold and modelled by Riemannian random processes, i.e., we treat Riemannian trajectories as realizations of a Riemannian random process.

Analysis of Riemannian functional data is not only challenged by the infinite dimensionality and compactness of covariance operator from functional data, but also obstructed by the nonlinearity of the range of functions, since manifolds are generally not vector spaces and render many techniques relying on linear structure ineffective or inapplicable. For instance, if the sample mean curve is computed for bird migration trajectories as if they were sampled from the ambient space $\mathbb{R}^{3}$, this naïve sample mean in general does not fall on the sphere of earth. For manifolds of tree-structured data studied in Wang and Marron (2007), as they are naturally not Euclidean submanifolds which refer to Riemannian submanifolds of a Euclidean space in this paper, the naïve sample mean can not even be defined from ambient spaces, and thus a proper treatment of manifold structure is necessary. While the literature for Euclidean functional data is abundant, works involving nonlinear manifold structure are scarce. Chen and Müller (2012) and Lin and Yao (2017) respectively investigate representation and regression for functional data living in a low-dimensional nonlinear manifold that is embedded in an infinite-dimensional space, while Lila and Aston (2017) focuses principal component analysis on functional data whose domain is a two-dimensional manifold. None of these deal with functional data that take values on a nonlinear manifold, while Dai and Müller (2017) is the only endeavor in this direction for Euclidean submanifolds.

As functional principal component analysis (FPCA) is an essential tool for FDA, it is of importance and interest to develop this notion for Riemannian functional data. Since manifolds are in general not vector spaces, classic covariance functions/operators do not exist naturally for a Riemannian random process. A strategy that is often adopted, e.g., Shi et al. (2009) and Cornea et al. (2017), to overcome the lack of vectorial structure is to map data on the manifold into tangent spaces via Riemannian logarithm map defined in Section 2.2. As tangent spaces at different points are different vector spaces, in order to handle observations from different tangent spaces, some existing works assume a Euclidean ambient space for the manifold and identify tangent vectors as Euclidean vectors. This strategy is adopted by Dai and Müller (2017) on Riemannian functional data such as compositional data modelled on the unit sphere for the first time. Specifically, they assume that functional data are sampled from a time-varying geodesic submanifold, where at a given time point, the functions take values on a geodesic submanifold of a common manifold. Such a common manifold is further assumed to be a Euclidean submanifold that allows to identify all tangent spaces as hyperplanes in a common Euclidean space (endowed with the usual Euclidean inner product). Then, with the aid of Riemannian logarithm map, Dai and Müller (2017) are able to transform Riemannian functional data into Euclidean ones while accounting for the intrinsic curvature of the underlying manifold.

To avoid confusion, we distinguish two different perspectives to deal with Riemannian manifolds. One is to work with the manifold under consideration without assuming an ambient space surrounding it or an isometric embedding into a Euclidean space. This perspective is regarded as completely intrinsic, or simply intrinsic. Although generally difficult to work with, it can fully respect all geometric structure of the manifold. The other one, referred to as ambient here, assumes that the manifold under consideration is isometrically embedded in a Euclidean ambient space, so that geometric objects such as tangent vectors can be processed within the ambient space. For example, from this point of view, the local polynomial regression for SPD proposed by Yuan et al. (2012) is intrinsic, while the aforementioned work by Dai and Müller (2017) takes the ambient perspective.


Figure 1. Left panel: illustration of ambient movement of tangent vectors. The tangent vector $v_{0}$ at the point $Q$ of a unit circle embedded in a Euclidean plane is moved to the point $P_{1}$ and $P_{2}$ within the ambient space. $v_{1}$ (resp. $v_{2}$ ) is a tangent vector at $P_{1}$ (resp. $P_{2}$ ). The differences $v_{1}-v_{0}$ and $v_{2}-v_{0}$ are not tangent to the circle at $P_{1}$ and $P_{2}$, respectively. If $v_{0}, v_{1}$ and $v_{2}$ have the same length, then the intrinsic parallel transport of $v_{0}$ to $P_{k}$ shall coincide with $v_{k}$, and $\mathcal{P} v_{0}-v_{k}=0$, where $k=1,2$ and $\mathcal{P}$ represents the parallel transport on the unit circle with the canonical metric tensor. Thus, $\left\|\mathcal{P} v_{0}-v_{k}\right\|_{\mathbb{R}^{2}}=0$. However, $\left\|v_{0}-v_{k}\right\|_{\mathbb{R}^{2}}>0$, and this non-zero value completely results from the departure of the Euclidean geometry from the unit circle geometry. The ambient discrepancy $\left\|v_{0}-v_{1}\right\|_{\mathbb{R}^{2}}$ is small as $P_{1}$ is close to $P$, while $\left\|v_{0}-v_{2}\right\|_{\mathbb{R}^{2}}$ is large since $P_{2}$ is far away from $Q$. Right panel: illustration of parallel transport. A tangent vector $v_{1}$ at the point $p_{1}$ on the unit sphere is parallelly transported to the point $p_{2}$ and $p_{3}$ along curves $C_{1}$ and $C_{2}$, respectively. During parallel transportation, the transported tangent vector always stays within the tangent spaces along the curve.

Although it is possible to account for some of geometric structure in the ambient perspective, e.g., the curved nature of manifold via Riemannian logarithm map, several issues arise due to manipulation of geometric objects such as tangent vectors in the ambient space. First, the essential dependence on an ambient space restricts potential applications. It is not immediately applicable to manifolds that are not a Euclidean submanifold or do not have a natural isometric embedding into a Euclidean space, e.g., the Riemannian manifold of $p \times p(p \geq 2)$ SPD matrices endowed with the affine-invariant metric (Moakher, 2005) which is not compatible with the $p(p+1) / 2-$ dimensional Euclidean metric. Second, although an ambient space provides a common stage for tangent vectors at different points, operation on tangent vectors from this ambient perspective can potentially violate the intrinsic geometry of the manifold. To illustrate this, consider comparison of two tangent vectors at different points (this comparison is needed in the asymptotic analysis of Section 3.2; see also Section 2.4). From the ambient perspective, taking difference of tangent vectors requires moving a tangent vector parallelly within the ambient space to the base point of the other tangent vector. However, the resultant tangent vector after movement in the ambient space is generally not a tangent vector for the base point of the other tangent vector; see the left panel of Figure 1 for a geometric illustration. In another word, the ambient difference of two tangent vectors at different points is not an intrinsic geometric object on the manifold, and the departure from intrinsic geometry can potentially affect the statistical efficacy and/or efficiency. Lastly, since manifolds might be embedded into more than one ambient space, the interpretation of statistical results crucially depends on the ambient space and could be misleading if one does not choose the ambient space appropriately.

In the paper, we develop a completely intrinsic framework that provides a foundational the-
ory for general Riemannian functional data that paves the way for the development of intrinsic Riemannian functional principal component analysis and intrinsic Riemannian functional linear regression, among other potential applications. The key building block is a new concept of tensor Hilbert space along a curve on the manifold, which is described in Section 2. On one hand, our approach experiences dramatically elevated technical challenge relative to the ambient counterparts. For example, without an ambient space, it is nontrivial to perceive and handle tangent vectors. On the other hand, the advantages of the intrinsic perspective are at least three-fold, in contrast to ambient approaches. First, our results immediately apply to many important Riemannian manifolds that are not naturally a Euclidean submanifold but commonly seen in statistical analysis and machine learning, such as the aforementioned SPD manifolds and Grassmannian manifolds. Second, our framework features a novel intrinsic proposal for coherent comparison of objects from different tensor Hilbert spaces on the manifold, and hence makes the asymptotic analysis sensible. Third, results produced by our approach are invariant to embeddings and ambient spaces, and can be interpreted independently, which avoid potential misleading interpretation in practice.

As important applications of the proposed framework, we develop intrinsic Riemannian functional principal component analysis (iRFPCA) and intrinsic Riemannian functional linear regression (iRFLR). Specifically, estimation procedures of intrinsic eigenstructure are provided and their asymptotic properties are investigated within the intrinsic geometry. For iRFLR, we study a Riemannian functional linear regression model, where a scalar response intrinsically and linearly depends on a Riemannian functional predictor through a Riemannian slope function, a concept that is formulated in Section 4, along with the concept of linearity in the context of Riemannian functional data. We present an FPCA-based estimator and a Tikhonov estimator for the Riemannian slope function and explore their asymptotic properties, where the proposed framework of tensor Hilbert space again plays an essential role.

The rest of the paper is structured as follows. The foundational framework for intrinsic Riemannian functional data analysis is laid in Section 2. Intrinsic Riemannian functional principal component analysis is presented in Section 3, while intrinsic Riemannian functional regression is studied in Section 4. In Section 5, numerical performance is illustrated through simulations, and an application to Human Connectome Project analyzing functional connectivity and behavioral data is provided.
2. Tensor Hilbert Space and Riemannian Random Process. In this section, we first define the concept of tensor Hilbert space and discuss its properties, including a mechanism to deal with two different tensor Hilbert spaces at the same time. Then, random elements on tensor Hilbert space are investigated, with the proposed intrinsic Karhunen-Loève expansion for the random elements. Finally, practical computation with respect to an orthonormal frame is given. Throughout this section, we assume a $d$-dimensional, connected and geodesically complete Riemannian manifold $\mathcal{M}$ equipped with a Riemannian metric $\langle\cdot, \cdot\rangle$, which defines a scalar product $\langle\cdot, \cdot\rangle_{p}$ for the tangent space $T_{p} \mathcal{M}$ at each point $p \in \mathcal{M}$. This metric also induces a distance function $d_{\mathcal{M}}$ on $\mathcal{M}$. A preliminary for Riemannian manifolds can be found in the appendix. For a comprehensive treatment on Riemannian manifolds, we recommend the introductory text by Lee (1997) and also Lang (1995).
2.1. Tensor Hilbert Spaces along Curves. Let $\mu$ be a measurable curve on a manifold $\mathcal{M}$ and parameterized by a compact domain $\mathcal{T} \subset \mathbb{R}$ equipped with a finite measure $v$. A vector field $V$ along $\mu$ is a map from $\mathcal{T}$ to the tangent bundle $T \mathcal{M}$ such that $V(t) \in T_{\mu(t)} \mathcal{M}$ for all $t \in \mathcal{T}$. It is seen that the collection of vector fields $V$ along $\mu$ is a vector space, where the vector addition between two vector fields $V_{1}$ and $V_{2}$ is a vector field $U$ such that $U(t)=V_{1}(t)+V_{2}(t)$ for all $t \in \mathcal{T}$,
and the scalar multiplication between a real number $a$ and a vector field $V$ is a vector field $U$ such that $U(t)=a V(t)$ for all $t \in \mathcal{T}$. Let $\mathscr{T}(\mu)$ be the collection of (equivalence classes of) measurable vector fields $V$ along $\mu$ such that $\|V\|_{\mu}:=\left\{\int_{\mathcal{T}}\langle V(t), V(t)\rangle_{\mu(t)} \mathrm{d} v(t)\right\}^{1 / 2}<\infty$ with identification between $V$ and $U$ in $\mathscr{T}(\mu)$ (or equivalently, $V$ and $U$ are in the same equivalence class) when $v(\{t \in \mathcal{T}: V(t) \neq U(t)\})=0$. Then $\mathscr{T}(\mu)$ is turned into an inner product space by the inner product $\langle V, U\rangle_{\mu}:=\int_{\mathcal{T}}\langle V(t), U(t)\rangle_{\mu(t)} \mathrm{d} v(t)$, with the induced norm given by $\|\cdot\|_{\mu}$. Moreover, we have that

## Theorem 1. For a measurable curve $\mu$ on $\mathcal{M}, \mathscr{T}(\mu)$ is a separable Hilbert space.

We call the space $\mathscr{T}(\mu)$ the tensor Hilbert space along $\mu$, as tangent vectors are a special type of tensor and the above Hilbertian structure can be defined for tensor fields along $\mu$. The above theorem is of paramount importance, in the sense that it suggests $\mathscr{T}(\mu)$ to serve as a cornerstone for Riemannian functional data analysis for two reasons. First, as shown in Section 2.2, via Riemannian logarithm maps, a Riemannian random process may be transformed into a tangent-vector-valued random process (called log-process in Section 2.2) that can be regarded as a random element in a tensor Hilbert space. Second, the rigorous theory of functional data analysis formulated in Hsing and Eubank (2015) by random elements in separable Hilbert spaces fully applies to the log-process.

One distinct feature of the tensor Hilbert space is that, different curves that are even parameterized by the same domain give rise to different tensor Hilbert spaces. In practice, one often needs to deal with two different tensor Hilbert spaces at the same time. For example, in the next subsection we will see that under some conditions, a Riemannian random process $X$ can be conceived as a random element on the tensor Hilbert space $\mathscr{T}(\mu)$ along the intrinsic mean curve $\mu$. However, the mean curve is often unknown and estimated from a random sample of $X$. Since the sample mean curve $\hat{\mu}$ generally does not agree with the population one, quantities of interest such as covariance operator and their sample versions are defined on two different tensor Hilbert spaces $\mathscr{T}(\mu)$ and $\mathscr{T}(\hat{\mu})$, respectively. For statistical analysis, one needs to compare the sample quantities with their population counterparts and hence involves objects such as covariance operators from two different tensor Hilbert spaces.

In order to intrinsically quantify the discrepancy between objects of the same kind from different tensor Hilbert spaces, we utilize the Levi-Civita connection (p. 18, Lee 1997) associated with the Riemannian manifold $\mathcal{M}$. The Levi-Civita connection is uniquely determined by the Riemannian structure. It is the only torsion-free connection compatible with the Riemannian metric. Associated with this connection is a unique parallel transport operator $\mathcal{P}_{p, q}$ that smoothly transports tangent vectors at $p$ along a curve to $q$ and preserves the inner product. We shall emphasize that, the parallel transportation is performed intrinsically. For instance, tangent vectors being transported always stay tangent to the manifold during transportation, which is illustrated by the right panel of Figure 1. Although transport operator $\mathcal{P}_{p, q}$ depends on the curve connecting $p$ and $q$, there exists a canonical choice of the curve connecting two points, which is the minimizing geodesic between $p$ and $q$ (under some conditions, almost surely the minimizing geodesic is unique between two points randomly sampled from the manifold). The smoothness of parallel transport also implies that if $p$ and $q$ are not far apart, then the initial tangent vector and the transported one stays close (in the space of tangent bundle endowed with the Sasaki metric (Sasaki, 1958)). This feature is desirable for our purpose, as when sample mean $\hat{\mu}(t)$ approaches to $\mu(t)$, one expects a tangent vector at $\hat{\mu}(t)$ converges to its transported version at $\mu(t)$. Owing to these nice properties of parallel transport, it becomes an ideal tool to construct a mechanism of comparing objects from different tensor Hilbert spaces as follows.

Suppose $f$ and $h$ are two measurable curves on $\mathcal{M}$ defined on $\mathcal{T}$. Let $\gamma_{t}(\cdot):=\gamma(t, \cdot)$ be a family of smooth curves that is parameterized by the interval $[0,1]$ (the way of parameterization here does not matter) and connects $f(t)$ to $h(t)$, i.e., $\gamma_{t}(0)=f(t)$ and $\gamma_{t}(1)=h(t)$, such that $\gamma(\cdot, s)$ is measurable for all $s \in[0,1]$. Suppose $v \in T_{f(t)} \mathcal{M}$ and let $V$ be a smooth vector field along $\gamma_{t}$ such that $\nabla \dot{\gamma} V=0$ and $V(0)=v$, where $\nabla$ denotes the Levi-Civita connection of the manifold $\mathcal{M}$. The theory of Riemannian manifolds shows that such a vector field $V$ uniquely exists. This gives rise to the parallel transporter $\mathcal{P}_{f(t), h(t)}: T_{f(t)} \mathcal{M} \rightarrow T_{h(t)} \mathcal{M}$ along $\gamma_{t}$, defined by $\mathcal{P}_{f(t), h(t)}(v)=V(1)$. In other words, $\mathcal{P}_{f(t), h(t)}$ parallelly transports $v$ to $V(1) \in T_{h(t)} \mathcal{M}$ along the curve $\gamma_{t}$. As the parallel transporter determined by the Levi-Civita connection, $\mathcal{P}$ preserves the inner product of tangent vectors along transportation, i.e. $\langle u, v\rangle_{f(t)}=\left\langle\mathcal{P}_{f(t), h(t)} u, \mathcal{P}_{f(t), h(t)} v\right\rangle_{h(t)}$ for $u, v \in T_{f(t)} \mathcal{M}$. Then we can define the parallel transport of vector fields from $\mathscr{T}(f)$ to $\mathscr{T}(h)$, denoted by $\Gamma_{f, h}$, $\left(\Gamma_{f, h} U\right)(t)=\mathcal{P}_{f(t), h(t)}(U(t))$ for all $U \in \mathscr{T}(f)$ and $t \in \mathcal{T}$. One immediately sees that $\Gamma_{f, h}$ is a linear operator on tensor Hilbert space. Its adjoint, denoted by $\Gamma_{f, h}^{*}$, is a map from $\mathscr{T}(h)$ to $\mathscr{T}(f)$ and is given by $\left\langle U, \Gamma_{f, h}^{*} V\right\rangle_{f}=\left\langle\left\langle\Gamma_{f, h} U, V\right\rangle_{h}\right.$ for $U \in \mathscr{T}(f)$ and $V \in \mathscr{T}(h)$. Let $\mathscr{C}(f)$ denote the set of all Hilbert-Schmidt operators on $\mathscr{T}(f)$, which is a Hilbert space with the Hilbert-Schmidt norm $\|\cdot\|_{f}$. We observe that the operator $\Gamma_{f, h}$ also gives rise to a mapping $\Phi_{f, h}$ from $\mathscr{C}(f)$ to $\mathscr{C}(h)$, defined by $\left(\Phi_{f, h} \mathcal{A}\right) V=\Gamma_{f, h}\left(\mathcal{A}\left(\Gamma_{f, h}^{*} V\right)\right)$ for $\mathcal{A} \in \mathscr{C}(f)$ and $V \in \mathscr{T}(h)$. The operator $\Phi_{f, h}$ is called the parallel transporter of Hilbert-Schmidt operators. Below are some important properties of $\Gamma_{f, h}$ and $\Phi_{f, h}$, where $\left(x_{1} \otimes x_{2}\right) x_{3}:=\left\langle\left\langle x_{1}, x_{3}\right\rangle_{f} x_{2}\right.$ for $x_{1}, x_{2}, x_{3} \in \mathscr{T}(f)$.

Proposition 2. Suppose $U \in \mathscr{T}(f)$, $V \in \mathscr{T}(h), \mathcal{A} \in \mathscr{C}(f)$ and $\mathcal{B} \in \mathscr{C}(h)$ for two measurable curves $f$ and $h$ on $\mathcal{M}$. Let $\Gamma_{f, h}$ and $\Phi_{f, h}$ be parallel transporters along a family of smooth curves $\gamma_{t}$ defined above, such that $\gamma(t, \cdot)$ is smooth and $\gamma(, s)$ is measurable. Then the following statements regarding $\Gamma_{f, h}$ and $\Phi_{f, h}$ hold.

1. The operator $\Gamma_{f, h}$ is a unitary transformation from $\mathscr{T}(f)$ to $\mathscr{T}(h)$.
2. $\Gamma_{f, h}^{*}=\Gamma_{h, f}$. Also, $\left\|\Gamma_{f, h} U-V\right\|_{h}=\left\|U-\Gamma_{h, f} V\right\|_{f}$.
3. $\Gamma_{f, h}(\mathcal{A} U)=\left(\Phi_{f, h} \mathcal{A}\right)\left(\Gamma_{f, h} U\right)$.
4. If $\mathcal{A}$ is invertible, then $\Phi_{f, h} \mathcal{A}^{-1}=\left(\Phi_{f, h} \mathcal{A}\right)^{-1}$.
5. $\Phi_{f, h} \sum_{k} c_{k} \varphi_{k} \otimes \varphi_{k}=\sum_{k} c_{k}\left(\Gamma_{f, h} \varphi_{k}\right) \otimes\left(\Gamma_{f, h} \varphi_{k}\right)$, where $c_{k}$ are scalar constants, and $\varphi_{k} \in \mathscr{T}(f)$.
6. $\left\|\Phi_{f, h} \mathcal{A}-\mathcal{B}\right\|_{h}=\left\|\mathcal{A}-\Phi_{h, f} \mathcal{B}\right\|_{f}$.

We define $U \ominus_{\Gamma} V:=\Gamma_{f, h} U-V$ for $U \in \mathscr{T}(f)$ and $V \in \mathscr{T}(h)$, and $\mathcal{A} \ominus_{\Phi} \mathcal{B}:=\Phi_{f, h} \mathcal{A}-\mathcal{B}$ for operators $\mathcal{A}$ and $\mathcal{B}$. To quantify the discrepancy between an element $U$ in $\mathscr{T}(f)$ and another one $V$ in $\mathscr{T}(h)$, we can use the quantity $\left\|U \ominus_{\Gamma} V\right\|_{h}$. Similarly, we adopt $\left\|\mathcal{A} \ominus_{\Phi} \mathcal{B}\right\|_{h}$ as discrepancy measure for two covariance operators $\mathcal{A}$ and $\mathcal{B}$. These quantities are intrinsic as they are built on intrinsic geometric concepts. In light of Proposition 2, they are symmetric under the parallel transport, i.e., transporting $\mathcal{A}$ to $\mathcal{B}$ yields the same discrepancy measure as transporting $\mathcal{B}$ to $\mathcal{A}$. We also note that, when $\mathcal{M}=\mathbb{R}^{d}$, the difference operators $\theta_{\Gamma}$ and $\theta_{\Phi}$ reduce to the regular vector and operator difference, i.e., $U \ominus_{\Gamma} V$ becomes $U-V$, while $\mathcal{A} \ominus_{\Phi} \mathcal{B}$ becomes $\mathcal{A}-\mathcal{B}$. Therefore, $\ominus_{\Gamma}$ and $\ominus_{\Phi}$ can be viewed as generalization of the regular vector and operator difference to a Riemannian setting. One shall note that $\Gamma$ and $\Phi$ depend on the choice of the family of curves $\gamma$, a canonical choice of which is discussed in Section 3.2.
2.2. Random Elements on Tensor Hilbert Spaces. Let $X$ be a Riemannian random process. In order to introduce the concept of intrinsic mean function for $X$, we define a family of functions
indexed by $t$ :

$$
\begin{equation*}
F(p, t)=\mathbb{E} d_{\mathcal{M}}^{2}(X(t), p), \quad p \in \mathcal{M}, t \in \mathcal{T} \tag{1}
\end{equation*}
$$

For a fixed $t$, if there exists a unique $q \in \mathcal{M}$ that minimizes $F(p, t)$ over all $p \in \mathcal{M}$, then $q$ is called the intrinsic mean (also called Fréchet mean) at $t$, denoted by $\mu(t)$, i.e.,

$$
\mu(t)=\underset{p \in \mathcal{M}}{\arg \min } F(p, t) .
$$

As required for intrinsic analysis, we assume the following condition.

## A. $1 \quad$ The intrinsic mean function $\mu$ exists.

We refer readers to Bhattacharya and Patrangenaru (2003) and Afsari (2011) for conditions under which the intrinsic mean of a random variable on a general manifold exists and is unique. For example, according to Cartan-Hadamard theorem, if the manifold is simply connected and complete with non-positive sectional curvature, then intrinsic mean function always exists and is unique as long as for all $t \in \mathcal{T}, F(p, t)<\infty$ for some $p \in \mathcal{M}$.

Since $\mathcal{M}$ is geodesically complete, by Hopf-Rinow theorem (p. 108, Lee 1997), its exponential map $\operatorname{Exp}_{p}$ at each $p$ is defined on the entire $T_{p} \mathcal{M}$. As $\operatorname{Exp}_{p}$ might not be injective, in order to define its inverse, we restrict $\operatorname{Exp}_{p}$ to a subset of the tangent space $T_{p} \mathcal{M}$. Let $\operatorname{Cut}(p)$ denote the set of all tangent vectors $v \in T_{p} \mathcal{M}$ such that the geodesic $\gamma(t)=\operatorname{Exp}_{p}(t v)$ fails to be minimizing for $t \in[0,1+\epsilon)$ for each $\epsilon>0$. Now, we define $\operatorname{Exp}_{p}$ only on $\mathscr{D}_{p}:=T_{p} \mathcal{M} \backslash \operatorname{Cut}(p)$. The image of $\operatorname{Exp}_{p}$, denoted by $\operatorname{Im}\left(\operatorname{Exp}_{p}\right)$, consists of points $q$ in $\mathcal{M}$, such that $q=\operatorname{Exp}_{p} v$ for some $v \in \mathscr{D}_{p}$. In this case, the inverse of $\operatorname{Exp}_{p}$ exists and is called Riemannian logarithm map, which is denoted by $\log _{p}$ and maps $q$ to $v$. We shall make the following assumption.

## A. $2 \operatorname{Pr}\left\{\forall t \in \mathcal{T}: X(t) \in \operatorname{Im}\left(\operatorname{Exp}_{\mu(t)}\right)\right\}=1$.

Then, $\log _{\mu(t)} X(t)$ is almost surely defined for all $t \in \mathcal{T}$. The condition is superfluous if $\operatorname{Exp}_{\mu(t)}$ is injective for all $t$, like the manifold of $m \times m$ SPDs endowed with the affine-invariant metric.

In the sequel we shall assume $X$ satisfies conditions A. 1 and A.2. An important observation is that the log-process $\left\{\log _{\mu(t)} X(t)\right\}_{t \in \mathcal{T}}$ (denoted by $\log _{\mu} X$ for short) is a random vector field along $\mu$. If we assume continuity for the sample paths of $X$, then the process $\log _{\mu} X$ is measurable with respect to the product $\sigma$-algebra $\mathscr{B}(\mathcal{T}) \times \mathscr{E}$ and the Borel algebra $\mathscr{B}(T \mathcal{M})$, where $\mathscr{E}$ is the $\sigma$-algebra of the probability space. Furthermore, if $\mathbb{E}\left\|\log _{\mu} X\right\|_{\mu}^{2}<\infty$, then according to Theorem 7.4.2 of Hsing and Eubank (2015), $\log _{\mu} X$ can be viewed as a tensor Hilbert space $\mathscr{T}(\mu)$ valued random element. Observing that $\mathbb{E} \log _{\mu} X=0$ according to Theorem 2.1 of Bhattacharya and Patrangenaru (2003), the intrinsic covariance operator for $\log _{\mu} X$ is given by $\mathcal{C}=\mathbb{E}\left(\log _{\mu} X \otimes \log _{\mu} X\right)$. This operator is nuclear and self-adjoint. It then admits the following eigendecomposition (Theorem 7.2.6, Hsing and Eubank, 2015)

$$
\begin{equation*}
\mathcal{C}=\sum_{k=1}^{\infty} \lambda_{k} \phi_{k} \otimes \phi_{k} \tag{2}
\end{equation*}
$$

with eigenvalues $\lambda_{1} \geq \lambda_{2} \cdots \geq 0$ and orthonormal eigenelements $\phi_{k}$ that form a complete orthonormal system for $\mathscr{T}(\mu)$. Also, with probability one, the log-process of $X$ has the following Karhunen-Loève expansion

$$
\begin{equation*}
\log _{\mu} X=\sum_{k=1}^{\infty} \xi_{k} \phi_{k} \tag{3}
\end{equation*}
$$

with $\xi_{k}:=\left\langle\left\langle X, \phi_{k}\right\rangle_{\mu}\right.$ being uncorrelated and centered random variables. Therefore, we obtain the intrinsic Riemannian Karhunen-Loève (iRKL) expansion for $X$ given by

$$
\begin{equation*}
X(t)=\operatorname{Exp}_{\mu(t)} \sum_{k=1}^{\infty} \xi_{k} \phi_{k}(t) \tag{4}
\end{equation*}
$$

The elements $\phi_{k}$ are called intrinsic Riemannian functional principal component (iRFPC), while the variables $\xi_{k}$ are called intrinsic iRFPC scores. This result is summarized in the following theorem whose proof is already contained in the above derivation and hence omitted. We shall note that the continuity assumption on sample paths can be weakened to piece-wise continuity.

Theorem 3 (Intrinsic Karhunen-Loève Representation). Assume that $X$ satisfies assumptions A.1 and A.2. If sample paths of $X$ are continuous and $\mathbb{E}\left\|\log _{\mu} X\right\|_{\mu}^{2}<\infty$, then the intrinsic covariance operator $\mathcal{C}=\mathbb{E}\left(\log _{\mu} X \otimes \log _{\mu} X\right)$ of $\log _{\mu} X$ admits the decomposition (2), and the random process $X$ admits the representation (4).

In practice, the series at (4) is truncated at some positive integer $K$, resulting in a truncated intrinsic Riemannian Karhunen-Loève expansion of $X$, given by $X_{K}=\operatorname{Exp}_{\mu} W_{K}$ with $W_{K}=\sum_{k=1}^{K} \xi_{k} \phi_{k}$. The quality of the approximation of $X_{K}$ for $X$ is quantified by $\int_{\mathcal{T}} d_{\mathcal{M}}^{2}\left(X(t), X_{K}(t)\right) \mathrm{d} v(t)$, and can be shown by a method similar to Dai and Müller (2017) that if the manifold has non-negative sectional curvature everywhere, then $\int_{\mathcal{T}} d_{\mathcal{M}}^{2}\left(X(t), X_{K}(t)\right) \mathrm{d} v(t) \leq\left\|\log _{\mu} X-W_{K}\right\|_{\mu}^{2}$. For manifolds with negative sectional curvatures, such inequality in general does not hold. However, for Riemannian random process $X$ that almost surely lies in a compact subset of $\mathcal{M}$, the residual $\int_{\mathcal{T}} d_{\mathcal{M}}^{2}\left(X(t), X_{K}(t)\right) \mathrm{d} v(t)$ can be still bounded by $\left\|\log _{\mu} X-W_{K}\right\|_{\mu}^{2}$ up to a scaling constant.

Proposition 4. Assume that conditions $A .1$ and A.2 hold, and the sectional curvature of $\mathcal{M}$ is bounded from below by $\kappa \in \mathbb{R}$. Let $\mathcal{K}$ be a subset of $\mathcal{M}$. If $\kappa \geq 0$, we let $\mathcal{K}=\mathcal{M}$, and if $\kappa<0$, we assume that $\mathcal{K}$ is compact. Then, for some constant $C>0, d_{\mathcal{M}}(P, Q) \leq \sqrt{C}\left|\log _{O} P-\log _{O} Q\right|$ for all $O, P, Q \in$ $\mathcal{K}$. Consequently, if $X \in \mathcal{K}$ almost surely, then $\int_{\mathcal{T}} d_{\mathcal{M}}^{2}\left(X(t), X_{K}(t)\right) \mathrm{d} v(t) \leq C\left\|\log _{\mu} X-W_{K}\right\|_{\mu}^{2}$.
2.3. Computation in Orthonormal Frames. In practical computation, one might want to work with specific orthonormal bases for tangent spaces. A choice of orthonormal basis for each tangent space constitutes an orthonormal frame on the manifold. In this section, we study the representation of the intrinsic Riemannian Karhunen-Loève expansion under an orthonormal frame and formulas for change of orthonormal frames.

Let $\mathbf{E}=\left(E_{1}, \ldots, E_{d}\right)$ be a continuous orthonormal frame, i.e., each $E_{j}$ is a vector field of $\mathcal{M}$ such that $\left\langle E_{j}(p), E_{j}(p)\right\rangle_{p}=1$ and $\left\langle E_{j}(p), E_{k}(p)\right\rangle_{p}=0$ for $j \neq k$ and all $p \in \mathcal{M}$. At each point $p$, $\left\{E_{1}(p), \ldots, E_{d}(p)\right\}$ form an orthonormal basis for $T_{p} \mathcal{M}$. The coordinate of $\log _{\mu(t)} X(t)$ with respect to $\left\{E_{1}(\mu(t)), \ldots, E_{d}(\mu(t))\right\}$ is denoted by $Z_{\mathbf{E}}(t)$, with the subscript $\mathbf{E}$ indicating its dependence on the frame. The resulting process $Z_{\mathbf{E}}$ is called the $\mathbf{E}$-coordinate process of $X$. Note that $Z_{\mathbf{E}}$ is a regular $\mathbb{R}^{d}$ valued random process defined on $\mathcal{T}$, and classic theory in Hsing and Eubank (2015) applies to $Z_{\mathbf{E}}$. For example, its $\mathcal{L}^{2}$ norm is defined by $\left\|Z_{\mathbf{E}}\right\|_{\mathcal{L}^{2}}=\left\{\mathbb{E} \int_{\mathcal{T}}\left|Z_{\mathbf{E}}(t)\right|^{2} \mathrm{~d} t\right\}^{1 / 2}$, where $|\cdot|$ denotes the canonical norm on $\mathbb{R}^{d}$. One can show that $\left\|Z_{\mathbf{E}}\right\|_{\mathcal{L}^{2}}^{2}=\mathbb{E}\left\|\log _{\mu} X\right\|_{\mu}^{2}$. Therefore, if $\mathbb{E}\left\|\log _{\mu} X\right\|_{\mu}^{2}<\infty$, then the covariance function exists and is $d \times d$ matrix-valued, quantified by $\mathcal{C}_{\mathbf{E}}(s, t)=\mathbb{E}\left\{Z_{\mathbf{E}}(s) Z_{\mathbf{E}}(t)^{T}\right\}$ (Kelly and Root, 1960; Balakrishnan, 1960), noting that $\mathbb{E} Z_{\mathbb{E}}(t)=0$ as $\mathbb{E} \log _{\mu(t)} X(t)=0$ for all $t \in \mathcal{T}$. Also, the vector-valued Mercer's theorem implies the eigendecomposition

$$
\begin{equation*}
\mathcal{C}_{\mathbf{E}}(s, t)=\sum_{k=1}^{\infty} \lambda_{k} \phi_{\mathbf{E}, k}(s) \phi_{\mathbf{E}, k}^{T}(t) \tag{5}
\end{equation*}
$$

with eigenvalues $\lambda_{1} \geq \lambda_{2} \geq \cdots$ and corresponding eigenfunctions $\phi_{\mathbf{E}, k}$. Here, the subscript $\mathbf{E}$ in $\phi_{\mathbf{E}, k}$ is to emphasize the dependence on the chosen frame. One can see that $\phi_{\mathbf{E}, k}$ is a coordinate representation of $\phi_{k}$, i.e., $\phi_{k}=\phi_{\mathbf{E}, k}^{T} \mathbf{E}$.

The coordinate process $Z_{\mathbf{E}}$ admits the vector-valued Karhunen-Loève expansion

$$
\begin{equation*}
Z_{\mathbf{E}}(t)=\sum_{k=1}^{\infty} \xi_{k} \phi_{\mathbf{E}, k}(t) \tag{6}
\end{equation*}
$$

under the assumption of mean square continuity of $Z_{\mathbf{E}}$, according to Theorem 7.3.5 of Hsing and Eubank (2015), where $\xi_{k}=\int_{\mathcal{T}} Z_{\mathbf{E}}^{T}(t) \phi_{\mathbf{E}, k}(t) \mathrm{d} v(t)$. While the covariance function and eigenfunctions of $Z_{\mathbf{E}}$ depend on frames, $\lambda_{k}$ and $\xi_{k}$ in (4) and (6) are not, which justifies the absence of $\mathbf{E}$ in their subscripts and the use of same notation for eigenvalues and iRFPC scores in (2), (4), (5) and (6). This follows from the formulas for change of frames that we shall develop below.

Suppose $\mathbf{A}=\left(A_{1}, \ldots, A_{d}\right)$ is another orthonormal frame. Change from $\mathbf{E}(p)=\left\{E_{1}(p), \ldots, E_{d}(p)\right\}$ to $\mathbf{A}(p)=\left\{A_{1}(p), \ldots, A_{d}(p)\right\}$ can be characterized by a unitary matrix $\mathbf{O}_{p}$. For example, $\mathbf{A}(t)=$ $\mathbf{O}_{\mu(t)}^{T} \mathbf{E}(t)$ and hence $Z_{\mathbf{A}}(t)=\mathbf{O}_{\mu(t)} Z_{\mathbf{E}}(t)$ for all $t$. Then the covariance function of $Z_{\mathbf{A}}$ is given by

$$
\begin{equation*}
\mathcal{C}_{\mathbf{A}}(s, t)=\mathbb{E}\left\{Z_{\mathbf{A}}(s) Z_{\mathbf{A}}(t)^{T}\right\}=\mathbb{E}\left\{\mathbf{O}_{\mu(s)} Z_{\mathbf{E}}(s) Z_{\mathbf{E}}(t)^{T} \mathbf{O}_{\mu(t)}^{T}\right\}=\mathbf{O}_{\mu(s)} \mathcal{C}_{\mathbf{E}}(s, t) \mathbf{O}_{\mu(t)}^{T}, \tag{7}
\end{equation*}
$$

and consequently,

$$
\mathcal{C}_{\mathbf{A}}(s, t)=\sum_{k=1}^{\infty} \lambda_{k}\left\{\mathbf{O}_{\mu(s)} \phi_{\mathbf{E}, k}(s)\right\}\left\{\mathbf{O}_{\mu(t)} \phi_{\mathbf{E}, k}(t)\right\}^{T} .
$$

From the above calculation, we immediately see that $\lambda_{k}$ are also eigenvalues of $\mathcal{C}_{\mathbf{A}}$. Moreover, the eigenfunction associated with $\lambda_{k}$ for $\mathcal{C}_{\mathbf{A}}$ is given by

$$
\begin{equation*}
\phi_{\mathbf{A}, k}(t)=\mathbf{O}_{\mu(t)} \phi_{\mathbf{E}, k}(t) . \tag{8}
\end{equation*}
$$

Also, the variable $\xi_{k}$ in (4) and (6) is the functional principal component score for $Z_{\mathbf{A}}$ associated with $\phi_{\mathbf{A}, k}$, as seen by $\int_{\mathcal{T}} Z_{\mathbf{A}}^{T}(t) \phi_{\mathbf{A}, k}(t) \mathrm{d} v(t)=\int_{\mathcal{T}} Z_{\mathbf{E}}^{T}(t) \mathbf{O}_{\mu(t)}^{T} \mathbf{O}_{\mu(t)} \phi_{\mathbf{E}, k}(t) \mathrm{d} v(t)=\int_{\mathcal{T}} Z_{\mathbf{E}}^{T}(t) \phi_{\mathbf{E}, k}(t) \mathrm{d} v(t)$. The following proposition summarizes the above results.

Proposition 5 (Invariance Principle). Let $X$ be a $\mathcal{M}$-valued random process satisfying conditions A. 1 and A.2. Suppose $\mathbf{E}$ and $\mathbf{A}$ are measurable orthonormal frames that are continuous on a neighborhood of the image of $\mu$, and $Z_{\mathbf{E}}$ denotes the $\mathbf{E}$-coordinate log-process of $X$. Assume $\mathbf{O}_{p}$ is the unitary matrix continuously varying with $p$ such that $\mathbf{A}(p)=\mathbf{O}_{p}^{T} \mathbf{E}(p)$ for $p \in \mathcal{M}$.

1. The $\mathcal{L}^{r}$-norm of $Z_{\mathbf{E}}$ for $r>0$, defined by $\left\|Z_{\mathbf{E}}\right\|_{\mathcal{L}^{r}}=\left\{\mathbb{E} \int_{\mathcal{T}}\left|Z_{\mathbf{E}}(t)\right|^{r} \mathrm{~d} v(t)\right\}^{1 / r}$, is independent of the choice of frames. In particular, $\left\|Z_{\mathbf{E}}\right\|_{\mathcal{L}^{2}}^{2}=\mathbb{E}\left\|\log _{\mu} X\right\|_{\mu}^{2}$ for all orthonormal frames $\mathbf{E}$.
2. If $\mathbb{E}\left\|\log _{\mu} X\right\|_{\mu}^{2}<\infty$, then the covariance function of $Z_{\mathbf{E}}$ exists for all $\mathbf{E}$ and admits decomposition of (5). Also, (2) and (5) are related by $\phi_{k}(t)=\phi_{\mathbf{E}, k}^{T}(t) \mathbf{E}(\mu(t))$ for all $t$, and the eigenvalues $\lambda_{k}$ coincide. Furthermore, the eigenvalues of $\mathcal{C}_{\mathbf{E}}$ and the principal component scores of Karhunen-Loève expansion of $Z_{\mathbf{E}}$ do not depend on $\mathbf{E}$.
3. The covariance functions $\mathcal{C}_{\mathbf{A}}$ and $\mathcal{C}_{\mathbf{E}}$ of respectively $Z_{\mathbf{A}}$ and $Z_{\mathbf{E}}$, if exist, are related by (7). Furthermore, their eigendecomposions are related by (8) and $Z_{\mathbf{A}}(t)=\mathbf{O}_{\mu(t)} Z_{\mathbf{E}}(t)$ for all $t \in \mathcal{T}$.
4. If $\mathbb{E}\left\|\log _{\mu} X\right\|_{\mu}^{2}<\infty$ and sample paths of $X$ are continuous, then the scores $\xi_{k}$ (6) coincides with the iRFPC scores in (4).

We conclude this subsection by emphasizing that the concept of covariance function of the logprocess depends on the frame $\mathbf{E}$, while the covariance operator, eigenvalues, eigenelements and iRFPC scores do not. In particular, the scores $\xi_{k}$, which are often the input for further statistical analysis such as regression and classification, are invariant to the choice of coordinate frames. An important consequence of the invariance principle is that, these scores can be safely computed in any convenient coordinate frame without altering the subsequent analysis.
2.4. Connection to the Special Case of Euclidean Submanifolds. Our framework applies to general manifolds that include Euclidean submanifolds as special examples to which the methodology of Dai and Müller (2017) also applies. When the underlying manifold is a $d$-dimensional submanifold of the Euclidean space $\mathbb{R}^{d_{0}}$ with $d<d_{0}$, we recall that the tangent space at each point is identified as a $d$-dimensional linear subspace of $\mathbb{R}^{d_{0}}$. For such Euclidean manifolds, Dai and Müller (2017) treat the log-process of $X$ as a $\mathbb{R}^{d_{0}}$-valued random process, and derive the representation for the log-process (Eq. (5) in their paper) within the ambient Euclidean space. This is distinctly different from our intrinsic representation (3) based on the theory of tensor Hilbert spaces, despite their similar appearance. For instance, Eq. (5) in their work can only be defined for Euclidean submanifolds, while ours is applicable to general Riemannian manifolds. Similarly, the covariance function defined in Dai and Müller (2017), denoted by $C^{D M}(s, t)$, is associated with the ambient log-process $V(t) \in \mathbb{R}^{d_{0}}$, i.e., $C^{D M}(s, t)=\mathbb{E} V(s)^{T} V(t)$. Such an ambient covariance function can only be defined for Euclidean submanifolds but not general manifolds.

Nevertheless, there are connections between the ambient method of Dai and Müller (2017) and our framework when $\mathcal{M}$ is a Euclidean submanifold. For instance, the mean curve is intrinsically defined in the same way in both works. For the covariance structure, although our covariance function $\mathcal{C}_{\mathbf{E}}$ is a $d \times d$ matrix-valued function while $C^{D M}(s, t)$ is a $d_{0} \times d_{0}$ matrix-valued function, they both represent the intrinsic covariance operator when $\mathcal{M}$ is a Euclidean submanifold. To see so, first, we observe that the ambient log-process $V(t)$ as defined in Dai and Müller (2017) at the time $t$, although is ambiently $d_{0}$-dimensional, lives in a $d$-dimensional linear subspace of $\mathbb{R}^{d_{0}}$. Second, the orthonormal basis $\mathbf{E}(t)$ for the tangent space at $\mu(t)$ can be realized by a $d_{0} \times d$ full-rank matrix $\mathbf{G}_{t}$ by concatenating vectors $E_{1}(\mu(t)), \ldots, E_{d}(\mu(t))$. Then $U(t)=\mathbf{G}_{t}^{T} V(t)$ is the $\mathbf{E}$-coordinate process of $X$. This implies that $\mathcal{C}_{\mathbf{E}}(s, t)=\mathbf{G}_{s}^{T} C^{D M}(s, t) \mathbf{G}_{t}$. On the other hand, since $V(t)=\mathbf{G}_{t} U(t)$, one has $C^{D M}(s, t)=\mathbf{G}_{t} \mathcal{C}_{\mathbf{E}}(s, t) \mathbf{G}_{t}^{T}$. Thus, $\mathcal{C}_{\mathbf{E}}$ and $C^{D M}$ determine each other and represent the same object. In light of this observation and the invariance principle stated in Proposition 5 , when $\mathcal{M}$ is a Euclidean submanifold, $C^{D M}$ can be viewed as the ambient representation of the intrinsic covariance operator $\mathcal{C}$, while $\mathcal{C}_{\mathbf{E}}$ is the coordinate representation of $\mathcal{C}$ with respect to the frame $\mathbf{E}$. Similarly, the eigenfunctions $\phi_{k}^{D M}$ of $C^{D M}$ are the ambient representation of the eigenelements $\phi_{k}$ of $\mathcal{C}$. The above reasoning also applies to sample mean functions and sample covariance structure. Specifically, when $\mathcal{M}$ is a Euclidean submanifold, our estimator for the mean function discussed in Section 3 is identical to the one in Dai and Müller (2017), while the estimators for the covariance function and eigenfunctions proposed in Dai and Müller (2017) are the ambient representation of our estimators stated in Section 3.

However, when quantifying the discrepancy between the population covariance structure and its estimator, Dai and Müller (2017) adopt the Euclidean difference as a measure. For instance, they use $\hat{\phi}_{k}^{D M}-\phi_{k}^{D M}$ to represent the discrepancy between the sample eigenfunctions and the population eigenfunctions, where $\hat{\phi}_{k}^{D M}$ is the sample version of $\phi_{k}^{D M}$. When $\hat{\mu}(t)$, the sample version of $\mu(t)$, is not equal to $\mu(t), \hat{\phi}_{k}^{D M}(t)$ and $\phi_{k}^{D M}(t)$ belong to different tangent spaces. In such case, the Euclidean difference $\hat{\phi}_{k}^{D M}-\phi_{k}^{D M}$ is a Euclidean vector that does not belong to the tangent space
at either $\hat{\mu}(t)$ or $\mu(t)$, as illustrated in the left panel of Figure 1. In other words, the Euclidean difference of ambient eigenfunctions does not obey the geometry of the manifold, hence might not properly measure the intrinsic discrepancy. In particular, the measure $\left\|\hat{\phi}_{k}^{D M}-\phi_{k}^{D M}\right\|_{\mathbb{R}^{d_{0}}}$ might completely result from the departure of the ambient Euclidean geometry from the manifold, rather than the intrinsic discrepancy between the sample and population eigenfunctions, as demonstrated in the left panel of Figure 1. Similar reasoning applies to $\hat{C}^{D M}-C^{D M}$. In contrast, we base on Proposition 2 to propose an intrinsic measure to characterize the intrinsic discrepancy between a population quantity and its estimator in Section 3.2.

## 3. Intrinsic Riemannian Functional Principal Component Analysis.

3.1. Model and Estimation. Suppose $X$ admits the intrinsic Riemannian Karhunen-Loève expansion (4), and $X_{1}, \ldots, X_{n}$ are a random sample of $X$. In the sequel, we assume that trajectories $X_{i}$ are fully observed. In the case that data are densely observed, each trajectory can be individually interpolated by using regression techniques for manifold valued data, such as Steinke, Hein and Schölkopf (2010), Cornea et al. (2017) and Petersen and Müller (2017). This way the densely observed data could be represented by their interpolated surrogates, and thus treated as if they were fully observed curves. When data are sparse, delicate information pooling of observations across different subjects is required. The development of such methods is substantial and beyond the scope of this paper.

In order to estimate the mean function $\mu$, we define the finite-sample version of $F$ in (1) by

$$
F_{n}(p, t)=\frac{1}{n} \sum_{i=1}^{n} d_{\mathcal{M}}^{2}\left(X_{i}(t), p\right) .
$$

Then, an estimator for $\mu$ is given by

$$
\hat{\mu}(t)=\underset{p \in \mathcal{M}}{\arg \min } F_{n}(p, t) .
$$

The computation of $\hat{\mu}$ depends on the Riemannian structure of the manifold. Readers are referred to Cheng et al. (2016) and Salehian et al. (2015) for practical algorithms. For a subset $A$ of $\mathcal{M}$, $A^{\epsilon}$ denotes the set $\cup_{p \in A} B(p ; \epsilon)$, where $B(p ; \epsilon)$ is the ball with center $p$ and radius $\epsilon$ in $\mathcal{M}$. We use $\operatorname{Im}^{-\epsilon}\left(\operatorname{Exp}_{\mu(t)}\right)$ to denote the set $\mathcal{M} \backslash\left\{\mathcal{M} \backslash \operatorname{Im}\left(\operatorname{Exp}_{\mu(t)}\right)\right\}^{\epsilon}$. In order to define $\log _{\hat{\mu}} X_{i}$, at least with a dominant probability for a large sample, we shall assume a slightly stronger condition than A.2:
A. $2^{\prime} \quad$ There is some constant $\epsilon_{0}>0$ such that $\operatorname{Pr}\left\{\forall t \in \mathcal{T}: X(t) \in \operatorname{Im}^{-\epsilon_{0}}\left(\operatorname{Exp}_{\mu(t)}\right)\right\}=1$.

Then, combining the fact $\sup _{t}|\hat{\mu}(t)-\mu(t)|=o_{\text {a.s. }}$ (1) that we will show later, we conclude that for a large sample, almost surely, $\operatorname{Im}^{-\epsilon}\left(\operatorname{Exp}_{\mu(t)}\right) \subset \operatorname{Im}\left(\operatorname{Exp}_{\hat{\mu}(t)}\right)$ for all $t \in \mathcal{T}$. Therefore, under this condition, $\log _{\hat{\mu}(t)} X_{i}(t)$ is well define almost surely for a large sample.

The intrinsic Riemannian covariance operator is estimated by its finite-sample version

$$
\hat{\mathcal{C}}=\frac{1}{n} \sum_{i=1}^{n}\left(\log _{\hat{\mu}} X_{i}\right) \otimes\left(\log _{\hat{\mu}} X_{i}\right) .
$$

This sample intrinsic Riemannian covariance operator also admits an intrinsic eigendecomposion $\hat{\mathcal{C}}=\sum_{k=1}^{\infty} \hat{\lambda}_{k} \hat{\phi}_{k}$ for $\hat{\lambda}_{1} \geq \hat{\lambda}_{2} \geq \cdots \geq 0$. Therefore, the estimates for the eigenvalues $\lambda_{k}$ are given by $\hat{\lambda}_{k}$, while the estimates for $\phi_{k}$ are given by $\hat{\phi}_{k}$. These estimates can also be conveniently obtained
under a frame, due to the invariance principle stated in Proposition 5. Let $\mathbf{E}$ be a chosen orthonormal frame and $\hat{\mathcal{C}}_{\mathbf{E}}$ be the sample covariance function based on $\hat{Z}_{\mathbf{E}, 1}, \ldots, \hat{Z}_{\mathbf{E}, n}$, where $\hat{Z}_{\mathbf{E}, i}$ is the coordinate process of $\log _{\hat{\mu}(t)} X_{i}(t)$ under the frame $\mathbf{E}$ with respect to $\hat{\mu}$. We can then obtain the eigendecomposition $\hat{\mathcal{C}}_{\mathbf{E}}(s, t)=\sum_{k=1}^{\infty} \hat{\lambda}_{k} \hat{\phi}_{\mathbf{E}, k}(s) \hat{\phi}_{\mathbf{E}, k}(t)^{T}$, which yields $\hat{\phi}_{k}(t)=\hat{\phi}_{\mathbf{E}, k}^{T}(t) \mathbf{E}(t)$ for $t \in \mathcal{T}$. Finally, the truncated process for $X_{i}$ is estimated by

$$
\begin{equation*}
\hat{X}_{i}^{(K)}=\operatorname{Exp}_{\hat{\mu}} \sum_{k=1}^{K} \hat{\xi}_{i k} \hat{\phi}_{k}, \tag{9}
\end{equation*}
$$

where $\hat{\xi}_{i k}=\left\langle\log _{\hat{\mu}} X_{i}, \hat{\phi}_{k}\right\rangle_{\hat{\mu}}$ are estimated iRFPC scores. The above truncated iRKL expansion can be regarded as generalization of the representation (10) in Dai and Müller (2017) from Euclidean submanifolds to general Riemannian manifolds.
3.2. Asymptotic Properties. To quantify the difference between $\hat{\mu}$ and $\mu$, it is natural to use the square geodesic distance $d_{\mathcal{M}}(\hat{\mu}(t), \mu(t))$ as a measure of discrepancy. For the asymptotic properties of $\hat{\mu}$, we need following regularity conditions.
B. 1 The manifold $\mathcal{M}$ is connected and complete. In addition, the exponential map $\operatorname{Exp}_{p}$ : $T_{p} \mathcal{M} \rightarrow \mathcal{M}$ is surjective at every point $p \in \mathcal{M}$.
B. 2 The sample paths of $X$ are continuous.
B. $3 \quad F$ is finite. Also, for all compact subsets $\mathcal{K} \subset \mathcal{M}, \sup _{t \epsilon \mathcal{T}} \sup _{p \in \mathcal{K}} \mathbb{E} d_{\mathcal{M}}^{2}(p, X(t))<\infty$.
B. 4 The image $\mathcal{U}$ of the mean function $\mu$ is bounded, i.e the diameter is finite, $\operatorname{diam}(\mathcal{U})<\infty$.
B. $5 \quad$ For all $\epsilon>0, \inf _{t \in \mathcal{T}} \inf _{p: d_{\mathcal{M}}(p, \mu(t)) \geq \epsilon} F(p, t)-F(\mu(t), t)>0$.

To state the next condition, let $V_{t}(p)=\log _{p} X(t)$. The calculus of manifolds suggests that $V_{t}(p)=$ $-d_{\mathcal{M}}(p, X(t)) \operatorname{grad}_{p} d_{\mathcal{M}}(p, X(t))=\operatorname{grad}_{p}\left(-d_{\mathcal{M}}^{2}(p, X(t)) / 2\right)$, where $\operatorname{grad}_{p}$ denotes the gradient operator at $p$. For each $t \in \mathcal{T}$, let $H_{t}$ denote the Hessian of the real function $d_{\mathcal{M}}^{2}(\cdot, X(t)) / 2$, i.e., for vector fields $U$ and $W$ on $\mathcal{M}$,

$$
\left\langle H_{t} U, W\right\rangle(p)=\left\langle-\nabla_{U} V_{t}, W\right\rangle(p)=\operatorname{Hess}_{p}\left(\frac{1}{2} d_{\mathcal{M}}^{2}(p, X(t))\right)(U, W) .
$$

B. $6 \inf _{t \in \mathcal{T}}\left\{\lambda_{\text {min }}\left(\mathbb{E} H_{t}\right)\right\}>0$, where $\lambda_{\text {min }}(\cdot)$ denotes the smallest eigenvalue of an operator or matrix.
B. $7 \mathbb{E} L(X)^{2}<\infty$ and $L(\mu)<\infty$, where $L(f):=\sup _{s \neq t} d_{\mathcal{M}}(f(s), f(t)) /|s-t|$ for a real function $f$ on $\mathcal{M}$.

The assumption B. 1 regarding the property of manifolds is met in general, e.g. the $d$-dimensional unit sphere $\mathbb{S}^{d}$, SPD manifolds, etc. By Hopf-Rinow theorem, the condition also implies that $\mathcal{M}$ is geodesically complete. Conditions similar to B.2, B.5, B. 6 and B. 7 are made in Dai and Müller (2017). The condition B. 4 is a weak requirement for the mean function and is automatically satisfied if the manifold is compact, while B. 3 is analogous to standard moment conditions in the literature of Euclidean functional data analysis and becomes superfluous when $\mathcal{M}$ is compact. If $\mathcal{M}=\mathbb{R}$, then B. 3 is equivalent to $\sup _{t \epsilon \mathcal{T}} \mathbb{E}|X(t)|^{2}<\infty$, a condition commonly made in the literature of functional data analysis, e.g., Hall and Hosseini-Nasab (2006). The following theorem summarizes asymptotic properties of $\hat{\mu}$ for general Riemannian manifolds. Its proof can be obtained by mimicking the techniques in Dai and Müller (2017), with additional care of the case that $\mathcal{M}$ is noncompact. For completeness, we provide its proof in Appendix. As noted by Dai and Müller (2017), the root- $n$ rate can not be improved in general.

ThEOREM 6. Under conditions A.1, A.2' and B.1-B.4, the following holds.

1. $\mu$ is uniformly continuous, and $\hat{\mu}$ is uniformly continuous with probability one.
2. $\hat{\mu}$ is a uniformly strong consistent estimator for $\mu$, i.e., $\sup _{t} d_{\mathcal{M}}(\hat{\mu}(t), \mu(t))=o_{\text {a.s. }}(1)$.
3. If additional conditions B.5-B.7 are assumed, then $\sqrt{n} \log _{\mu} \hat{\mu}$ converges in distribution to $a$ Gaussian measure on the tensor Hilbert space $\mathscr{T}(\mu)$.
4. If additional conditions B.5-B.7 are assumed, then $\sup _{t \in \mathcal{T}} d_{\mathcal{M}}^{2}(\hat{\mu}(t), \mu(t))=O_{P}\left(n^{-1}\right)$ and $\int_{\mathcal{T}} d_{\mathcal{M}}^{2}(\hat{\mu}(t), \mu(t)) \mathrm{d} v(t)=O_{P}\left(n^{-1}\right)$.

For asymptotic analysis of the estimated eigenstructure, as discussed in Section 2.1 and $2.4, \hat{\mathcal{C}}-\mathcal{C}$ and $\hat{\phi}_{k}-\phi_{k}$ are not defined for intrinsic Riemannian functional data analysis, since they are objects originated from different tensor Hilbert spaces $\mathscr{T}(\mu)$ and $\mathscr{T}(\hat{\mu})$ induced by different curves $\mu$ and $\hat{\mu}$. Therefore, we shall adopt the intrinsic measure of discrepancy developed in Section 2.1, namely, $\hat{\mathcal{C}} \ominus_{\Phi} \mathcal{C}$ and $\hat{\phi}_{k} \ominus_{\Gamma} \phi_{k}$, where the dependence of $\Phi$ and $\Gamma$ on $\hat{\mu}$ and $\mu$ is suppressed for simplicity. As mentioned at the end of Section 2.1 , both $\Gamma$ and $\Phi$ depend on the choice of the family of curves $\gamma$. Here, a canonical choice for each $\gamma(t, \cdot)$ is the minimizing geodesic between $\mu(t)$ and $\hat{\mu}(t)$. The existence and uniqueness of such geodesics can be deduced from assumptions A. 1 and A.2'. Also, the continuity of $\mu(t)$ and $\hat{\mu}(t)$ implies the continuity of $\gamma(\cdot, \cdot)$ and hence the measurability of $\gamma(\cdot, s)$ for each $s \in[0,1]$. By Proposition 2 , one sees that $\Phi \hat{\mathcal{C}}=n^{-1} \sum_{i=1}^{n}\left(\Gamma \hat{V}_{i} \otimes \Gamma \hat{V}_{i}\right)$, recalling that $\hat{V}_{i}=\log _{\hat{\mu}} X_{i}$ is a vector field along $\hat{\mu}$. It can also be seen that $\left(\hat{\lambda}_{k}, \Gamma \hat{\phi}_{k}\right)$ are eigenpairs of $\Phi \hat{\mathcal{C}}$. These identities match our intuition that the transported sample covariance operator ought to be an operator derived from transported sample vector fields, and that the eigenfunctions of the transported operator are identical to the transported eigenfunctions.

To state the asymptotic properties for the eigenstructure, we define

$$
\begin{array}{ll}
\eta_{k}=\min _{1 \leq j \leq k}\left(\lambda_{j}-\lambda_{j+1}\right), & J=\inf \left\{j \geq 1: \lambda_{j}-\lambda_{j+1} \leq 2\left\|\hat{\mathcal{C}} \ominus_{\Phi} \mathcal{C}\right\|_{\mu}\right\} \\
\hat{\eta}_{j}=\min _{1 \leq j \leq k}\left(\hat{\lambda}_{j}-\hat{\lambda}_{j+1}\right), & \hat{J}=\inf \left\{j \geq 1: \hat{\lambda}_{j}-\hat{\lambda}_{j+1} \leq 2\left\|\hat{\mathcal{C}} \ominus_{\Phi} \mathcal{C}\right\|_{\mu}\right\}
\end{array}
$$

THEOREM 7. Assume that every eigenvalue $\lambda_{k}$ has multiplicity one, and conditions A.1, A.2' and B.1-B. 7 hold. Suppose tangent vectors are parallel transported along minimizing geodesics for defining the parallel transporters $\Gamma$ and $\Phi$. If $\mathbb{E}\left\|\log _{\mu} X\right\|_{\mu}^{4}<\infty$, then $\left\|\hat{\mathcal{C}} \ominus_{\Phi} \mathcal{C}\right\|_{\mu}^{2}=O_{P}\left(n^{-1}\right)$. Furthermore, $\sup _{k \geq 1}\left|\hat{\lambda}_{k}-\lambda_{k}\right| \leq\left\|\hat{\mathcal{C}} \ominus_{\Phi} \mathcal{C}\right\|_{\mu}$ and for all $1 \leq k \leq J-1$,

$$
\begin{equation*}
\left\|\hat{\phi}_{k} \ominus_{\Gamma} \phi_{k}\right\|_{\mu}^{2} \leq 8\| \| \hat{\mathcal{C}} \ominus_{\Phi} \mathcal{C} \|_{\mu}^{2} / \eta_{k}^{2} \tag{10}
\end{equation*}
$$

If $\left(J, \eta_{j}\right)$ is replaced by $\left(\hat{J}, \hat{\eta}_{j}\right)$, then (10) holds with probability 1.
In this theorem, (10) generalizes Lemma 4.3 of Bosq (2000) to the Riemannian setting. Note that he intrinsic rate for $\hat{\mathcal{C}}$ is optimal. Also, from (10) one can deduce the optimal rate $\left\|\hat{\phi}_{k} \ominus_{\Gamma} \phi_{k}\right\|_{\mu}^{2}=$ $O_{P}\left(n^{-1}\right)$ for a fixed $k$. We stress that, these results apply to not only Euclidean submanifolds, but also general Riemannian manifolds.

## 4. Intrinsic Riemannian Funcitonal Linear Regression.

4.1. Regression Model and Estimation. Classical functional linear regression for Euclidean functional data is well studied in the literature, i.e., the model relating a scalar response $Y$ and a functional predictor $X$ by $Y=\alpha+\int_{\mathcal{T}} X(t) \beta(t) \mathrm{d} v(t)+\varepsilon$, where $\alpha$ is the intercept, $\beta$ is the slope function and $\varepsilon$ represents measurement errors, e.g., Cardot, Ferraty and Sarda (2003), Cardot, Mas and Sarda (2007), Hall and Horowitz (2007) and Yuan and Cai (2010), among others. However, for Riemannian functional data, both $X(t)$ and $\beta(t)$ take values in a manifold and hence the product $X(t) \beta(t)$ is not well defined. Rewriting the model as $Y=\alpha+\left\langle\langle X, \beta\rangle_{\mathcal{L}^{2}}+\varepsilon\right.$, where $\langle\cdot \cdot \cdot \cdot\rangle_{\mathcal{L}^{2}}$ is the canonical inner product of the $\mathcal{L}^{2}$ square integrable functions, we propose to replace $\left\langle\langle\cdot, \cdot\rangle_{\mathcal{L}^{2}}\right.$ by the inner product on the tensor Hilbert space $\mathscr{T}(\mu)$, and define the following Riemannian functional linear regression model:

$$
\begin{equation*}
Y=\alpha+\left\langle\left\langle\log _{\mu} X, \log _{\mu} \beta\right\rangle_{\mu}+\varepsilon\right. \tag{11}
\end{equation*}
$$

where we require conditions A. 1 and A.2. Note that $\beta$ is a manifold valued function defined on $\mathcal{T}$, namely the Riemannian slope function of the model (11), and this model is linear in terms of $\log _{\mu(t)} \beta(t)$. We stress that the model (11) is intrinsic to the Riemannian structures of the manifold.

According to Theorem 2.1 of Bhattacharya and Patrangenaru (2003), the process $\log _{\mu(t)} X(t)$ is centered at its mean function, i.e., $\mathbb{E} \log _{\mu(t)} X(t) \equiv 0$, which also implies that $\alpha=\mathbb{E} Y$. Since the focus is the Riemannian slope function $\beta$, in the sequel, without loss of generality, we assume that $Y$ is centered and hence $\alpha=0$. In practice, we may use sample mean of $Y$ for centering. With the same reasoning as Cardot, Ferraty and Sarda (2003), one can see that $\log _{\mu} \beta$ and hence $\beta$ are identifiable if A.1, A. 2 and the following condition hold.
C. 1 The random pair $(X, Y)$ satisfies $\sum_{k=1}^{\infty} \lambda_{k}^{-2}\left[\mathbb{E}\left\{Y\left\langle\left\langle\log _{\mu} X, \phi_{k}\right\rangle{ }_{\mu}\right\}\right]^{2}<\infty\right.$, where $\left(\lambda_{k}, \phi_{k}\right)$ are eigenpairs of the covariance operator $\mathcal{C}$ of $\log _{\mu} X$.
To estimate the Riemannian slope function $\beta$, by analogy to the FPCA approach in traditional functional linear regression (Hall and Horowitz, 2007), we represent $\beta$ by $\beta=\operatorname{Exp}_{\mu} \sum_{k} b_{k} \phi_{k}$. Then $b_{k}=\lambda_{k}^{-1} a_{k}$ with $a_{k}=\left\langle\chi \chi, \phi_{k}\right\rangle_{\mu}$ and $\chi(t)=\mathbb{E}\left\{Y \log _{\mu(t)} X(t)\right\}$. The empirical iRFPCA estimate of $\beta$ is then given by

$$
\begin{equation*}
\hat{\beta}=\operatorname{Exp}_{\hat{\mu}} \sum_{k=1}^{K} \hat{b}_{k} \hat{\phi}_{k}, \tag{12}
\end{equation*}
$$

where $\hat{b}_{k}=\hat{\lambda}_{k}^{-1} \hat{a}_{k}, \hat{a}_{k}=\left\langle\left\langle\hat{\chi}, \hat{\phi}_{k}\right\rangle_{\hat{\mu}}, \hat{\chi}(t)=n^{-1} \sum_{i=1}^{n}\left(Y_{i}-\bar{Y}\right) \log _{\hat{\mu}(t)} X_{i}(t)\right.$ and $\bar{Y}=n^{-1} \sum_{i=1}^{n} Y_{i}$. We can also obtain a Tikhonov estimator $\tilde{\beta}$ as follows. For $\rho>0$, define $\hat{\mathcal{C}}^{+}=\left(\hat{\mathcal{C}}+\rho I_{\hat{\mu}}\right)^{-1}$, where $I_{\hat{\mu}}$ is the identity operator on $\mathscr{T}(\hat{\mu})$. The Tikhonov estimator is given by

$$
\begin{equation*}
\tilde{\beta}=\operatorname{Exp}_{\hat{\mu}}\left(\hat{\mathcal{C}}^{+} \hat{\chi}\right) \tag{13}
\end{equation*}
$$

In practice, it is convenient to conduct computation with respect to an orthonormal frame $\mathbf{E}$. For each $X_{i}$, the $\mathbf{E}$-coordinate along the curve $\hat{\mu}$ is denoted by $\hat{Z}_{\mathbf{E}, i}$. Note that by Theorem 2.1 of Bhattacharya and Patrangenaru (2003), $n^{-1} \sum_{i=1}^{n} \log _{\hat{\mu}} X_{i}=0$ implies that $n^{-1} \sum_{i=1}^{n} \hat{Z}_{\mathbf{E}, i}(t) \equiv 0$. Thus, the empirical covariance function $\hat{\mathcal{C}}_{\mathbf{E}}$ is computed by $\hat{\mathcal{C}}_{\mathbf{E}}(s, t)=n^{-1} \sum_{i=1}^{n} \hat{Z}_{\mathbf{E}, i}(s) \hat{Z}_{\mathbf{E}, i}^{T}(t)$. Then, $\hat{\chi}_{\mathbf{E}}=n^{-1} \sum_{i=1}^{n}\left(Y_{i}-\bar{Y}\right) \hat{Z}_{\mathbf{E}, i}$ and $\hat{a}_{k}=\int_{\mathcal{T}} \hat{\chi}_{\mathbf{E}, i}^{T}(t) \hat{\phi}_{\mathbf{E}, k}(t) \mathrm{d} v(t)$, recalling that $\hat{\phi}_{\mathbf{E}, k}$ are eigenfunctions of $\hat{\mathcal{C}}_{\mathbf{E}}$. Finally, the $\mathbf{E}$-coordinate of $\log _{\hat{\mu}} \hat{\beta}$ is given by $\sum_{k=1}^{K} \hat{b}_{k} \hat{\phi}_{\mathbf{E}, k}$. For the Tikhonov estimator, we first observe that $\hat{\mathcal{C}}^{+}$shares the same eigenfunctions with $\hat{\mathcal{C}}$, while the eigenvalues $\hat{\lambda}_{k}^{+}$of $\hat{\mathcal{C}}^{+}$is related to $\hat{\lambda}_{k}$ by $\hat{\lambda}_{k}^{+}=1 /\left(\hat{\lambda}_{k}+\rho\right)$. Therefore, $\hat{\mathcal{C}}^{+} \hat{\chi}=\sum_{k=1}^{\infty} \hat{\lambda}_{k}^{+}\left\langle\left\langle\hat{\phi}_{k}, \hat{\chi}\right\rangle_{\hat{\mu}} \hat{\phi}_{k}\right.$ and its $\mathbf{E}$-coordinate is given
by $\sum_{k=1}^{\infty} \hat{\lambda}_{k}^{+} \hat{a}_{k} \hat{\phi}_{\mathbf{E}, k}$. We emphasize that both estimators $\hat{\beta}$ and $\tilde{\beta}$ are intrinsic and hence independent of frames, while their coordinate representations depend on the choice of frames. In addition, as a special case, when $\mathcal{M}$ is a Euclidean submanifold, an argument similar to that in Section 2.4 can show that, if one treats $X$ as an ambient random process and adopts the FPCA and Tikhonov regularization approaches (Hall and Horowitz, 2007) to estimate the slope function $\beta$ in the ambient space, then the estimates are the ambient representation of our estimates $\log _{\hat{\mu}} \hat{\beta}$ and $\log _{\hat{\mu}} \tilde{\beta}$ in (12) and (13), respectively.
4.2. Asymptotic Properties. In order to derive convergence of the iRFPCA estimator and the Tikhonov estimator, we shall assume the sectional curvature of the manifold is bounded from below by $\kappa$ to exclude pathological cases. The compact support condition on $X$ in the case $\kappa<0$ might be relaxed to weaker assumptions on the tail decay of the distribution of $\log _{\mu} X$. Such weaker conditions do not provide more insight for our derivation, but complicate the proofs significantly, which is not pursued further.
C. 2 If $\kappa<0, X$ is assumed to lie in a compact subset $\mathcal{K}$ almost surely. Moreover, errors $\varepsilon_{i}$ are identically distributed with zero mean and variance not exceeding a constant $C>0$.
The follow conditions concern the spacing and the decay rate of eigenvalues $\lambda_{k}$ of the covariance operator, as well as the strength of the signal $b_{k}$. They are standard in the literature of functional linear regression, e.g. Hall and Horowitz (2007).
C. 3 For $k \geq 1, \lambda_{k}-\lambda_{k+1} \geq C k^{-\alpha-1}$.
C. $4 \quad\left|b_{k}\right| \leq C k^{-\varrho}, \alpha>1$, and $(\alpha+1) / 2<\varrho$.

Let $\mathcal{F}(C, \alpha, \varrho)$ be the collection of distributions $f$ of ( $X, Y$ ) satisfying conditions C.2-C.4. The following theorem establish the convergence rate of the iRFPCA estimator $\hat{\beta}$ for the class of models in $\mathcal{F}(C, \alpha, \varrho)$.

Theorem 8. Assume that conditions A.1, A.2', B.1-B. 7 and C.1-C.4 hold. If $K \asymp n^{1 /(2 \alpha+2 \varrho)}$, then

$$
\lim _{c \rightarrow \infty} \limsup _{n \rightarrow \infty} \sup _{f \in \mathcal{F}} \operatorname{Pr}_{f}\left\{\int_{\mathcal{T}} d_{\mathcal{M}}^{2}(\hat{\beta}(t), \beta(t)) \mathrm{d} v(t)>c n^{-\frac{2 \varrho-1}{2 \rho+2 \alpha}}\right\}=0 .
$$

For the Tikhonov estimator $\tilde{\beta}$, we have a similar result. Instead of conditions C.3-C.4, we make the following assumptions, which again are standard in the functional data literature.
C. $5 \quad k^{-\alpha} \leq C \lambda_{k}$.
C. $6\left|b_{k}\right| \leq C k^{-\varrho}, \alpha>1$, and $\alpha-1 / 2<\varrho$.

Let $\mathcal{G}(C, \alpha, \varrho)$ be the class of distributions of $(X, Y)$ that satisfy C. 2 and C.5-C.6. The following result shows that, the convergence rate of $\tilde{\beta}$ is at least $n^{-(2 \varrho-\alpha) /(\alpha+2 \varrho)}$ in terms of integrated square geodesic distance.

Theorem 9. Assume that conditions A.1, A.2', B.1-B.7, C.1-C.2, and C.5-C.6 hold. If $\rho \leftrightharpoons$ $n^{-\alpha /(\alpha+2 \varrho)}$, then

$$
\lim _{c \rightarrow \infty} \limsup _{n \rightarrow \infty} \sup _{f \in \mathcal{G}} \operatorname{Pr}_{f}\left\{\int_{\mathcal{T}} d_{\mathcal{M}}^{2}(\tilde{\beta}(t), \beta(t)) \mathrm{d} v(t)>c n^{-\frac{20-\alpha}{2 \alpha+\alpha}}\right\}=0 .
$$

It is important to point out that the theory in Hall and Horowitz (2007) is formulated for Euclidean functional data and hence do not apply to Riemannian functional data. In particular, their proof machinery depends on the linear structure of the sample mean function $n^{-1} \sum_{i=1}^{n} X_{i}$ for Euclidean functional data. However, the intrinsic empirical mean generally does not admit an analytic expression, which hinges derivation of the optimal convergence rate. We leave the refinement on minimax rates of iRFPCA and Tikhonov estimators to future research. Note that model (11) can be extended to include a finite and fixed number of scalar predictors with slight modification, and the asymptotic properties of $\hat{\beta}$ and $\tilde{\beta}$ remain unchanged.

## 5. Numerical Examples.

5.1. Simulation Studies. We consider two manifolds that are frequently encountered in practice ${ }^{1}$. The first one is the unit sphere $\mathbb{S}^{d}$ which is a compact nonlinear Riemannian submanifold of $\mathbb{R}^{d+1}$ for a positive integer $d$. The sphere can be used to model compositional data, as exhibited in Dai and Müller (2017) which also provides details of the geometry of $\mathbb{S}^{d}$. Here we consider the case of $d=2$. The sphere $\mathbb{S}^{2}$ consists of points $(x, y, z) \in \mathbb{R}^{3}$ satisfying $x^{2}+y^{2}+z^{2}=1$. Since the intrinsic Riemannian geometry of $\mathbb{S}^{2}$ is the same as the one inherited from its ambient space (refereed as ambient geometry hereafter), according to the discussion in Section 2.4, the ambient approach to FPCA and functional linear regression yields the same results as our intrinsic approach.

The other manifold considered is the space of $m \times m$ symmetric positive definite matrices, denoted by $\operatorname{Sym}_{\star}^{+}(m)$. The space $\operatorname{Sym}_{\star}^{+}(m)$ includes nonsingular covariance matrices which naturally arise from the study of DTI data (Dryden, Koloydenko and Zhou, 2009) and functional connectivity (Friston, 2011). Transitionally, it is treated as a convex cone of the linear space $\operatorname{Sym}(m)$ of symmetric $m \times m$ matrices. However, as discussed in Arsigny et al. (2007), this geometry suffers from the undesired swelling effect. To rectify the issue, several intrinsic geometric structures have been proposed, including the affine-invariant metric (Moakher, 2005) which is also known as the trace metric (Chapter XII, Lang 1999). This metric is invariant to affine transformation (i.e., change of coordinates) on $S \in \operatorname{Sym}_{\star}^{+}(m)$ and thus suitable for covariance matrices; see Arsigny et al. (2007) as well as Fletcher and Joshib (2007) for more details. Moreover, the affine-invariant metric has a negative sectional curvature, and thus the Fréchet mean is unique if it exists. In our simulation, we consider $m=3$. We emphasize that the affine-invariant geometry of $\mathrm{Sym}_{\star}^{+}(m)$ is different from the geometry inherited from the linear space $\operatorname{Sym}(m)$. Thus, the ambient RFPCA of Dai and Müller (2017) might yield inferior performance on this manifold.

We simulate data as follows. First, the time domain is set to $\mathcal{T}=[0,1]$. The mean curves for $\mathbb{S}^{2}$ and $\operatorname{Sym}_{\star}^{+}(m)$ are, respectively, $\mu(t)=(\sin \varphi(t) \cos \theta(t), \sin \varphi(t) \sin \theta(t), \cos \varphi(t))$ with $\theta(t)=$ $2 t^{2}+4 t+1 / 2$ and $\varphi(t)=\left(t^{3}+3 t^{2}+t+1\right) / 2$, and $\mu(t)=\left(t^{0.4}, 0.5 t, 0.1 t^{1.5} ; 0.5 t, t^{0.5}, 0.5 t ; 0.1 t^{1.5}, 0.5 t\right.$, $t^{0.6}$ ) that is a $3 \times 3$ matrix. The Riemannian random processes are produced in accordance to $X=\operatorname{Exp}\left(\sum_{k=1}^{20} \sqrt{\lambda_{k}} \xi_{k} \phi_{k}\right)$, where $\xi_{k} \stackrel{i . i . d .}{\sim}$ Uniform $(-\pi / 4, \pi / 4)$ for $\mathbb{S}^{2}$ and $\xi_{k} \stackrel{i . i . d .}{\sim} N(0,1)$ for $\operatorname{Sym}_{\star}^{+}(m)$. We set iRFPCs $\phi_{k}(t)=\left(A \psi_{k}(t)\right)^{T} \mathbf{E}(t)$, where $\mathbf{E}(t)=\left(E_{1}(\mu(t)), \ldots, E_{d}(\mu(t))\right)$ is an orthonormal frame over the path $\mu, \psi_{k}(t)=\left(\psi_{k, 1}(t), \ldots, \psi_{k, d}(t)\right)^{T}$ with $\psi_{k, j}$ being orthonormal Fourier basis functions on $\mathcal{T}$, and $A$ is an orthonormal matrix that is randomly generated but fixed throughout all simulation replicates. We take $\lambda_{k}=2 k^{-1.2}$ for all manifolds. Each curve $X(t)$ is observed at $M=101$ regular design points $t=0,0.01, \ldots, 1$. The slope function is $\beta=\sum_{k=1}^{K} c_{k} \phi_{k}$ with $c_{k}=3 k^{-2} / 2$. Two different types of distribution for $\varepsilon$ in (11) are considered, namely, normal and Student's $t$ distribution with degree of freedom $\mathrm{df}=2.1$. Note that the latter is a heavy-tailed distribution,

[^1]Table 1
The root mean integrated squared error (RMISE) of the estimation of the mean function, and the intrinsic RMISE (iRMISE) and the ambient RMISE (aRMISE) of the estimation for the first two eigenfunctions in the case of $\mathbb{S}^{2}$ manifold. The Monte Carlo standard error based on 100 simulation runs is given in parentheses.

|  | $n=50$ |  | $n=150$ |  | $n=500$ |  |
| :---: | :---: | :---: | :---: | :---: | :---: | :---: |
| $\mu$ | $0.244(0.056)$ | $0.135(0.029)$ |  | $0.085(0.019)$ |  |  |
|  | iRMISE | aRMISE | iRMISE | aRMISE | iRMISE | aRMISE |
| $\phi_{1}$ | $0.279(0.073)$ | $0.331(0.078)$ | $0.147(0.037)$ | $0.180(0.042)$ | $0.086(0.022)$ | $0.106(0.027)$ |
| $\phi_{2}$ | $0.478(0.133)$ | $0.514(0128)$ | $0.264(0.064)$ | $0.287(0.061)$ | $0.147(0.044)$ | $0.167(0.042)$ |

with a smaller df suggesting a heavier tail and $\mathrm{df}>2$ ensuring the existence of variance. In addition, the noise $\varepsilon$ is scaled to make the signal-to-noise ratio equal to 2 . Three different training sample sizes are considered, namely, 50,150 and 500 , while the sample size for test data is 5000 . Each simulation setup is repeated independently 100 times.

First, we illustrate the difference between the intrinsic measure and the ambient counterpart for the discrepancy of two random objects residing on different tangent spaces, through the examples of the sphere manifold $\mathbb{S}^{2}$ and the first two iRFPCs. Recall that the metric of $\mathbb{S}^{2}$ agrees with its ambient Euclidean geometry, so that both iRFPCA and RFPCA essentially yield the same estimates for iRFPCs. We propose to use the intrinsic root mean integrated squared error (iRMISE) $\left\{\mathbb{E}\left\|\hat{\phi}_{k} \Theta_{\Gamma} \phi_{k}\right\|_{\mu}^{2}\right\}^{1 / 2}$ to characterize the difference between $\phi_{k}$ and its estimator $\hat{\phi}_{k}$, while Dai and Müller (2017) adopt the ambient RMISE (aRMISE) $\left\{\mathbb{E}\left\|\hat{\phi}_{k}-\phi_{k}\right\|_{\mathbb{R}^{d_{0}}}^{2}\right\}^{1 / 2}$, as discussed in Section 2.4. The numerical results of iRMISE and aRMISE for $\hat{\phi}_{1}$ and $\hat{\phi}_{2}$, as well as the RMISE for $\hat{\mu}$, are showed in Table 1 . We see that, when $n$ is small and hence $\hat{\mu}$ is not sufficiently close to $\mu$, the difference between iRMISE and aRMISE is visible, while such difference decreases as the sample size grows and $\hat{\mu}$ converges to $\mu$. In particular, aRMISE is always larger than iRMISE since aRMISE contains an additional ambient component that is not intrinsic to the manifold, as illustrated on the left panel of Figure 1.

We now use iRMISE to assess the performance of iRFPCA by comparing to the ambient counterpart RFPCA proposed by Dai and Müller (2017). Table 2 presents the results for the top 5 eigenelements. The first observation is that iRFPCA and RFPCA yield the same results on the manifold $\mathbb{S}^{2}$, which numerically verifies our discussion in Section 2.4. We notice that in Dai and Müller (2017) the quality of estimation of principal components is not evaluated, likely due to the lack of a proper tool to do so. In contrast, our framework of tensor Hilbert space provides an intrinsic gauge (e.g. iRMISE) to naturally compare two vector fields along different curves. For the case of $\operatorname{Sym}_{\star}^{+}(m)$ which is not a Euclidean submanifold, the iRFPCA produces more accurate estimation than RFPCA. In particular, as sample size grows, the estimation error for iRFPCA decreases quickly, while the error of RFPCA persists. This coincides with our intuition that when the geometry induced from the ambient space is not the same as the intrinsic geometry, the ambient RFPCA incurs loss of statistical efficiency, or even worse, inconsistent estimation. In summary, the results $\operatorname{Sym}_{\star}^{+}(m)$ numerically demonstrate that the RFPCA proposed by Dai and Müller (2017) does not apply to manifolds that do not have an ambient space or whose intrinsic geometry differs from its ambient geometry, while our iRFPCA are applicable to such Riemannian manifolds.

For functional linear regression, we adopt iRMISE to quantify the quality of the estimator $\hat{\beta}$ for

Table 2
Intrinsic root integrated mean squared error (iRMISE) of estimation for eigenelements. The first column denotes the manifolds, where $\mathbb{S}^{2}$ is the unit sphere and $\operatorname{Sym}_{\star}^{+}(m)$ is the space of $m \times m$ symmetric positive-definite matrices endowed with the affine-invariant metric. In the second column, $\phi_{1}, \ldots, \phi_{5}$ are the top five intrinsic Riemannian functional principal components. Columns 3-5 are (iRMISE) of the iRFPCA estimators for $\phi_{1}, \ldots, \phi_{5}$ with different sample sizes, while columns 5-8 are iRMISE for the RFPCA estimators. The Monte Carlo standard error based on 100 simulation runs is given in parentheses.

| Manifold | FPC | iRFPCA |  |  | RFPCA |  |  |
| :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: |
|  |  | $n=50$ | $n=150$ | $n=500$ | $n=50$ | $n=150$ | $n=500$ |
|  | $\phi_{1}$ | $0.279(.073)$ | $0.147(.037)$ | $0.086(.022)$ | $0.279(.073)$ | $0.147(.037)$ | $0.086(.022)$ |
|  | $\phi_{2}$ | $0.475(.133)$ | $0.264(.064)$ | $0.147(.044)$ | $0.475(.133)$ | $0.264(.064)$ | $0.147(.044)$ |
|  | $\phi_{3}$ | $0.647(.153)$ | $0.389(.120)$ | $0.206(.054)$ | $0.647(.153)$ | $0.389(.120)$ | $0.206(.054)$ |
|  | $\phi_{4}$ | $0.818(.232)$ | $0.502(.167)$ | $0.261(.065)$ | $0.818(.232)$ | $0.502(.167)$ | $0.261(.065)$ |
|  | $\phi_{5}$ | $0.981(.223)$ | $0.586(.192)$ | $0.329(.083)$ | $0.981(.223)$ | $0.586(.192)$ | $0.329(.083)$ |
| $\operatorname{Sym}_{\star}^{+}(m)$ | $\phi_{1}$ | $0.291(.105)$ | $0.155(.046)$ | $0.085(.025)$ | $0.707(.031)$ | $0.692(.021)$ | $0.690(.014)$ |
|  | $\phi_{2}$ | $0.523(.203)$ | $0.283(.087)$ | $0.143(.040)$ | $0.700(.095)$ | $0.838(.113)$ | $0.684(.055)$ |
|  | $\phi_{3}$ | $0.734(.255)$ | $0.418(.163)$ | $0.206(.067)$ | $0.908(.116)$ | $0.904(.106)$ | $0.981(.039)$ |
|  | $\phi_{4}$ | $0.869(.251)$ | $0.566(.243)$ | $0.288(.086)$ | $0.919(.115)$ | $1.015(.113)$ | $0.800(.185)$ |
|  | $\phi_{5}$ | $1.007(.231)$ | $0.699(.281)$ | $0.378(.156)$ | $0.977(.100)$ | $1.041(.140)$ | $1.029(.058)$ |

slope function $\beta$, and assess the prediction performance by prediction RMSE on independent test dataset. For comparison, we also fit the functional linear model using the principal components produced by RFPCA (Dai and Müller, 2017), and hence we refer this competing method as RFLR. For both methods, the tuning parameter which is the number of principal components included for $\hat{\beta}$, is selected by using an independent validation data of the same size of the training data to ensure fair comparison between two methods. The simulation results are presented in Table 3. As expected, we observe that on $\mathbb{S}^{2}$ both methods produce the same results. For the SPD manifold, in terms of estimation, we see that iRFLR yields far better estimators than RFLR does. Particularly, we again observe that, the quality of RFLR estimators does not improve significantly when sample size increases, in contrast to estimators based on the proposed iRFLR. For prediction, iRFLR outperforms RFLR by a significant margin. Interestingly, comparing to estimation of slope function where the RFLR estimator is much inferior to the iRFLR one, the prediction performance by RFLR is relatively closer to that by iRFLR. We attribute this to the smoothness effect brought by the integration in model (11). Nevertheless, although the integration cancels out certain discrepancy between the intrinsic and the ambient geometry, the loss of efficiency is inevitable for the RFLR method that is bound to the ambient spaces. In addition, we observe that the performance of both methods for Gaussian noise is slightly better than that in the case of heavy-tailed noise.
5.2. Data Application. We apply the proposed iRFPCA and iRFLR to analyze the relationship between functional connectivity and behavioral data from the HCP 900 subjects release (Essen et al., 2013). Although neural effects on language (Binder et al., 1997), emotion (Phana et al., 2002), and fine motor skills (Dayan and Cohen, 2011) have been extensively studied in the literature, scarce is the exploration on human behaviors that do not seem related to neural activities, such as endurance. Nevertheless, a recent research by Raichlen et al. (2016) suggests that endurance can be related to

Table 3
Estimation quality of slope function $\beta$ and prediction of $y$ on test datasets. The second column indicates the distribution of noise, while the third column indicates the manifolds, where $\mathbb{S}^{2}$ is the unit sphere and $\operatorname{Sym}_{\star}^{+}(m)$ is the space of $m \times m$ symmetric positive-definite matrices endowed with the affine-invariant metric. Columns 4-6 are
performance of the $i R F L R$ on estimating the slope curve $\beta$ and predicting the response on new instances of predictors, while columns 7-9 are performance of the RFLR method. Estimation quality of slope curve is quantified by intrinsic root mean integrated squared errors (iRMISE), while the performance of prediction on independent test data is measured by root mean squared errors (RMSE). The Monte Carlo standard error based on 100 simulation runs is given in parentheses.

|  |  |  | iRFLR |  |  | RFLR |  |  |
| :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: |
|  |  |  | $n=50$ | $n=150$ | $n=500$ | $n=50$ | $n=150$ | $n=500$ |
|  | त्व̃ | $\mathbb{S}^{2}$ | 0.507 (0.684) | $0.164(0.262)$ | 0.052 (0.045) | $0.507(0.684)$ | $0.164(0.262)$ | $0.052(0.045)$ |
|  |  | SPD | 1.116 (2.725) | 0.311 (0.362) | 0.100 (0.138) | 2.091 (0.402) | 1.992 (0.218) | 1.889 (0.126) |
|  | $\underset{\underset{\sim}{\mathrm{Y}}}{\underset{\sim}{\mathrm{Y}}}$ | $\mathbb{S}^{2}$ | 0.575 (0.768) | 0.183 (0.274) | 0.053 (0.050) | 0.575 (0.768) | 0.183 (0.274) | 0.053 (0.050) |
|  |  | SPD | 1.189 (2.657) | 0.348 (0.349) | 0.108 (0.141) | 2.181 (0.439) | 1.942 (0.209) | 1.909 (0.163) |
|  | $\begin{aligned} & \text { డ్ } \\ & \text { वี } \\ & \text { Oan } \end{aligned}$ | $\mathbb{S}^{2}$ | 0.221 (0.070) | 0.135 (0.046) | 0.083 (0.019) | 0.221 (0.070) | 0.135 (0.046) | 0.083 (0.019) |
|  |  | SPD | 0.496 (0.184) | $0.284(0.092)$ | 0.165 (0.062) | 0.515 (0.167) | 0.328 (0.083) | 0.248 (0.047) |
|  | $\underset{\underset{\sim}{\mathrm{J}}}{\underset{\sim}{\mathrm{I}}}$ | $\mathbb{S}^{2}$ | 0.251 (0.069) | 0.142 (0.042) | 0.088 (0.020) | 0.251 (0.069) | 0.142 (0.042) | 0.088 (0.020) |
|  |  | SPD | 0.532 (0.189) | 0.298 (0.097) | 0.172 (0.066) | 0.589 (0.185) | 0.360 (0.105) | 0.268 (0.051) |

functional connectivity. Our goal is to study the endurance performance of subjects based on their functional connectivity.

The data consists of $n=330$ subjects who are healthy young adults, in which each subject is asked to walk for 2 minutes and the distance in feet is recorded. Also, each subject participates in a motor task, where participants are asked to act according to presented visual cues, such as tapping their fingers, squeezing their toes, or moving their tongue. During the task, the brain of each subject is scanned and the neural activities are recorded at 284 equispaced time points. After preprocessing, the average BOLD (blood-oxygen-level dependent) signals at 68 different brain regions are obtained. The details of experiment and data acquisition can be found in the reference manual of WU-Minn HCP 900 Subjects Data Release that is available on the website of human connectome project.

Our study focuses on $m=6$ regions that are related to primary motor cortex, including precentral gyrus, Broca's area, etc. At each design time point $t$, the functional connectivity of the $i$ th subject is represented by the covariance matrix $S_{i}(t)$ of BOLD signals from regions of interest (ROI). To practically compute $S_{i}(t)$, let $V_{i t}$ be an $m$-dimensional column vector that represents the BOLD signals at time $t$ from the $m$ ROIs of the $i$ th subject. We then adopt a local sliding window approach (Park et al., 2017) to compute $S_{i}(t)$ by

$$
S_{i}(t)=\frac{1}{2 h+1} \sum_{j=t-h}^{t+h}\left(V_{i j}-\bar{V}_{i t}\right)\left(V_{i j}-\bar{V}_{i t}\right)^{T} \quad \text { with } \quad \bar{V}_{i t}=\frac{1}{2 h+1} \sum_{j=t-h}^{t+h} V_{i j}
$$

where $h$ is a positive integer that represents the length of the sliding window to compute $S_{i}(t)$ for $t=h+1, h, \ldots, 284-h$. Without loss of generality, we reparameterize each $S_{i}(\cdot)$ from the domain $[h+1,284-h]$ to $[1,284-2 h]$. In practice, the window length $h$ is required to be sufficiently large to make $S_{t}$ nonsingular, but not too large in order to avoid significant bias. We found that the analysis below is not sensitive to the choice of $h$ as long as it is in a reasonable range, e.g., $h \in[m, 3 m]$.

Thus, we simply set $h=2 m$ throughout our study.
The functional predictors are then the functions $S_{i}(\cdot)$ whose values are covariance matrices, or SPDs, and hence is regarded as a curve on $\operatorname{Sym}_{\star}^{+}(m)$. The heat maps of the first two iRFPCs along 8 time points are depicted in the top two rows of Figure 2, which shows the modes of variation in the functional connectivity with contrasts between early and middle/later times in both iRFPCs. The scalar response is the distance that subjects walked in two minutes. A scatterplot of the first two principal component scores labeled by the response variable for 10 randomly selected subjects is given in Figure 3 respectively for iRFPCA and RFPCA, to visualize the difference of scores produced by these methods. Gender, age, gait speed and strength are included as baseline covariates, selected by the forward-stepwise selection method (Section 3.3, Hastie, Tibshirani and Friedman 2009). Among these covariates, gender and age are in accordance with the common sense about endurance, while gait speed and muscle strength could be influential since endurance is measured by the distance walked in two minutes. Our primary interest is to assess the significance of the functional predictor when effect of the baseline covariates is controlled.

To fit the intrinsic functional linear model, we adopt the cross-validation procedure to select the number of components to be included in representing the Riemannian functional predictor and the Riemannian slope function $\beta$. For assessment, we conduct 100 runs of 10 -fold cross-validation, where in each run we permute the data independently. In each run, the model is fitted on $90 \%$ data and the MSE for predicting the walking distance is computed on the other $10 \%$ data for both iRFLR and RFLR methods. The fitted intrinsic Riemannian slope function $\log _{\hat{\mu}} \hat{\beta}$ displayed in the bottom panel of Figure 2 shows the pattern of weight changes. The MSE for iRFLR is reduced by around $9.7 \%$, compared to that for RFLR. Moreover, the $R^{2}$ for iRFLR is 0.338 , with a p-value 0.012 based on a permutation test of 1000 permutations, which is significant at level $5 \%$. In contrast, the $R^{2}$ for RFLR drops to 0.296 and the p-value is 0.317 that does not spell significance at all.


Figure 2. The $\hat{\phi}_{1}, \hat{\phi}_{2}$ and $\log _{\hat{\mu}} \hat{\beta}$ at time points $t=1+24 k, k=1, \ldots, 8$.

## Acknowledgements

This research is partially supported by Natural Science and Engineering Research Council of Canada. Data were provided by the Human Connectome Project, WU-Minn Consortium (Principal Investigators: David Van Essen and Kamil Ugurbil; 1U54MH091657) funded by the 16 NIH


Figure 3. Scatterplot of the first two Riemannian functional principal component scores labeled by the response variable endurance in iRFPCA (left panel) and RFPCA (right panel) for 10 randomly selected subjects.

Institutes and Centers that support the NIH Blueprint for Neuroscience Research; and by the McDonnell Center for Systems Neuroscience at Washington University.

## APPENDIX A: BACKGROUND ON RIEMANNIAN MANIFOLD

We introduce geometric concepts related to Riemannian manifolds from an intrinsic perspective without referring to any ambient space.

A smooth manifold is a differentiable manifold with all transition maps being $C^{\infty}$ differentiable. For each point $p$ on the manifold $\mathcal{M}$, there is a linear space $T_{p} \mathcal{M}$ of tangent vectors which are derivations. A derivation is a linear map that sends a differentiable function on $\mathcal{M}$ into $\mathbb{R}$ and satisfies the Leibniz property. For example, if $D_{v}$ is the derivation associate with the tangent vector $v$ at $p$, then $D_{v}(f g)=\left(D_{v} f\right) \cdot g(p)+f(p) \cdot D_{v}(g)$ for any $f, g \in A(\mathcal{M})$, where $A(\mathcal{M})$ is a collection of real-valued differentiable functions on $\mathcal{M}$. For submanifolds of a Euclidean space $\mathbb{R}^{d_{0}}$ for some $d_{0}>0$, tangent vectors are often perceived as vectors in $\mathbb{R}^{d_{0}}$ that are tangent to the submanifold surface. If one interprets a Euclidean tangent vector as a directional derivative along the vector direction, then Euclidean tangent vectors coincide with our definition of tangent vectors on a general manifold. The linear space $T_{p} \mathcal{M}$ is called the tangent space at $p$. The disjoint union of tangent spaces at each point constitutes the tangent bundle, which is also equipped with a smooth manifold structure induced by $\mathcal{M}$. The tangent bundle of $\mathcal{M}$ is conventionally denoted by $T \mathcal{M}$. A (smooth) vector field $V$ is a map from $\mathcal{M}$ to $T \mathcal{M}$ such that $V(p) \in T_{p} \mathcal{M}$ for each $p \in \mathcal{M}$. It is also called a smooth section of $T \mathcal{M}$. Noting that a tangent vector is a tensor of type ( 0,1 ), a vector field can be viewed as a kind of tensor field, which assigns a tensor to each point on $\mathcal{M}$. A vector field along a curve $\gamma: I \rightarrow \mathcal{M}$ on $\mathcal{M}$ is a map $V$ from an interval $I \subset \mathbb{R}$ to $T \mathcal{M}$ such that $V(t) \in T_{\gamma(t)} \mathcal{M}$. For a smooth function from a manifold $\mathcal{M}$ and to another manifold $\mathcal{N}$, the differential $\mathrm{d} \varphi_{p}$ of $f$ at $p \in \mathcal{M}$ is a linear map from $T_{p} \mathcal{M}$ to $T_{\varphi(p)} \mathcal{N}$, such that $\mathrm{d} \varphi_{p}(v)(f)=D_{v}(f \circ \varphi)$ for all $f \in A(\mathcal{M})$ and $v \in T_{p} \mathcal{M}$.

An affine connection $\nabla$ on $\mathcal{M}$ is a bilinear mapping that sends a pair of smooth vector fields $(U, V)$ to another smooth vector field $\nabla_{U} V$ and satisfies $\nabla_{f U} V=f \nabla_{U} V$ and the Leibniz rule in the second argument, i.e., $\nabla_{U}(f V)=\mathrm{d} f(U) V+f \nabla_{U} V$ for smooth real-valued functions $f$ on $\mathcal{M}$. An interesting fact is that $\left(\nabla_{U} V\right)(p)$ is determined by $U(p)$ and a neighborhood of $V(p)$. A vector
field $V$ is parallel if $\nabla_{U} V=0$ for all vector fields $U$. When a connection is interpreted as the covariant derivative of a vector field with respect to another one, a parallel vector field is intuitively a constant vector field in some sense, as vanishing derivative indicates constant. Similarly, a vector field $U$ along a curve $\gamma$ is parallel if $\nabla_{\dot{\gamma}(t)} U=0$ for all $t \in I$, where $\dot{\gamma}(t)$ denotes the differential of $\gamma$ at $t$. Given a tangent vector $u$ at $\gamma(0)$, there exists a unique vector field $U$ along $\gamma$ such that $\nabla_{\dot{\gamma}(t)} U=0$ and $U(\gamma(0))=u$. In this case, $U(\gamma(t))$ is the parallel transport of $u$ along $\gamma$ from $\gamma(0)$ to $\gamma(t)$. In particular, if $\nabla_{\dot{\gamma}} \dot{\gamma}=0$, then $\gamma$ is called a geodesic of the connection $\nabla$. Also, for any $u \in T_{\gamma(0)} \mathcal{M}$, there exists a unique geodesic $\gamma$ such that $\nabla_{\dot{\gamma}} \dot{\gamma}=0$ and $\dot{\gamma}(0)=u$. Then the exponential map at $p=\gamma(0)$, denoted by $\operatorname{Exp}_{p}$, is defined by $\operatorname{Exp}_{p}(u)=\gamma(1)$.

A Riemannian manifold is a smooth manifold with a metric tensor $\langle\cdot, \cdot\rangle$, such that for each $p \in \mathcal{M}$, the tensor $\langle\cdot, \cdot\rangle_{p}$ defines an inner product on $T_{p} \mathcal{M} \times T_{p} \mathcal{M}$. For Riemannian manifold, the fundamental theorem of Riemannian geometry says that there is a unique connection that is 1 ) compatible with the metric tensor, i.e, for every pair of vector fields $U$ and $V$, and every vector $\left.v \in T_{p} \mathcal{M}, v(\langle U, V\rangle)=\left\langle\nabla_{v} U, V\right\rangle+\left\langle U, \nabla_{v} V\right\rangle ; 2\right)$ torsion-free, i.e., for any vector fields $U$ and $V$, $\nabla_{U} V-\nabla_{V} U=[U, V]$, where $[U, V]$ denotes the Lie bracket of a pair of vector fields $(U, V)$, and is defined by $[U, V](f)=U(V(f))-V(U(f))$ for all $f \in A(\mathcal{M})$ (by definition, [U,V] is also a vector field). This connection is called Levi-Civita connection, which is uniquely determined by the Riemannian metric tensor. The Riemannian curvature tensor $R$ is a map that sends a triple of smooth vector fields ( $U, V, W$ ) to another vector field $R(U, V) W=\nabla_{U} \nabla_{V} W-\nabla_{V} \nabla_{U} W-\nabla_{[U, V]} W$. Then the sectional curvature is defined for a 2-plane spanned by two linearly independent tangent vector at the same point $p$, given by

$$
K(u, v)=\frac{\langle R(u, v) v, u\rangle_{p}}{\langle u, u\rangle_{p}\langle v, v\rangle_{p}-\langle u, v\rangle_{p}^{2}} \in \mathbb{R} .
$$

The metric tensor also induces a distance function $d_{\mathcal{M}}$ on $\mathcal{M}$ that turns a connected Riemannian manifold $\mathcal{M}$ into a metric space. The function $d_{\mathcal{M}}$ is defined in the following way. For a continuously differentiable curve $\gamma:[a, b] \rightarrow \mathcal{M}$, the length of $\gamma$ is given by $L(\gamma)=\int_{a}^{b} \sqrt{\langle\dot{\gamma}(t), \dot{\gamma}(t)\rangle}$ d $t$. Then $d_{\mathcal{M}}(p, q)$ is the infimum of $L(\gamma)$ over all continuously differentiable curves joining $p$ and $q$. For a connected and complete Riemannian, given two points on the manifold, there is a minimizing geodesic connecting these two points.

## APPENDIX B: PROOFS OF MAIN THEOREMS

Proof of Theorem 1. We first show that $\mathscr{T}(\mu)$ is a Hilbert space. It is sufficient to prove that the inner product space $\mathscr{T}(\mu)$ is complete. Suppose $\left\{V_{n}\right\}$ is a Cauchy sequence in $\mathscr{T}(\mu)$. We will later show that there exists a subsequence $\left\{V_{n_{k}}\right\}$ such that

$$
\begin{equation*}
\sum_{k=1}^{\infty}\left|V_{n_{k+1}}(t)-V_{n_{k}}(t)\right|<\infty, \quad \text { v-a.s. } \tag{14}
\end{equation*}
$$

Since $T_{\mu(t)} \mathcal{M}$ is complete, the limit $V(t)=\lim _{k \rightarrow \infty} V_{n_{k}}(t)$ is $v$-a.s. well defined and in $T_{\mu(t)} \mathcal{M}$. Fix any $\epsilon>0$ and choose $N$ such that $n, m \geq M$ implies $\left\|V_{n}-V_{m}\right\|_{\mu} \leq \epsilon$. Fatou's lemma applying to the function $\left|V(t)-V_{m}(t)\right|$ implies that if $m \geq N$, then $\left\|V-V_{m}\right\|_{\mu}^{2} \leq \liminf _{k \rightarrow \infty}\left\|V_{n_{k}}-V_{m}\right\|_{\mu}^{2} \leq \epsilon^{2}$. This shows that $V-V_{m} \in \mathscr{T}(\mu)$. Since $V=\left(V-V_{m}\right)+V_{m}$, we see that $V \in \mathscr{T}(\mu)$. The arbitrariness of $\epsilon$ implies that $\lim _{m \rightarrow \infty}\left\|V-V_{m}\right\|_{\mu}=0$. Because $\left\|V-V_{n}\right\|_{\mu} \leq\left\|V-V_{m}\right\|_{\mu}+\left\|V_{m}-V_{n}\right\|_{\mu} \leq 2 \epsilon$, we conclude that $V_{n}$ converges to $V$ in $\mathscr{T}(\mu)$.

It remains to show (14). To do so, we choose $\left\{n_{k}\right\}$ so that $\left\|V_{n_{k}}-V_{n_{k+1}}\right\|_{\mu} \leq 2^{-k}$. This is possible since $V_{n}$ is a Cauchy sequence. Let $U \in \mathscr{T}(\mu)$. By Cauchy-Schwarz inequality, $\int_{\mathcal{T}}|U(t)| \cdot \mid V_{n_{k}}(t)$ $V_{n_{k+1}}(t) \mid \mathrm{d} v(t) \leq\|U\|_{\mu}\left\|V_{n_{k}}-V_{n_{k+1}}\right\|_{\mu} \leq 2^{-k}\|U\|_{\mu}$. Thus, $\sum_{k} \int_{\mathcal{T}}|U(t)| \cdot\left|V_{n_{k}}(t)-V_{n_{k+1}}(t)\right| \mathrm{d} v(t) \leq\|U\|_{\mu}<$ $\infty$. Then (14) follows, because otherwise, if the series diverges on a set $A$ with $v(A)>0$, then a choice of $U$ such that $|U(t)|>0$ for $t \in A$ contradicts the above inequality.

Now let $\mathbf{E}$ be a measurable orthonormal frame. For every element $U \in \mathscr{T}(\mu)$, the coordinate representation of $U$ with respect to $\mathbf{E}$ is denoted by $U_{\mathbf{E}}$. One can see that $U_{\mathbf{E}}$ is an element in the Hilbert space $\mathcal{L}^{2}\left(\mathcal{T}, \mathbb{R}^{d}\right)$ of square integrable $\mathbb{R}^{d}$-valued measurable functions with norm $\|f\|_{\mathcal{L}^{2}}=\left\{\int_{\mathcal{T}}|f(t)|^{2} \mathrm{~d} v(t)\right\}^{1 / 2}$ for $f \in \mathcal{L}^{2}\left(\mathcal{T}, \mathbb{R}^{d}\right)$. If we define the map $\Upsilon: \mathscr{T}(\mu) \rightarrow \mathcal{L}^{2}\left(\mathcal{T}, \mathbb{R}^{d}\right)$ by $\Upsilon(U)=U_{\mathbf{E}}$, we can immediately see that $\Upsilon$ is a linear map. It is also surjective, because for any $f \in \mathcal{L}^{2}\left(\mathcal{T}, \mathbb{R}^{d}\right)$, the vector field $U$ along $\mu$ given by $U_{f}(t)=f(t) \mathbf{E}(\mu(t))$ for $t \in \mathcal{T}$ is an element in $\mathscr{T}(\mu)$, since $\left\|U_{f}\right\|_{\mu}=\|f\|_{\mathcal{L}^{2}}$. It can be also verified that $\Upsilon$ preserves the inner product. Therefore, it is a Hilbertian isomorphism. Since $\mathcal{L}^{2}\left(\mathcal{T}, \mathbb{R}^{d}\right)$ is separable, the isomorphism between $\mathcal{L}^{2}\left(\mathcal{T}, \mathbb{R}^{d}\right)$ and $\mathscr{T}(\mu)$ implies that $\mathscr{T}(\mu)$ is also separable.

Proof of Proposition 2. The regularity conditions on $f, h$ and $\gamma$ ensure that $\Gamma$ and $\Phi$ are measurable. Part 1,2 and 6 can be deduced from the fact that $\mathcal{P}_{f, h}$ is a unitary operator between two finite-dimensional real Hilbert spaces and its inverse is $\mathcal{P}_{h, f}$. To reduce notational burden, we shall suppress the subscripts $f, h$ from $\Gamma_{f, h}$ and $\Phi_{f, h}$ below. For Part 3,

$$
\left.(\Phi \mathcal{A})(\Gamma U)=\Gamma\left(\mathcal{A} \Gamma^{*} \Gamma U\right)\right)=\Gamma(\mathcal{A} U) .
$$

To prove Part 4, assume $V \in \mathscr{T}(g)$. Then, noting that $\Gamma\left(\Gamma^{*} V\right)=V$ and $\Gamma^{*}(\Gamma U)=U$, we have

$$
(\Phi \mathcal{A})\left(\left(\Phi \mathcal{A}^{-1}\right) V\right)=(\Phi \mathcal{A})\left(\Gamma\left(\mathcal{A}^{-1} \Gamma^{*} V\right)\right)=\Gamma\left(\mathcal{A} \Gamma^{*}\left(\Gamma\left(\mathcal{A}^{-1} \Gamma^{*} V\right)\right)\right)=\Gamma\left(\mathcal{A} \mathcal{A}^{-1} \Gamma^{*} V\right)=\Gamma\left(\Gamma^{*} V\right)=V,
$$

and

$$
\left(\Phi \mathcal{A}^{-1}\right)(\Phi \mathcal{A} V)=\left(\Phi \mathcal{A}^{-1}\right)\left(\Gamma\left(\mathcal{A} \Gamma^{*} V\right)\right)=\Gamma\left(\mathcal{A}^{-1} \Gamma^{*}\left(\Gamma\left(\mathcal{A} \Gamma^{*} V\right)\right)\right)=\Gamma\left(\mathcal{A}^{-1} \mathcal{A} \Gamma^{*} V\right)=\Gamma\left(\Gamma^{*} V\right)=V .
$$

Part 5 is seen by the following calculation: for $V \in \mathscr{T}(g)$,

$$
\begin{aligned}
\left(\Phi_{f, g} \sum c_{k} \varphi_{k} \otimes \varphi_{k}\right) V & =\Gamma\left(\sum c_{k}\left\langle\left\langle\varphi_{k}, \Gamma^{*} V\right\rangle_{f} \varphi_{k}\right)=\sum c_{k}\left\langle\left\langle\phi_{k}, \Gamma^{*} V\right\rangle_{f} \Gamma \varphi_{k}\right.\right. \\
& \left.=\sum c_{k}\left\langle\Gamma \varphi_{k}, V\right\rangle\right\rangle_{g} \Gamma \varphi_{k}=\left(\sum c_{k} \Gamma \varphi_{k} \otimes \Gamma \varphi_{k}\right) V .
\end{aligned}
$$

Proof of Proposition 4. The case $\kappa \geq 0$ is already given by Dai and Müller (2017) with $C=1$. Suppose $\kappa<0$. The second statement follows from the first one if we let $O=\mu(t), P=X(t)$ and $Q=X_{K}(t)$ for any fixed $t$ and note that $C$ is independent of $t$.

For the first statement, the inequality is clearly true if $P=O, Q=O$ or $P=Q$. Now suppose $O, P$ and $Q$ are distinct points on $\mathcal{M}$. The minimizing geodesic curves between these points form a geodesic triangle on $\mathcal{M}$. By Toponogov's theorem (the hinge version), $d_{\mathcal{M}}(P, Q) \leq d_{\mathbb{M}_{k}}\left(P^{\prime}, Q^{\prime}\right)$, where $\mathbb{M}_{\kappa}$ is the model space with constant sectional curvature $\kappa$. For $\kappa<0$, it is taken as the hyperboloid with curvature $\kappa$. Let $a=d_{\mathcal{M}}(O, P), b=d_{\mathcal{M}}(O, Q)$ and $c=d_{\mathcal{M}}(P, Q)$. The interior angle of geodesics connecting $O$ to $P$ and $O$ to $Q$ is denoted by $\gamma$. Denote $\delta=\sqrt{-\kappa}$, the law of cosine on $\mathbb{M}_{\kappa}$ gives

$$
\cosh (\delta c)=\{\cosh (\delta a) \cosh (\delta b)-\sinh (\delta a) \sinh (\delta b)\}+\{\sinh (\delta a) \sinh (\delta b)(1-\cos \gamma)\} \equiv I_{1}+I_{2} .
$$

By definition that $\cosh (x)=\left(e^{x}+e^{-x}\right) / 2$ and $\sinh (x)=\left(e^{x}-e^{-x}\right) / 2, I_{1}=\left(e^{\delta a-\delta b}+e^{\delta b-\delta a}\right) / 2$. By Taylor expansion $e^{x}=\sum x^{k} / k$ !, we have

$$
\begin{equation*}
I_{1}=\sum_{k=0}^{\infty} \frac{(\delta a-\delta b)^{2 k}}{(2 k)!} \leq e^{\delta^{2}(a-b)^{2}} \leq e^{\delta^{2} h^{2}}, \tag{15}
\end{equation*}
$$

where $h=\left|\log _{O} P-\log _{O} Q\right|$ with $|\cdot|$ denotes the norm of tangent vectors. The last inequality is due to the fact that $a=\left|\log _{O} P\right|, b=\left|\log _{O} Q\right|$.

For the second term $I_{2}$, we first have $\sinh (\delta a) \sinh (\delta b) \leq \delta^{2} a b e^{\delta^{2} a^{2}+\delta^{2} b^{2}}$. Also, the Euclidean law of cosine implies that $h^{2}=a^{2}+b^{2}-2 a b \cos \gamma$ and $2 a b(1-\cos \gamma)=h^{2}-(a-b)^{2} \leq h^{2}$. Therefore,

$$
I_{2} \leq \delta^{2} h^{2} e^{\delta^{2} a^{2}+\delta^{2} b^{2}} / 2 \leq \delta^{2} h^{2} e^{\delta^{2} h^{2}+2 \delta^{2} a b \cos \gamma} / 2 \leq \delta^{2} h^{2} e^{\delta^{2} h^{2}} e^{2 D^{2} \delta^{2}} / 2,
$$

where $D$ is the diameter of $\mathcal{K}$. Combining with (15),

$$
I_{1}+I_{2} \leq e^{\delta^{2} h^{2}}+\delta^{2} h^{2} e^{\delta^{2} h^{2}} e^{2 D^{2} \delta^{2}} / 2 \leq e^{\delta^{2} h^{2}}\left(1+A \delta^{2} h^{2}\right) / 2 \leq e^{\delta^{2} h^{2}+A \delta^{2} h^{2}}=e^{(1+A) \delta^{2} h^{2}}=e^{B h^{2}},
$$

where $A=e^{2 D^{2} \delta^{2}}$ and $B=(1+A) \delta^{2}$ constants uniform for $t$. Therefore

$$
\delta c=\cosh ^{-1}\left(I_{1}+I_{2}\right)=\log \left(I_{1}+I_{2}+\sqrt{\left(I_{1}+I_{2}\right)^{2}-1}\right) \leq \log \left(e^{B h^{2}}+\sqrt{e^{2 B h^{2}-1}}\right) .
$$

It can be shown that $\sqrt{e^{2 B x}-1} \leq e^{B x} \sqrt{2 B x}$. Thus,

$$
\delta c \leq \log \left(e^{B h^{2}}(1+\sqrt{2 B} h)\right) \leq \log \left(e^{B h^{2}+\sqrt{2 B} h}\right) \leq B h^{2}+\sqrt{2 B} h \leq 2 B D h+\sqrt{2 B} h=\sqrt{C} \delta h,
$$

where $C=\{(2 B D+\sqrt{2 B}) / \delta\}^{2}$, or in other words, $d_{\mathcal{M}}(P, Q) \leq \sqrt{C}\left|\log _{O} P-\log _{O} Q\right|$.
Proof of Proposition 5. Part 1 follows from a simple calculation. To lighten notations, let $\mathbf{f}^{T} \mathbf{E}$ denote $\mathbf{f}^{T}(\cdot) \mathbf{E}(\mu(\cdot))$ for a $\mathbb{R}^{d}$ valued function defined on $\mathcal{T}$. Suppose $\phi_{\mathbf{E}, k}$ is the coordinate of $\phi_{k}$ under E. Because

$$
\left(\mathcal{C}_{\mathbf{E}} \phi_{\mathbf{E}, k}\right)^{T} \mathbf{E}=\mathbb{E}\left\langle Z_{\mathbf{E}}, \phi_{\mathbf{E}, k}\right\rangle Z_{\mathbf{E}} \mathbf{E}=\mathbb{E}\left\langle\left\langle\log _{\mu} X, \phi_{k}\right\rangle\right\rangle_{\mu} \log _{\mu} X=\lambda_{k} \phi_{k}=\lambda_{k} \phi_{\mathbf{E}, k}^{T} \mathbf{E}
$$

one concludes that $\mathcal{C}_{\mathbf{E}} \phi_{\mathbf{E}, k}=\lambda_{k} \phi_{\mathbf{E}, k}$ and hence $\phi_{\mathbf{E}, k}$ is an eigenfunction of $\mathcal{C}_{\mathbf{E}}$ corresponding to the eigenvalue $\lambda_{k}$. Other results in Part 2 and 3 have been derived in Section 3. The continuity of $X$ and $\mathbf{E}$, in conjunction with $\mathbb{E}\left\|\log _{\mu} X\right\|_{\mu}^{2}<\infty$, implies that $Z_{\mathbf{E}}$ is a mean square continuous random process and the joint measurability of $X$ passes to $Z_{\mathbf{E}}$. Then $Z_{\mathbf{E}}$ can be regarded a random element of the Hilbert space $\mathcal{L}^{2}(\mathcal{T}, \mathscr{B}(\mathcal{T}), v)$ that is isomorphic to $\mathscr{T}(\mu)$. Also, the isomorphism maps $Z_{\mathbf{E}}$ to $X$ for each $\omega$ in the sample space. Then, Part 4 follows from Theorem 7.4.3 of Hsing and Eubank (2015).

Proof of Theorem 6. The strong consistency stated in Part 2 is an immediate consequence of Lemma 12. For Part 1, to prove continuity of $\mu$, fix $t \in \mathcal{T}$. Let $\mathcal{K} \supset \mathcal{U}$ be compact. By B.3, $c:=\sup _{p \in \mathcal{K}} \sup _{s \in \mathcal{T}} \mathbb{E} d_{\mathcal{M}}^{2}(\mu(t), X(s))<\infty$. Thus,

$$
\begin{aligned}
|F(\mu(t), s)-F(\mu(s), s)| & \leq|F(\mu(t), t)-F(\mu(s), s)|+|F(\mu(t), s)-F(\mu(t), t)| \\
& \leq \sup _{p \in \mathcal{K}}|F(p, t)-F(p, s)|+2 c \mathbb{E} d_{\mathcal{M}}(X(s), X(t)) \leq 4 c \mathbb{E} d_{\mathcal{M}}(X(s), X(t)) .
\end{aligned}
$$

The continuity assumption of sample paths implies $\mathbb{E} d_{\mathcal{M}}(X(s), X(t)) \rightarrow 0$ as $s \rightarrow t$. Then by condition B.5, $d_{\mathcal{M}}(\mu(t), \mu(s)) \rightarrow 0$ as $s \rightarrow t$, and the the continuity of $\mu$ follows. The uniform continuity follows from the compactness of $\mathcal{T}$. Given Lemma 11 and 12, the a.s. continuity of $\hat{\mu}$ can be derived in a similar way. The first statement of Part 4 is a corollary of Part 3, while the second statement follows from the first one and the compactness of $\mathcal{T}$. It remains to show Part 3 in order to conclude the proof, as follows.

Let $V_{t, i}(p)=\log _{p} X_{i}(t)$ and $\gamma_{t, p}$ be the minimizing geodesic from $\mu(t)$ to $p \in \mathcal{M}$ at unit time. The first-order Taylor series expansion at $\mu(t)$ yields

$$
\begin{align*}
\mathcal{P}_{\hat{\mu}(t), \mu(t)} \sum_{i=1}^{n} V_{t, i}(\hat{\mu}(t)) & =\sum_{i=1}^{n} V_{t, i}(\mu(t))+\sum_{i=1}^{n} \nabla_{\gamma_{t, \hat{\mu}(t)}^{\prime}(0)} V_{t, i}(\mu(t))+\Delta_{t}(\hat{\mu}(t)) \gamma_{t, \hat{\mu}(t)}^{\prime}(0) \\
& =\sum_{i=1}^{n} V_{t, i}(\mu(t))-\sum_{i=1}^{n} H_{t}(\mu(t)) \gamma_{t, \hat{\mu}(t)}^{\prime}(0)+\Delta_{t}(\hat{\mu}(t)) \gamma_{t, \hat{\mu}(t)}^{\prime}(0), \tag{16}
\end{align*}
$$

where an expression for $\Delta_{t}$ is provided in the proof of Lemma 10.
Since $\sum_{i=1}^{n} V_{t, i}(\hat{\mu}(t))=\sum_{i=1}^{n} \log _{\hat{\mu}(t)} X_{i}(t)=0$, we deduce from (16) that

$$
\begin{equation*}
\frac{1}{n} \sum_{i=1}^{n} \log _{\mu(t)} X_{i}-\left(\frac{1}{n} \sum_{i=1}^{n} H_{t, i}(\mu(t))-\frac{1}{n} \Delta_{t}(\hat{\mu}(t))\right) \log _{\mu(t)} \hat{\mu}(t)=0 \tag{17}
\end{equation*}
$$

By LLN, $\frac{1}{n} \sum_{i=1}^{n} H_{t, i}(\mu(t)) \rightarrow \mathbb{E} H_{t}(\mu(t))$ in probability, while $\mathbb{E} H_{t}(\mu(t))$ is invertible for all $t$ by condition B.6. In light of Lemma 10, this result suggests that with probability tending to one, for all $t \in \mathcal{T}, \frac{1}{n} \sum_{i=1}^{n} H_{t, i}(\mu(t))-\frac{1}{n} \Delta_{t}(\hat{\mu}(t))$ is invertible, and also

$$
\left(\frac{1}{n} \sum_{i=1}^{n} H_{t, i}(\mu(t))-\frac{1}{n} \Delta_{t}(\hat{\mu}(t))\right)^{-1}=\left\{\mathbb{E} H_{t}(\mu(t))\right\}^{-1}+o_{P}(1),
$$

and according to (17),

$$
\log _{\mu(t)} \hat{\mu}(t)=\left\{\mathbb{E} H_{t}(\mu(t))\right\}^{-1}\left(\frac{1}{n} \sum_{i=1}^{n} \log _{\mu(t)} X_{i}(t)\right)+o_{P}(1)
$$

where the $o_{P}(1)$ terms do not depend on $t$. Given this, we can now conclude the proof of Part 3 by applying a central limit theorem in Hilbert spaces (Aldous, 1976) to establish that the process $\frac{1}{\sqrt{n}} \sum_{i=1}^{n}\left\{\mathbb{E} H_{t}(\mu(t))\right\}^{-1} \log _{\mu(t)} X_{i}(t)$ converges to a Gaussian measure on tensor Hilbert space $\mathscr{T}(\mu)$ with covariance operator $\mathcal{C}(\cdot)=\mathbb{E}\left(\langle V V \cdot \cdot\rangle_{\mu} V\right)$ for a random element $V(t)=\left\{\mathbb{E} H_{t}(\mu(t))\right\}^{-1} \log _{\mu(t)} X(t)$ in the tensor Hilbert space $\mathscr{T}(\mu)$.

Proof of Theorem 7. Note that

$$
\begin{aligned}
\Phi \hat{\mathcal{C}}-\mathcal{C}= & n^{-1} \sum\left(\Gamma \log _{\hat{\mu}} X_{i}\right) \otimes\left(\Gamma \log _{\hat{\mu}} X_{i}\right)-\mathcal{C} \\
= & n^{-1} \sum\left(\log _{\mu} X_{i}\right) \otimes\left(\log _{\mu} X_{i}\right)-\mathcal{C} \\
& +n^{-1} \sum\left(\Gamma \log _{\hat{\mu}} X_{i}-\log _{\mu} X_{i}\right) \otimes\left(\log _{\mu} X_{i}\right) \\
& +n^{-1} \sum\left(\log _{\mu} X_{i}\right) \otimes\left(\Gamma \log _{\hat{\mu}} X_{i}-\log _{\mu} X_{i}\right) \\
& +n^{-1} \sum\left(\Gamma \log _{\hat{\mu}} X_{i}-\log _{\mu} X_{i}\right) \otimes\left(\Gamma \log _{\hat{\mu}} X_{i}-\log _{\mu} X_{i}\right) \\
\equiv & A_{1}+A_{2}+A_{3}+A_{4} .
\end{aligned}
$$

For $A_{2}$, it is seen that
$\left\|A_{2}\right\|_{H S}^{2} \leq$ const. $\frac{1}{n^{2}} \sum_{i=1}^{n} \sum_{j=1}^{n}\left(\left\|\log _{\mu} X_{i}\right\|_{\mu}^{2}+\left\|\log _{\mu} X_{j}\right\|_{\mu}^{2}\right)\left(\left\|\Gamma \log _{\hat{\mu}} X_{i}-\log _{\mu} X_{i}\right\|_{\mu}^{2}+\left\|\Gamma \log _{\hat{\mu}} X_{j}-\log _{\mu} X_{j}\right\|_{\mu}^{2}\right)$.
With smoothness of $d_{\mathcal{M}}^{2}$, continuity of $\mu$ and compactness of $\mathcal{T}$, one can show that $\sup _{t \epsilon \mathcal{T}}\left\|H_{t}(\mu(t))\right\|<$ $\infty$. By the uniform consistency of $\hat{\mu}$, with the same Taylor series expansion in (16) and the technique in the proof of Lemma 10, it can be established that $n^{-1} \sum_{i=1}^{n}\left\|\Gamma \log _{\hat{\mu}} X_{i}-\log _{\mu} X_{i}\right\|_{\mu}^{2}\left\|\log _{\mu} X_{i}\right\|_{\mu}^{2} \leq$ const. $\left(1+o_{P}(1)\right) \sup _{t \in \mathcal{T}} d_{\mathcal{M}}^{2}(\hat{\mu}(t), \mu(t))$. Also note that by LLN, $n^{-1} \sum_{j=1}^{n}\left\|\log _{\mu} X_{j}\right\|_{\mu}^{2}=O_{P}(1)$. Then, with Part 4 of Theorem 6,

$$
\left\|A_{2}\right\|_{H S}^{2} \leq \text { const. }\left\{4+o_{P}(1)+O_{P}(1)\right\} \sup _{t \in \mathcal{T}} d_{\mathcal{M}}^{2}(\hat{\mu}(t), \mu(t))=O_{P}(1 / n) .
$$

Similar calculation shows that $\left\|A_{3}\right\|_{H S}^{2}=O_{P}(1 / n)$ and $\left\|A_{4}\right\|_{H S}^{2}=O_{P}\left(1 / n^{2}\right)$. Now, by Dauxois, Pousse and Romain (1982), $\left\|n^{-1} \sum\left(\log _{\mu} X_{i}\right) \otimes\left(\log _{\mu} X_{i}\right)-\mathcal{C}\right\|_{H S}^{2}=O_{P}(1 / n)$. Thus, $\|\Phi \hat{\mathcal{C}}-\mathcal{C}\|_{H S}^{2}=$ $O_{P}(1 / n)$. According to Part $1 \& 5$ of Proposition $2, \hat{\lambda}_{k}$ are also eigenvalues of $\Phi \hat{\mathcal{C}}$. The results for $\hat{\lambda}_{k}$ and $\left(J, \delta_{j}\right)$ follow from Bosq (2000). Those for $\left(\hat{J}, \hat{\delta}_{j}\right)$ are due to $\sup _{k \geq 1}\left|\hat{\lambda}_{k}-\lambda_{k}\right| \leq\left\|\hat{\mathcal{C}} \ominus_{\Phi} \mathcal{C}\right\|_{H S}$.

Proof of Theorem 8. In this proof, both $o_{P}(\cdot)$ and $O_{P}(\cdot)$ are understood to be uniform for the class $\mathcal{F}$. Let $\check{\beta}=\operatorname{Exp}_{\mu} \sum_{k=1}^{K} \hat{b}_{k} \Gamma \hat{\phi}_{k}$. Then

$$
d_{\mathcal{M}}^{2}(\hat{\beta}, \beta) \leq 2 d_{\mathcal{M}}^{2}(\hat{\beta}, \check{\beta})+2 d_{\mathcal{M}}^{2}(\check{\beta}, \beta)
$$

The first term is of order $O_{p}(1 / n)$ uniform for the class $\mathcal{F}$, according to a technique similar to the one in the proof of Lemma 10, as well as Theorem 6 (note that the results in Theorem 6 are uniform for the class $\mathcal{F}$ ). Then the convergence rate is established if one can show that

$$
d_{\mathcal{M}}^{2}(\check{\beta}, \beta)=O_{P}\left(n^{-\frac{2 \varrho-1}{2 \varrho+2 \alpha}}\right),
$$

which follows from

$$
\begin{equation*}
\left\|\sum_{k=1}^{K} \hat{b}_{k} \Gamma \hat{\phi}_{k}-\sum_{k=1}^{\infty} b_{k} \phi_{k}\right\|_{\mu}^{2}=O_{P}\left(n^{-\frac{2 \varrho-1}{2 Q+2 \alpha}}\right) \tag{18}
\end{equation*}
$$

and Proposition 4. It remains to show (18).
We first observe that because $b_{k} \leq C k^{-\beta}$,

$$
\begin{equation*}
\left\|\sum_{k=1}^{K} \hat{b}_{k} \Gamma \hat{\phi}_{k}-\sum_{k=1}^{\infty} b_{k} \phi_{k}\right\|_{\mu}^{2} \leq 2\left\|\sum_{k=1}^{K} \hat{b}_{k} \Gamma \hat{\phi}_{k}-\sum_{k=1}^{K} b_{k} \phi_{k}\right\|_{\mu}^{2}+O\left(K^{-2 \varrho+1}\right) . \tag{19}
\end{equation*}
$$

Define

$$
A_{1}=\sum_{k=1}^{K}\left(\hat{b}_{k}-b_{k}\right) \phi_{k}, \quad A_{2}=\sum_{k=1}^{K} b_{k}\left(\Gamma \hat{\phi}_{k}-\phi_{k}\right), \quad A_{3}=\sum_{k=1}^{K}\left(\hat{b}_{k}-b_{k}\right)\left(\Gamma \hat{\phi}_{k}-\phi_{k}\right) .
$$

Then

$$
\left\|\sum_{k=1}^{K} \hat{b}_{k} \Gamma \hat{\phi}_{k}-\sum_{k=1}^{K} b_{k} \phi_{k}\right\|_{\mu}^{2} \leq 2\left\|A_{1}\right\|_{\mu}^{2}+2\left\|A_{2}\right\|_{\mu}^{2}+2\left\|A_{3}\right\|_{\mu}^{2} .
$$

It is clear that the term $A_{3}$ is asymptotically dominated by $A_{1}$ and $A_{2}$. Note that the compactness of $X$ in condition C. 2 implies $\mathbb{E}\left\|\log _{\mu} X\right\|_{\mu}^{4}<\infty$. Then, by Theorem 7 , for $A_{2}$, we have the bound

$$
\left\|A_{2}\right\|_{\mu}^{2} \leq 2 \sum_{k=1}^{K} b_{k}^{2}\left\|\Gamma \hat{\phi}_{k}-\phi_{k}\right\|_{\mu}^{2}= \begin{cases}O_{P}\left(\frac{1}{n} K^{-2 \varrho+2 \alpha+3}\right) & \text { if } \alpha>\varrho-3 / 2 \\ O_{P}\left(\frac{1}{n} \log K\right) & \text { if } \alpha=\varrho-3 / 2 \\ O_{P}\left(\frac{1}{n}\right) & \text { if } \alpha<\varrho-3 / 2\end{cases}
$$

It is easy to verify that, because $K \asymp n^{1 /(2 \alpha+2 \varrho)},\left\|A_{2}\right\|_{\mu}^{2}$ is asymptotically dominated by $K^{-2 \varrho+1}$. Thus,

$$
\left\|A_{2}\right\|_{\mu}^{2}=O_{P}\left(n^{-\frac{2 \varrho-1}{2 \varrho+2 \alpha}}\right)
$$

Now we focus on the term $A_{1}$. Note that $\hat{a}_{k}=\left\langle\left\langle\hat{\chi}, \hat{\phi}_{k}\right\rangle_{\hat{\mu}}=\left\langle\left\langle\Gamma \hat{\chi}, \Gamma \hat{\phi}_{k}\right\rangle_{\mu}\right.\right.$. Then

$$
\left\|A_{1}\right\|_{\mu}^{2}=\sum_{k=1}^{K}\left(\hat{b}_{k}-b_{k}\right)^{2}=\sum_{k=1}^{K}\left(\hat{\lambda}_{k}^{-1}\left\langle\left\langle\Gamma \hat{\chi}, \Gamma \hat{\phi}_{k}\right\rangle\right\rangle_{\mu}-b_{k}\right)^{2} .
$$

With Theorem 6 and CLT, it can be shown that $\left.\left\langle\Gamma \hat{\chi}, \Gamma \hat{\phi}_{k}\right\rangle\right\rangle_{\mu}=\lambda_{k} b_{k}+O_{P}(\sqrt{1 / n})$. Note that $\sup _{k}\left|\hat{\lambda}_{k}-\lambda_{k}\right|^{2}=O_{P}(1 / n)$ which implies that $\hat{\lambda}_{k}>\lambda_{k}-O_{P}(1 / n)$ uniformly for all $k$. In conjunction with the choice of $K \asymp n^{1 /(2 \alpha+2 \varrho)}$, one can conclude that

$$
\begin{aligned}
\left\|A_{1}\right\|_{\mu}^{2} & =\sum_{k=1}^{K}\left(\frac{\lambda_{k}-\hat{\lambda}_{k}}{\hat{\lambda}_{k}} b_{k}\right)^{2}+O_{P}\left(\frac{1}{n}\right) \sum_{k=1}^{K} \hat{\lambda}_{k}^{-2} \leq O_{P}\left(\frac{1}{n}\right) \sum_{k=1}^{K} \lambda_{k}^{-2} b_{k}^{2}+O_{P}\left(\frac{1}{n}\right) \sum_{k=1}^{K} \lambda_{k}^{-2} \\
& =O_{P}\left(\frac{1}{n}\right) \sum_{k=1}^{K} \lambda_{k}^{-2}=O_{P}\left(\frac{K^{2 \alpha+1}}{n}\right)=O_{P}\left(n^{-\frac{2 \varrho-1}{2 \varrho+2 \alpha}}\right) .
\end{aligned}
$$

Thus,

$$
\left\|\sum_{k=1}^{K} \hat{b}_{k} \Gamma \hat{\phi}_{k}-\sum_{k=1}^{K} b_{k} \phi_{k}\right\|_{\mu}^{2}=O_{P}\left(n^{-\frac{2 \varrho-1}{2 \varrho+2 \alpha}}\right)
$$

Finally, (18) follows from the above equation, the equation (19) and $K^{-2 \varrho+1}=O\left(n^{-\frac{2 \varrho-1}{2 \varrho+2 \alpha}}\right)$.
Proof of Theorem 9. In this proof, both $o_{P}(\cdot)$ and $O_{P}(\cdot)$ are understood to be uniform for the class $\mathcal{G}$. First, $d_{\mathcal{M}}^{2}(\hat{\beta}, \beta) \leq 2 d_{\mathcal{M}}^{2}\left(\hat{\beta}, \operatorname{Exp}_{\mu} \Gamma\left(\hat{\mathcal{C}}^{+} \hat{\chi}\right)\right)+2 d_{\mathcal{M}}^{2}\left(\operatorname{Exp}_{\mu} \Gamma\left(\hat{\mathcal{C}}^{+} \hat{\chi}\right), \beta\right)$. The first term is of order $O_{p}(1 / n)$ uniform for the class $\mathcal{G}$, according to a technique similar to the one in the proof of Lemma 10, as well as Theorem 6 (note that the results in Theorem 6 are uniform for the class $\mathcal{G}$ ). For the second term, using Proposition 2 , one can show that $\Phi \hat{\mathcal{C}}^{+}=\left(\Phi \hat{\mathcal{C}}+\rho \Phi I_{\hat{\mu}}\right)^{-1}=\left(\Phi \hat{\mathcal{C}}+\rho I_{\mu}\right)^{-1}$, where $I_{\hat{\mu}}$ and $I_{\mu}$ denote the identity operators on $\mathscr{T}(\hat{\mu})$ and $\mathscr{T}(\mu)$, respectively. It remains to show $d_{\mathcal{M}}^{2}\left(\operatorname{Exp}_{\mu} \Gamma\left(\hat{\mathcal{C}}^{+} \hat{\chi}\right), \beta\right)=O_{P}\left(n^{-(2 \varrho-\alpha) /(2 \varrho+\alpha)}\right)$.

Define $\mathcal{C}_{\rho}^{+}=\left(\mathcal{C}+\rho I_{\mu}\right)^{-1}, \chi_{n}=n^{-1} \sum_{i=1}^{n}\left(Y_{i}-\bar{Y}\right) \log _{\mu} X_{i}$, and

$$
\begin{aligned}
& A_{n 1}=\mathcal{C}_{\rho}^{+} \chi_{n}+\mathcal{C}_{\rho}^{+}\left(\hat{\chi}-\chi_{n}\right) \equiv A_{n 11}+A_{n 12} \\
& A_{n 2}=\left(\Phi \hat{\mathcal{C}}^{+}-\mathcal{C}_{\rho}^{+}\right) \chi_{n}+\left(\Phi \hat{\mathcal{C}}^{+}-\mathcal{C}_{\rho}^{+}\right)\left(\hat{\chi}_{n}-\chi_{n}\right) \equiv A_{n 21}+A_{n 22}
\end{aligned}
$$

It is easy to see that the dominant term is $A_{n 11}$ and $A_{n 21}$ for $A_{n 1}$ and $A_{n 2}$, respectively. With $\rho=$ $n^{-\alpha /(2 \beta+\alpha)}$, it has been shown in Hall and Horowitz (2007) that $\mathbb{E}\left\|A_{n 1}-\log _{\mu} \beta\right\|_{\mu}^{2}=O\left(n^{-(2 \varrho-1) /(2 \varrho+\alpha)}\right)$. Denote $\Delta=\Phi \hat{\mathcal{C}}-\mathcal{C}_{\rho}^{+}$. Then

$$
\begin{aligned}
A_{n 21} & =\left(\Phi \hat{\mathcal{C}}^{+}-\mathcal{C}_{\rho}^{+}\right) \chi_{n}=-\left(I_{\mu}+\mathcal{C}_{\rho}^{+} \Delta\right)^{-1} \mathcal{C}_{\rho}^{+} \Delta \mathcal{C}_{\rho}^{+} \chi_{n} \\
& =-\left(I_{\mu}+\mathcal{C}_{\rho}^{+} \Delta\right)^{-1} \mathcal{C}_{\rho}^{+} \Delta \log _{\mu} \beta-\left(I_{\mu}+\mathcal{C}_{n}^{+} \Delta\right)^{-1} \mathcal{C}_{\rho}^{+} \Delta\left(\mathcal{C}_{\rho}^{+} \chi_{n}-\log _{\mu} \beta\right) \\
& \equiv A_{n 211}+A_{n 212} .
\end{aligned}
$$

By Theorem $7,\|\Delta\|_{\mu}=O_{P}(1 / n)$. Also, one can see that $\left\|\left(I_{\mu}+\mathcal{C}_{\rho}^{+} \Delta\right)^{-1}\right\|_{\mu}=O_{P}(1)$, with the assumption that $\rho^{-1} / n=o(1)$. Also, $\left\|\left(I_{\mu}+\mathcal{C}_{\rho}^{+} \Delta\right)^{-1} \mathcal{C}_{\rho}^{+} \Delta\right\|_{o p}=O_{P}\left(\rho^{-2} / n\right)$. Using the similar technique in Hall and Horowitz (2005), we can show that $\left\|\mathcal{C}_{\rho}^{+} \chi_{n}-\log _{\mu} \beta\right\|_{\mu}^{2}=O_{P}\left(n^{-(2 \varrho-1) /(2 \varrho+\alpha)}\right)$, and hence conclude that $\left\|A_{n 212}\right\|_{\mu}^{2}=O_{P}\left(n^{-(2 \varrho-1) /(2 \varrho+\alpha)}\right)$. For $A_{n 211}$,

$$
\left\|A_{n 211}\right\|_{\mu}^{2}=\left\|\left(I_{\mu}+\mathcal{C}_{n}^{+} \Delta\right)^{-1} \mathcal{C}_{n}^{+} \Delta \log _{\mu} \beta\right\|_{\mu}^{2} \leq\left\|\left(I_{\mu}+\mathcal{C}_{n}^{+} \Delta\right)^{-1}\right\|_{o p}^{2}\left\|\mathcal{C}_{n}^{+} \Delta\right\|_{o p}^{2}\left\|\log _{\mu} \beta\right\|_{\mu}^{2}=O_{P}\left(n^{-(2 \varrho-\alpha) /(2 \varrho+\alpha)}\right)
$$

Combining all results above, we deduce that $\left\|\Gamma\left(\hat{\mathcal{C}}^{+} \hat{\chi}\right)-\log _{\mu} \beta\right\|_{\mu}^{2}=O_{P}\left(n^{-(2 \varrho-\alpha) /(2 \varrho+\alpha)}\right)$ and thus

$$
d_{\mathcal{M}}^{2}\left(\operatorname{Exp}_{\mu} \Gamma\left(\hat{\mathcal{C}}^{+} \hat{\chi}\right), \beta\right)=O_{P}\left(n^{-(2 \varrho-\alpha) /(2 \varrho+\alpha)}\right)
$$

according to condition C. 2 and Proposition 4.

## APPENDIX C: ANCILLARY LEMMAS

## Lemma 10. $\sup _{t \in \mathcal{T}} n^{-1}\left\|\Delta_{t}(\hat{\mu}(t))\right\|=o_{P}(1)$, where $\Delta_{t}$ is as in (16).

Proof. With the continuity of $\mu$ and compactness of $\mathcal{T}$, the existence of local smooth orthnormal frames (e.g., Proposition 11.17 of Lee (2002)) suggests that we can find a finite open cover $\mathcal{T}_{1}, \ldots, \mathcal{T}_{m}$ for $\mathcal{T}$ such that there exists a smooth orthonormal frame $b_{j, 1}, \ldots, b_{j, d}$ for the $j$ th piece $\{\mu(t): t \in$ $\left.\operatorname{cl}\left(\mathcal{T}_{j}\right)\right\}$ of $\mu$, where $\operatorname{cl}(A)$ denotes topological closure of a set $A$. For fixed $t \in \mathcal{T}_{j}$, by mean value theorem, it can be shown that

$$
\begin{equation*}
\Delta_{t}(\hat{\mu}(t)) U=\sum_{r=1}^{d} \sum_{i=1}^{n}\left(\mathcal{P}_{\gamma_{t, \hat{\mu}(t)}\left(\theta_{t}^{r, j}\right), \mu(t)} \nabla_{U} W_{t, i}^{r, j}\left(\gamma_{t, \hat{\mu}(t)}\left(\theta_{t}^{r, j}\right)\right)-\nabla_{U} W_{t, i}^{r, j}(\mu(t))\right) \tag{20}
\end{equation*}
$$

for $\theta_{t}^{r, j} \in[0,1]$ and $W_{t, i}^{r, j}=\left\langle V_{t, i}, e_{t}^{r, j}\right\rangle e_{t}^{r, j}$, where $e_{t}^{1, j}, \ldots, e_{t}^{d, j}$ is the orthonormal frame extended by parallel transport of $b_{j, 1}(\mu(t)), \ldots, b_{j, d}(\mu(t))$ along minimizing geodesic.

Take $\epsilon=\epsilon_{n} \downarrow 0$ as $n \rightarrow \infty$. For each $j$, by the same argument of Lemma 3 of Kendall and Le (2011), together with continuity of $\mu$ and the continuity of the frame $b_{j, 1}, \ldots, b_{j, d}$, we can choose a continuous positive $\rho_{t}^{j}$ such that, $\hat{\mu}(t) \in B\left(\mu(t), \rho_{t}^{j}\right)$ and for $p \in B\left(\mu(t), \rho_{t}^{j}\right)$ where $B(q, \rho)$ denotes the ball on $\mathcal{M}$ centered at $q$ with radius $\rho$,

$$
\begin{aligned}
\left\|\mathcal{P}_{p, \mu(t)} \nabla W_{t, i}^{r, j}(p)-\nabla W_{t, i}^{r, j}(\mu(t))\right\| & \leq\left(1+2 \epsilon \rho_{t}^{j}\right) \sup _{q \in B\left(\mu(t), \rho_{t}^{j}\right)}\left\|\mathcal{P}_{q, \mu(t)} \nabla V_{t, i}(q)-\nabla V_{t, i}(\mu(t))\right\| \\
& +2 \epsilon\left(\left\|V_{t, i}(\mu(t))\right\|+\rho_{t}^{j}\left\|\nabla V_{t, i}(\mu(t))\right\|\right) .
\end{aligned}
$$

In the above, $p$ plays a role of $\gamma_{t, \hat{\mu}(t)}\left(\theta_{t}^{r, j}\right)$ in (20). Let $\rho^{j}=\max \left\{\rho_{t}: t \in \operatorname{cl}\left(\mathcal{T}_{j}\right)\right\}$ and $\rho_{\max }=\max _{j} \rho^{j}$. We then have

$$
\begin{align*}
\sup _{t \in \mathcal{T}}\left\|\Delta_{t}(\hat{\mu}(t))\right\| & \leq \max _{j} \sup _{t \in \mathcal{T}_{j}}\left\|\Delta_{t}(\hat{\mu}(t))\right\| \\
& =\sum_{i=1}^{n} \max _{j} \sup _{t \in \mathcal{T}_{j}}\left\|\mathcal{P}_{\gamma_{t, \hat{\mu}(t)}\left(\theta_{t}^{r, j}\right), \mu(t)} \nabla_{U} W_{r, i}^{t, j}\left(\gamma_{t, \hat{\mu}(t)}\left(\theta_{t}^{r, j}\right)\right)-\nabla_{U} W_{r, i}^{t, j}(\mu(t))\right\| \\
& \leq\left(1+2 \epsilon \rho_{\max }\right) m \sum_{i=1}^{n} \sup _{t \in \mathcal{T}} \sup _{q \in B\left(\mu(t), \rho_{\max }\right)}\left\|\mathcal{P}_{q, \mu(t)} \nabla V_{t, i}(q)-\nabla V_{t, i}(\mu(t))\right\|  \tag{21}\\
& +2 \epsilon \sum_{i=1}^{n} \sup _{t \in \mathcal{T}}\left\|V_{t, i}(\mu(t))\right\|+2 \epsilon \rho_{\max } \sum_{i=1}^{n} \sup _{t \in \mathcal{T}}\left\|\nabla V_{t, i}(\mu(t))\right\| . \tag{22}
\end{align*}
$$

For (21), the Lipschitz condition of B. 7 and smoothness of $d_{\mathcal{M}}$ imply that

$$
\lim _{\rho \downarrow 0} \mathbb{E} \sup _{q \in B(\mu(t), \rho)}\left\|\mathcal{P}_{q, \mu(t)} \nabla V_{t, i}(q)-\nabla V_{t, i}(\mu(t))\right\|=\lim _{\rho \downarrow 0} \mathbb{E} \sup _{q \in B(\mu(t), \rho)}\left\|\mathcal{P}_{q, \mu(t)} H_{t}(q)-H_{t}(\mu(t))\right\|=0 .
$$

As $\sup _{t \in \mathcal{T}} d_{\mathcal{M}}(\hat{\mu}(t), \mu(t))=o_{\text {a.s. }}(1), \rho_{\max }$ could be chosen so that $\rho_{\max } \downarrow 0$ as $n \rightarrow \infty$. Thus, with probability tending to one, by Markov inequality, we deduce that

$$
\begin{equation*}
\left(1+2 \epsilon \rho_{\max }\right) m \frac{1}{n} \sum_{i=1}^{n} \sup _{t \in \mathcal{T}_{t}} \sup _{q \in B\left(\mu(t), \rho_{\max }\right)}\left\|\mathcal{P}_{q, \mu(t)} \nabla V_{t, i}(q)-\nabla V_{t, i}(\mu(t))\right\|=o_{P}(1) . \tag{23}
\end{equation*}
$$

For the first term in (22), LLN shows that

$$
\frac{1}{n} \sum_{i=1}^{n} \sup _{t \in \mathcal{T}}\left\|V_{t, i}(\mu(t))\right\| \xrightarrow{P} \mathbb{E} \sup _{t \in \mathcal{T}}\left\|V_{t, i}(\mu(t))\right\|=\mathbb{E} \sup _{t \in \mathcal{T}} d_{\mathcal{M}}(X(t), \mu(t))<\infty,
$$

or

$$
\begin{equation*}
\frac{1}{n} \sum_{i=1}^{n} \sup _{t \in \mathcal{T}}\left\|V_{t, i}(\mu(t))\right\|=O_{P}(1) . \tag{24}
\end{equation*}
$$

For the second term in (22), the compactness of $\mathcal{T}$, the Lipschitz condition of B. 7 and smoothness of $d_{\mathcal{M}}$ also imply that $\mathbb{E} \sup _{t \in \mathcal{T}}\left\|\nabla V_{t, i}(\mu(t))\right\|=\mathbb{E} \sup _{t \epsilon \mathcal{T}}\left\|H_{t}(\mu(t))\right\|<\infty$. Consequently, by LLN,

$$
\begin{equation*}
\frac{1}{n} \sum_{i=1}^{n} \sup _{t \in \mathcal{T}}\left\|\nabla V_{t, i}(\mu(t))\right\|=O_{P}(1) . \tag{25}
\end{equation*}
$$

Combining (23), (24) and (25), with $\epsilon=\epsilon_{n} \downarrow 0$, one concludes that $\sup _{t \epsilon \mathcal{T}} n^{-1}\left\|\Delta_{t}(p)\right\|=o_{P}(1)$.
Lemma 11. Suppose conditions A. 1 and B.1-B. 3 hold. For any compact subset $\mathcal{K} \subset \mathcal{M}$, one has

$$
\sup _{p \in \mathcal{K}} \sup _{t \in \mathcal{T}}\left|F_{n}(p, t)-F(p, t)\right|=o_{\text {a.s. }}(1)
$$

Proof. By applying the uniform SLLN to $n^{-1} \sum_{i=1}^{n} d_{\mathcal{M}}\left(X_{i}(t), p_{0}\right)$, for a given $p_{0} \in \mathcal{K}$,

$$
\begin{aligned}
\sup _{p \in \mathcal{K}} \sup _{t \in \mathcal{T}} \frac{1}{n} \sum_{i=1}^{n} d_{\mathcal{M}}\left(X_{i}(t), p\right) & \leq \sup _{t \in \mathcal{T}} \frac{1}{n} \sum_{i=1}^{n} d_{\mathcal{M}}\left(X_{i}(t), p_{0}\right)+\sup _{p \in \mathcal{K}} d_{\mathcal{M}}\left(p_{0}, p\right) \\
& \leq \sup _{t \in \mathcal{T}} \mathbb{E} d_{\mathcal{M}}\left(X(t), p_{0}\right)+\operatorname{diam}(\mathcal{K})+o_{\text {a.s. }}(1) .
\end{aligned}
$$

Therefore, there exists a set $\Omega_{1} \subset \Omega$ such that $\operatorname{Pr}\left(\Omega_{1}\right)=1, N_{1}(\omega)<\infty$ and for all $n \geq N_{1}(\omega)$,

$$
\sup _{p \in \mathcal{K}} \sup _{t \in \mathcal{T}} \frac{1}{n} \sum_{i=1}^{n} d_{\mathcal{M}}\left(X_{i}(t), p\right) \leq \sup _{t \in \mathcal{T}} \mathbb{E} d_{\mathcal{M}}\left(X(t), p_{0}\right)+\operatorname{diam}(\mathcal{K})+1:=c_{1}<\infty,
$$

since $\sup _{t \in \mathcal{T}} \mathbb{E} d_{\mathcal{M}}\left(X(t), p_{0}\right)<\infty$ by condition B.3. Fix $\epsilon>0$. By the inequality $\left|d_{\mathcal{M}}^{2}(x, p)-d_{\mathcal{M}}^{2}(x, q)\right| \leq$ $\left\{d_{\mathcal{M}}(x, p)+d_{\mathcal{M}}(x, q)\right\} d_{\mathcal{M}}(p, q)$, for all $n \geq N_{1}(\omega)$ and $\omega \in \Omega_{1}$,

$$
\sup _{p, q \in \mathcal{K}: d_{\mathcal{M}}(p, q)<\delta_{1}} \sup _{t \in \mathcal{T}}\left|F_{n, \omega}(p, t)-F_{n, \omega}(q, t)\right| \leq 2 c_{1} \delta_{1}=\epsilon / 3
$$

with $\delta_{1}:=\epsilon /\left(6 c_{1}\right)$. Now, let $\delta_{2}>0$ be chosen such that $\sup _{t \in \mathcal{T}}|F(p, t)-F(q, t)|<\epsilon / 3$ if $p, q \in \mathcal{K}$ and $d_{\mathcal{M}}(p, q)<\delta_{2}$. Suppose $\left\{p_{1}, \ldots, p_{r}\right\} \subset \mathcal{K}$ is a $\delta$-net in $\mathcal{K}$ with $\delta:=\min \left\{\delta_{1}, \delta_{2}\right\}$. Applying uniform SLLN again, there exists a set $\Omega_{2}$ such that $\operatorname{Pr}\left(\Omega_{2}\right)=1, N_{2}(\omega)<\infty$ for all $\omega \in \Omega_{2}$, and

$$
\max _{j=1, \ldots, r} \sup _{t \epsilon \mathcal{T}}\left|F_{n, \omega}\left(p_{j}, t\right)-F\left(p_{j}, t\right)\right|<\epsilon / 3
$$

for all $n \geq N_{2}(\omega)$ with $\omega \in \Omega_{2}$. Then, for all $\omega \in \Omega_{1} \cap \Omega_{2}$, for all $n \geq \max \left\{N_{1}(\omega), N_{2}(\omega)\right\}$, we have

$$
\begin{aligned}
& \sup _{p \in \mathcal{K}} \sup _{t \in \mathcal{T}}\left|F_{n, \omega}(p, t)-F(p, t)\right| \\
& \leq \sup _{p \in \mathcal{K}} \sup _{t \in \mathcal{T}}\left|F_{n, \omega}(p)-F_{n, \omega}\left(u_{p}\right)\right|+\sup _{p \in \mathcal{K}} \sup _{t \in \mathcal{T}}\left|F_{n, \omega}\left(u_{p}, t\right)-F\left(u_{p}, t\right)\right|+\sup _{p \in \mathcal{K}} \sup _{t \in \mathcal{T}}\left|F\left(u_{p}, t\right)-F(p, t)\right| \\
& <\epsilon / 3+\epsilon / 3+\epsilon / 3=\epsilon,
\end{aligned}
$$

and this concludes the proof.
Lemma 12. Assume conditions A.1 and B.1-B.5 hold. Given any $\epsilon>0$, there exists $\Omega^{\prime} \subset \Omega$ such that $\operatorname{Pr}\left(\Omega^{\prime}\right)=1$ and for all $\omega \in \Omega^{\prime}, N(\omega)<\infty$ and for all $n \geq N(\omega), \sup _{t \in \mathcal{T}} d_{\mathcal{M}}\left(\hat{\mu}_{\omega}(t), \mu(t)\right)<\epsilon$.

Proof. Let $c(t)=F(\mu(t), t)=\min \{F(p, t): p \in \mathcal{M}\}$ and $\mathcal{N}(t):=\left\{p: d_{\mathcal{M}}(p, \mu(t)) \geq \epsilon\right\}$. It is sufficient to show that there exists $\delta>0$ and $N(\omega)<\infty$ for all $\omega \in \Omega^{\prime}$, such that for all $n \geq N(\omega)$,

$$
\sup _{t \in \mathcal{T}}\left\{F_{n, \omega}(\mu(t), t)-c(t)\right\} \leq \delta / 2 \quad \text { and } \quad \inf _{t \in \mathcal{T}}\left\{\inf _{p \in \mathcal{N}(t)} F_{n, \omega}(p, t)-c(t)\right\} \geq \delta .
$$

This is because the above two inequalities suggest that for all $t \in \mathcal{T}, \inf \left\{F_{n, \omega}(p, t): p \in \mathcal{M}\right\}$ is not attained at $p$ with $d_{\mathcal{M}}(p, \mu(t)) \geq \epsilon$, and hence $\sup _{t \epsilon \mathcal{T}} d_{\mathcal{M}}\left(\hat{\mu}_{\omega}(t), \mu(t)\right)<\epsilon$.

Let $\mathcal{U}=\{\mu(t): t \in \mathcal{T}\}$. We first show that there exists a compact set $\mathcal{A} \supset \mathcal{U}$ and $N_{1}(\omega)<\infty$ for some $\Omega_{1} \subset \Omega$ such that $\operatorname{Pr}\left(\Omega_{1}\right)=1$, and both $F(p, t)$ and $F_{n, \omega}(p, t)$ are greater than $c(t)+1$ for all $p \in \mathcal{M} \backslash \mathcal{A}, t \in \mathcal{T}$ and $n \geq N_{1}(\omega)$. This is trivially true when $\mathcal{M}$ is compact, by taking $\mathcal{A}=\mathcal{M}$. Now assume $\mathcal{M}$ is noncompact. By the inequality $d_{\mathcal{M}}(x, q) \geq\left|d_{\mathcal{M}}(q, y)-d_{\mathcal{M}}(y, x)\right|$, one has

$$
\mathbb{E} d_{\mathcal{M}}^{2}(X(t), q) \geq \mathbb{E}\left\{d_{\mathcal{M}}^{2}(q, \mu(t))+d_{\mathcal{M}}^{2}(X(t), \mu(t))-2 d_{\mathcal{M}}(q, \mu(t)) d_{\mathcal{M}}(X(t), \mu(t))\right\}
$$

and by Cauchy-Schwarz inequality,

$$
F(q, t) \geq d_{\mathcal{M}}^{2}(q, \mu(t))+F(\mu(t), t)-2 d_{\mathcal{M}}(q, \mu(t))\{F(\mu(t), t)\}^{1 / 2} .
$$

Similarly,

$$
F_{n, \omega}(q, t) \geq d_{\mathcal{M}}^{2}(q, \mu(t))+F_{n, \omega}(\mu(t), t)-2 d_{\mathcal{M}}(q, \mu(t))\left\{F_{n, \omega}(\mu(t), t)\right\}^{1 / 2}
$$

Now, we take $q$ at a sufficiently large distance $\Delta$ from $\mathcal{U}$ such that $F(q, t)>c(t)+1$ on $\mathcal{M} \backslash \mathcal{A}$ for all $t$, where $\mathcal{A}:=\left\{q: d_{\mathcal{M}}(q, \mathcal{U}) \leq \Delta\right\}$ (Heine-Borel property yields compactness of $\mathcal{A}$, since it is bounded and closed). Since $F_{n, \omega}(\mu(t), t)$ converges to $F(\mu(t), t)$ uniformly on $\mathcal{T}$ a.s. by Lemma 11, we can find a set $\Omega_{1} \subset \Omega$ such that $\operatorname{Pr}\left(\Omega_{1}\right)=1$ and $N_{1}(\omega)<\infty$ for $\omega \in \Omega_{1}$, and $F_{n, \omega}(q, t)>c(t)+1$ on $\mathcal{M} \backslash \mathcal{A}$ for all $t$ and $n \geq N_{1}(\omega)$.

Finally, let $\mathcal{A}_{\epsilon}(t):=\left\{p \in \mathcal{A}: d_{\mathcal{M}}(p, \mu(t)) \geq \epsilon\right\}$ and $c_{\epsilon}(t):=\min \left\{F(p, t): p \in \mathcal{A}_{\epsilon}\right\}$. Then $\mathcal{A}_{\epsilon}(t)$ is compact and by condition B.5, $\inf _{t}\left\{c_{\epsilon}(t)-c(t)\right\}>2 \delta>0$ for some constant $\delta$. By Lemma 11, one can find a set $\Omega_{2} \subset \Omega$ with $\operatorname{Pr}\left(\Omega_{2}\right)=1$ and $N_{2}(\omega)<\infty$ for $\omega \in \Omega_{2}$, such that for all $n \geq N_{2}(\omega)$, (i) $\sup _{t}\left\{F_{n, \omega}(\mu(t), t)-c(t)\right\} \leq \delta / 2$ and $(\mathrm{ii}) \inf _{t} \inf _{p \in \mathcal{A}_{\epsilon}(t)}\left\{F_{n, \omega}(p, t)-c(t)\right\}>\delta$. Since $\sup _{t}\left\{F_{n, \omega}(p, t)-\right.$ $c(t)\}>1$ on $\mathcal{M} \backslash \mathcal{A}$ for all $n \geq N_{1}(\omega)$ with $\omega \in \Omega_{1}$, we conclude that $\inf _{t}\left\{F_{n, \omega}(p, t)-c(t)\right\}>\min \{\delta, 1\}$ for all $p \in \mathcal{A}_{\epsilon} \cup(\mathcal{M} \backslash \mathcal{A})$ if $n \geq \max \left\{N_{1}(\omega), N_{2}(\omega)\right\}$ for $\omega \in \Omega_{1} \cap \Omega_{2}$. The proof is completed by noting that $\Omega_{1} \cap \Omega_{2}$ can serve the $\Omega^{\prime}$ we are looking for.

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[^0]:    AMS 2000 subject classifications: 62G05, 62J05
    Keywords and phrases: Functional principal component, Functional linear regression, Intrinsic Riemannian Karhunen-Loève expansion, Parallel transport, Tensor Hilbert space

[^1]:    ${ }^{1}$ Implementation of iRFPCA on these manifolds can be found on Github: https://github.com/linulysses/iRFDA.

