

# From microfacets to participating media: A unified theory of light transport with stochastic geometry – Supplemental

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## 1 DERIVING RECURSIVE ENSEMBLE AVERAGE LIGHT TRANSPORT

The step from Eq. (12) to Eq. (14) in the main document is not immediately obvious. It does not require any deep insights, but a thorough reproduction of the intermediate steps, while lengthy, provides extra intuition for our method. We provide this here in a slightly more general form that should apply to any quantity that can be estimated with a recursive Monte Carlo estimator.

### 1.1 Expectations over realizations of a Gaussian Process

We will often want to compute expectations “over realizations of a Gaussian Process”, that is, functional integrals of the form

$$\langle \mathcal{L}f \rangle_{\text{GP}(\mu, k)} = \int_{\text{GP}(\mu, k)} \mathcal{L}f \, d\gamma_{\mu, k}(f), \quad (1)$$

where  $f$  is a realization of the Gaussian process  $\text{GP}(\mu, k)$ ,  $\mathcal{L}$  an operator acting on  $f$ , and  $\gamma_{\mu, k}(f)$  the classical Wiener measure of  $f$  with respect to  $\text{GP}(\mu, k)$ , i.e. the probability density of sampling  $f \sim \text{GP}(\mu, k)$ . If the operator  $\mathcal{L}$  is linear in  $f$ , we can resolve this in the following straightforward fashion

$$\int_{\text{GP}(\mu, k)} \mathcal{L}f \, d\gamma_{\mu, k}(f) = \mathcal{L} \int_{\text{GP}(\mu, k)} f \, d\gamma_{\mu, k}(f) = \mathcal{L}\mu, \quad (2)$$

but many operators, in particular light transport, which we study in this paper, are not linear in  $f$ , and the above simplification does not hold. In the following, we will drop the mean  $\mu$  and covariance kernel  $k$  for notational convenience. A rigorous treatment of functional integrals of this form is out of the scope of this work. Instead, we can provide intuition for our results by restricting ourselves to Gaussian processes over a finite index set  $X$  with  $|X| = n$ , and discrete operators  $\mathcal{L}^n$ . In that case, the Gaussian process is simply a multivariate normal distribution in  $n$  dimensions, and we can write

$$\begin{aligned} \int_{\text{GP}^X} \mathcal{L}^n f \, d\gamma(f) &= \\ \int_{\mathbb{R}} \cdots \int_{\mathbb{R}} \mathcal{L}^n [f_1, \dots, f_n]^\top \, d\gamma^n([f_1, \dots, f_n]^\top) &= \\ \int_{\mathbb{R}^n} \mathcal{L}^n F \, d\gamma^n(F), \end{aligned} \quad (3)$$

where  $\gamma^n$  is  $n$  dimensional Gaussian measure over  $X$  with the same mean and covariance as the given Gaussian process. Informally, we

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write

$$\int_{\text{GP}} \mathcal{L}f \, d\gamma(f) := \lim_{n \rightarrow \infty} \int_{\mathbb{R}^n} \mathcal{L}^n F \, d\gamma^n(F). \quad (4)$$

Of course, there is no immediate reason to believe that this limit is well defined in any meaningful sense. That said, this technique, also known as “time slicing” [Grosche and Steiner 1998], is commonly used to make sense of functional integrals, and for the purposes of our work and the domains we consider, the time slicing view is sufficient.

In particular, it lets us treat Gaussian processes like we treat finite-dimensional distributions in many situations. For example, we can write

$$\begin{aligned} \int_{\text{GP}} \mathcal{L}f \, d\gamma(f) &= \int_{\text{GP}} \mathcal{L}[f_A, f_{A^c}] \, d\gamma([f_A, f_{A^c}]) \\ &= \int_{\text{GP}_A} \int_{\text{GP}_{A^c|f_A}} \mathcal{L}[f_A, f_{A^c}] \, d\gamma_{A^c}(f_{A^c} | f_A) \, d\gamma_A(f_A) \\ &= \int_{\text{GP}_A} \int_{\text{GP}|f_A} \mathcal{L}f \, d\gamma(f | f_A) \, d\gamma_A(f_A), \end{aligned} \quad (5)$$

where  $A$  is a subdomain of the domain of GP,  $A^c$  its complement,  $f_A$  and  $f_{A^c}$  realizations of GP over  $A$  and  $A^c$  respectively and  $[f_A, f_{A^c}]$  the “concatenation” of the individual realizations. The first equality simply decomposes  $f$  into  $f_A$  and  $f_{A^c}$ . Going to the second line, we apply the law of conditional expectations. That is, we can sample  $f$  “all at once” or first sample a part of  $f$  ( $f_A$ ) and then the rest ( $f_{A^c}$ ) conditioned on the part that we sampled first. The last line simply says that we don’t have to restrict the inner process to  $A^c$  and can instead “resample” over the whole domain. For the part of the domain covered by the “conditioning region”  $A$ , we will simply get back the same values that we conditioned on.

**1.1.1 Conditioned Expectations.** Additionally, we can write conditional expectations as usual:

$$\langle \mathcal{L}f \rangle_{\text{GP}|f(A)=\mathbf{a}} = \frac{\langle \mathcal{L}[f_{A^c}, \mathbf{a}] \rangle_{\text{GP}_{A^c}}}{\gamma_A(\mathbf{a})}. \quad (6)$$

**1.1.2 Expectations of delta functions.** We can write an expectation containing a delta function over some subset of the domain  $A$  as

$$\begin{aligned} \langle \delta(f(A) - \mathbf{a}) \mathcal{L}f \rangle_{\text{GP}} &= \int_{\text{GP}} \delta(f(A) - \mathbf{a}) \mathcal{L}f \, d\gamma(f) \\ &= \int_{\text{GP}_A} \delta(f_A - \mathbf{a}) \int_{\text{GP}|f_A} \mathcal{L}f \, d\gamma(f | f_A) \, d\gamma_A(f_A) \\ &= \gamma_A(\mathbf{a}) \langle \mathcal{L}f \rangle_{\text{GP}|\mathbf{a}} \end{aligned} \quad (7)$$

In the first step, we simply expand the ensemble average. Next, we apply the decomposition in Eq. (5) and pull out  $\delta(f(A) - \mathbf{a})$  from the

inner integral. We can then apply the definition of the delta function and drop the outer integral, replacing the integration variable  $f_A$  with  $\mathbf{a}$  and evaluating the measure  $\gamma_A(f_A)$  at  $\mathbf{a}$ .

**1.1.3 Expectations of indicator functions.** We can write an expectation containing an indicator function over some subset of the domain  $A$  as

$$\begin{aligned} \langle \mathbf{I}(f(A) > 0) \mathcal{L}f \rangle_{\text{GP}} &= \int_{\text{GP}} \mathbf{I}(f(A) > 0) \mathcal{L}f \, d\gamma(f) \\ &= \int_{\text{GP}_A} \mathbf{I}(f(A) > 0) \int_{\text{GP}_{A^c} | f_A} \mathcal{L}f \, d\gamma(f | f_A) \, d\gamma_A(f_A) \\ &= \int_{\text{GP}_A^+} \int_{\text{GP} | f_A} \mathcal{L}f \, d\gamma(f | f_A) \, d\gamma_A(f_A) \\ &= \langle \langle \mathcal{L}f \rangle_{\text{GP} | f_A} \rangle_{\text{GP}_A^+}, \end{aligned} \quad (8)$$

where  $\text{GP}_A^+$  is the process restricted to all positive realizations over the index set  $A$ . Note that  $A$  can be a manifold in the domain of the GP (e.g. a 1D line segment in a 3D domain).

## 1.2 Application to light transport

With all of the puzzle pieces in place, we can apply them to Eq. (12) from the main paper, letting  $\mathcal{L}f := L_i^f(\mathbf{x}, \boldsymbol{\omega})$ , we are particularly interested in the inner integral

$$\int_0^\infty \iint \rho(\mathbf{x}_t) \underbrace{\int_{\text{GP} | \zeta} \delta^f(\mathbf{x}_t, \mathbf{n}) \mathbf{I}^f(0, t) L_i^f(\mathbf{x}_t, \boldsymbol{\omega}_t) \, d\gamma_\zeta(f)}_{\langle \delta^f(\mathbf{x}_t, \mathbf{n}) \mathbf{I}^f(0, t) L_i^f(\mathbf{x}_t, \boldsymbol{\omega}_t) \rangle_\zeta} \, d\boldsymbol{\omega}_t \, d\mathbf{n} \, dt, \quad (9)$$

and pull it out for more compact derivations, treating  $\mathbf{n}$  and  $t$  as free variables. We first apply Eq. (7)

$$\begin{aligned} &\langle \delta(f(\mathbf{x}_t) - 0) \cdot \delta(\bar{\nabla}f(\mathbf{x}_t) - \mathbf{n}) \cdot \mathbf{I}^f(0, t) \cdot L_i^f(\mathbf{x}_t, \boldsymbol{\omega}_t) \rangle_\zeta \\ &= \gamma_{\mathbf{x}_t | \zeta}(0, \mathbf{n}) \langle \mathbf{I}^f(0, t) \cdot L_i^f(\mathbf{x}_t, \boldsymbol{\omega}_t) \rangle_{\zeta \wedge f(\mathbf{x}_t) = 0 \wedge \bar{\nabla}f(\mathbf{x}_t) = \mathbf{n}}, \end{aligned} \quad (10)$$

and then Eq. (8) and arrive at

$$\gamma_{\mathbf{x}_t | \zeta}(0, \mathbf{n}) \langle \langle L_i^f(\mathbf{x}_t, \boldsymbol{\omega}_t) \rangle_{f(0, \mathbf{x}_t)} \rangle_{\text{GP}_{(0, \mathbf{x}_t)}^+ | \zeta \wedge f(\mathbf{x}_t) = 0 \wedge \bar{\nabla}f(\mathbf{x}_t) = \mathbf{n}}. \quad (11)$$

Since  $f(0, \mathbf{x}_t)$  is already conditioned on  $\zeta \wedge f(\mathbf{x}_t) = 0 \wedge \bar{\nabla}f(\mathbf{x}_t) = \mathbf{n}$ , we can add these to the inner condition without changing the expected value.

$$\gamma_{\mathbf{x}_t | \zeta}(0, \mathbf{n}) \langle \langle L_i^f(\mathbf{x}_t, \boldsymbol{\omega}_t) \rangle_{\zeta \wedge \zeta'_\delta \wedge f(0, \mathbf{x}_t)} \rangle_{\text{GP}_{(0, \mathbf{x}_t)}^+ | \zeta \wedge \zeta'_\delta}, \quad (12)$$

where  $\zeta'_\delta = f(\mathbf{x}_t) = 0 \wedge \bar{\nabla}f(\mathbf{x}_t) = \mathbf{n}$ . Plugging this back into Eq. (9) and expanding the outer ensemble average into its integral form, we get Eq. (14) from the main text.

## 1.3 Deriving the GPIS density in the Renewal and Renewal+ models

In Section 5 of the main document, we use the *GPIS density* to make connections to microfacet and participating media theory. We briefly want to give the full derivation that connects Eq. (14) and Eq. (17).

Recall that for the Renewal and Renewal+ models, we only condition on values at path vertices, not values along path segments.

That means that  $\zeta'_{R/R^+}$  does not contain  $f(0, \mathbf{x}_t)$  and hence does not depend on the integration variable in the innermost integral. This allows us to pull the recursive ensemble average out of the inner integral in Eq. (14) of the main text as

$$\begin{aligned} &\langle L_i(\mathbf{x}^u, \boldsymbol{\omega}) \rangle_\zeta \\ &\approx \int_0^\infty \iint \rho(\mathbf{x}_t) \gamma_{\mathbf{x}_t}(0, \mathbf{n} | \zeta) \\ &\quad \int_{\text{GP}_{(\mathbf{x}, \mathbf{x}_t)}^+ | \zeta} \langle L_i(\mathbf{x}_t, \boldsymbol{\omega}_t) \rangle_{\zeta \wedge \zeta'_{R/R^+}} \, d\gamma(f(\mathbf{x}, \mathbf{x}_t) | \zeta \wedge \zeta_\delta) \, d\boldsymbol{\omega}_t \, d\mathbf{n} \, dt \end{aligned} \quad (13)$$

$$\begin{aligned} &= \int_0^\infty \iint \rho(\mathbf{x}_t) \gamma_{\mathbf{x}_t}(0, \mathbf{n} | \zeta) \\ &\quad \underbrace{\int_{\text{GP}_{(\mathbf{x}, \mathbf{x}_t)}^+ | \zeta} d\gamma(f(\mathbf{x}, \mathbf{x}_t) | \zeta \wedge \zeta_\delta)}_{\text{T}(\mathbf{x}_t | \zeta)} \langle L_i(\mathbf{x}_t, \boldsymbol{\omega}_t) \rangle_{\zeta \wedge \zeta'_{R/R^+}} \, d\boldsymbol{\omega}_t \, d\mathbf{n} \, dt. \end{aligned} \quad (14)$$

Replacing the now “empty” inner integral with  $\text{T}(\mathbf{x}_t | \zeta)$  gives us

$$\begin{aligned} &\langle L_i(\mathbf{x}^u, \boldsymbol{\omega}) \rangle_\zeta \\ &\approx \int_0^\infty \iint \rho(\mathbf{x}_t) \gamma_{\mathbf{x}_t}(0, \mathbf{n} | \zeta) \text{T}(\mathbf{x}_t | \zeta) \langle L_i(\mathbf{x}_t, \boldsymbol{\omega}_t) \rangle_{\zeta \wedge \zeta'_{R/R^+}} \, d\boldsymbol{\omega}_t \, d\mathbf{n} \, dt. \end{aligned} \quad (15)$$

And finally, defining  $\Gamma(t, \mathbf{n} | \zeta) := \gamma_{\mathbf{x}_t}(0, \mathbf{n} | \zeta) \text{T}(\mathbf{x}_t | \zeta)$ , we get Eq. (17) in the main text.

## 2 GAUSSIAN PROCESS DETAILS

### 2.1 Kernel Functions

We give analytic forms of a set of kernels and their spectral densities. Note that our method is not restricted to only these kernels since we support any *smooth* kernel. We pick these kernels in particular because they cover a range of different fundamental properties, such as non-locality and non-positiveness, which have the potential to strongly affect the results of our method. We give their most simple forms and denote the common parameters standard deviation and length scale with  $\sigma$  and  $l$ , respectively. Deriving the spectral density, and especially sampling from it, is often non-trivial and depends on the number of dimensions. We give spectral densities for  $d = 3$  where we can, and  $d = 2$  otherwise. While theoretically, one does not have to sample basis functions proportional to the spectral density (one could choose a set using some proposal distribution and re-weight, like in any other Monte Carlo estimator), sampling proportional to the spectral density drastically reduces the number of basis functions required. For a visual overview of the listed kernels, see Fig. 3.

*Squared Exponential.* This is probably the most widely used kernel in Gaussian process literature.

$$k^{\text{SE}}(x, y) = \sigma^2 \exp \left( -\frac{\|x - y\|^2}{l^2} \right) \quad (16)$$

$$p^{\text{SE}}(\omega) = \frac{\sigma^2 e^{-\frac{1}{2} l^2 \omega^2}}{\sqrt{\frac{1}{l^2}}} \quad (17)$$

We use this as the “default” kernel for our method. It produces smooth realizations and computations with it are very numerically

stable. It also induces the most “exponential” first-passage times and is a great choice when the aim is to represent close-to-classical participating media.

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**ALGORITHM 1:** Sample  $p^{\text{SE}}$ 


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 $\xi \sim \text{U}(0, 1)$ 
 $r \leftarrow \sqrt{2 - \log(\xi)}$ 
 $\phi \sim \text{U}(0, 2\pi)$ 
return  $[\sin(\phi), \cos(\phi)]^\top \cdot r$ 
    
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*Rational Quadratic.* The rational quadratic kernel is a superposition of infinitely many squared exponential kernels with different length scales. The weighting of these length scales is determined by the additional parameter  $a$ .

$$k_a^{\text{RQ}}(x, y) = \sigma^2 \left( 1 + \frac{\|x - y\|^2}{2al^2} \right)^{-a} \quad (18)$$

$$p_a^{\text{RQ}}(\omega) = \frac{2^{\frac{5}{4} - \frac{a}{2}} \sigma^2 \left( \frac{1}{al^2} \right)^{-\frac{a}{2} - \frac{1}{4}} |\omega|^{a - \frac{1}{2}} K_{\frac{1}{2} - a} \left( \frac{\sqrt{2}|\omega|}{\sqrt{al^2}} \right)}{\Gamma(a)} \quad (19)$$

In particular,  $\lim_{a \rightarrow \infty} k_a^{\text{RQ}}(x, y) = k^{\text{SE}}(x, y)$ . For small  $a$ , the realizations of the rational quadratic kernel have “fractal” properties.

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**ALGORITHM 2:** Sample  $p_a^{\text{RQ}}$ 


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 $\tau \sim \Gamma(a, l^{-2})$ 
return Sample  $p^{\text{SE}}$  with  $l = \tau^{-2}$ 
    
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*Periodic.* Periodic kernels produce periodic realizations. This strongly affects the light transport in the scene and is a very challenging case for our algorithm. Any limit to the memory will produce drastically different results. Additionally, this shows that the covariance matrix is not sparse in general.

$$k_\lambda^{\text{Per}}(x, y) = \sigma^2 \exp \left( -\frac{2 \sin^2(\pi \|x - y\| \lambda)}{l^2} \right) \quad (20)$$

$$(21)$$

*Locally Periodic.* To ease some of the difficulties that come with the periodic kernel, while still preserving its interesting properties locally, we make use of kernel composition to construct a “locally periodic” kernel.

$$k^{\text{LocPer}}(x, y) = k_\lambda^{\text{Per}}(x, y) \cdot k^{\text{SE}}(x, y) \quad (22)$$

$$p^{\text{LocPer}}(\omega) = (p_\lambda^{\text{Per}} * p^{\text{SE}})(\omega), \quad (23)$$

where  $*$  denotes convolution. This kernel is “locally periodic” in the sense that in realizations new each “repetition” is allowed to deviate slightly from the previous one. The length scale of the squared exponential kernel controls how quickly disorder settles in.

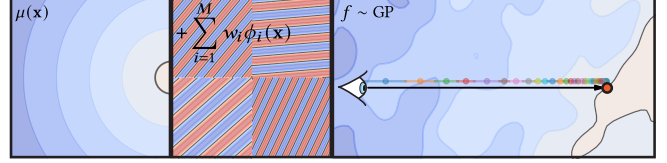


Fig. 1. We can use weight-space sampling to directly estimate the naive form of the ensemble average light transport in Eq. (8) for stationary kernels. We first sample a collection of basis functions based on the kernel and then a realization by choosing weights for each basis function (left). This gives us a fixed global realization that we can evaluate at arbitrary points in constant time. We then use affine arithmetic-based ray tracing [Sharp and Jacobson 2022] to find ray-surface intersection (right).

*Thin-plate.* Williams and Rasmussen [2006] suggest the “thin-plate” covariance. It is motivated by the fact that posterior GPISes should not “return to zero” for locations far away from the conditioning points. Instead, it is a more reasonable assumption that the magnitude of the sampled realizations should increase as you move away from the surface sample.

$$k^{\text{TP}}(x, y) = \sigma^2 (2\|x - y\|^3 + 3R\|x - y\|^2 + R^3) \quad (24)$$

$$p^{\text{TP}}(\omega) = \frac{3\sigma^2}{\pi^4 \omega^6} \quad (25)$$

Note that this covariance is only p.s.d. for  $\|x - y\| \leq R$ , so  $R$  has to be chosen to be at least as large as the size of the domain.

### 3 ALGORITHMS

In this section, we discuss the rendering algorithms we use to visualize Gaussian process implicit surfaces in more detail. These algorithms are, essentially, direct Monte Carlo estimators of Eq. (8) and Eq. (14). In particular, we formulate them in such a way that we avoid ever having to evaluate the GPIS  $\Gamma(t, \mathbf{n})$  density directly.

#### 3.1 Global realization sampling via weight-space GPs

We first describe a method to directly compute Eq. (9). Recall that there, we require realizations that span the whole space we want to trace paths through at once, and sampling from high dimensional multivariate Gaussian distribution quickly becomes expensive, as discussed in Section 3. If we limit ourselves to stationary kernels, we can apply the weight space decomposition to the Gaussian process instead [Wilson et al. 2020]. Here we approximate realizations as a weighted sum of a finite number  $M$  of basis functions, e.g. random Fourier features  $\phi_i(x) = \sqrt{2/M} \cos(\theta_i^\top x + \tau_i)$  where  $\theta_i$  are chosen according to the spectral density  $S(s) = \int k(r) J_{d/2-1}(2\pi r s) r^{d/2} dr$  of the covariance function, with  $J_{d/2-1}$  being a Bessel function of order  $d/2 - 1$ . Additionally,  $\tau_i \sim \text{Uniform}(0, 2\pi)$  and  $d$  is the dimensionality of the domain. Sampling a realization is then equivalent to sampling the weights for the basis function, and for an unconditioned process, the weights are independently Gaussian distributed. The realization can then be evaluated at *any* set of points  $X$  in  $O(N)$  time as

$$f(X) \approx \sum_{i=1}^M w_i \phi_i(X), \quad w_i \sim \mathcal{N}(0, 1). \quad (26)$$

This lets us draw correlated samples at a cost only dependent on the chosen number of basis functions instead of on the number of correlated samples.

**3.1.1 Practical considerations.** In particular, once we have sampled weights for the basis functions, we can treat the resulting linear combination like any other implicit function and trace against it by finding the first zero crossing along each ray. Unfortunately, we can’t use a fast root-finding method like sphere tracing because the Lipschitz constant of our basis grows linearly with the number of basis functions, forcing sphere tracing to take very small steps. Instead, we rely on the slower but more general method of root-finding via affine arithmetic [Sharp and Jacobson 2022]. Because the basis functions use a limited number of operations, implementing them in affine arithmetic is relatively straightforward. When we have found an intersection, we can analytically compute the normal at that point and do shading as for any other implicit surface. We illustrate this in Fig. 1. Note that because the basis functions use non-linear operations (cosines), the bounds produced by affine arithmetic are not tight. Nonetheless, we are guaranteed not to miss intersections. Weight-space GPIS rendering makes it practical to get results equivalent to the Global  $\infty$  model, assuming that the number of basis functions is chosen to be large enough. This means that this method is mainly useful for verifying more general ones, such as the function-space algorithms we will discuss next, in simplified scenarios. There are several practical concerns when using weight-space GPs for GPIS rendering. The most obvious one is that the standard weight-space formulation can only support stationary covariance kernels. The mean can still vary arbitrarily, allowing us to model interesting scenes, but we can not vary properties such as the roughness of surface-type GPIS or include surface-type and volume-type GPIS in the same scene. Choosing the correct number of basis functions is critical, and the exact number depends on the covariance kernel. When choosing a low number of basis functions, the underlying kernel is approximated poorly, which leads to significantly different realizations (see Fig. 2). Empirically, we observe that kernels with longer-than-exponential tails, such as the rational quadratic kernel, require a much larger number of basis functions

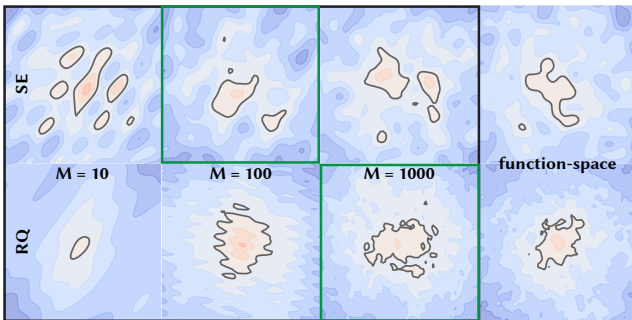


Fig. 2. When we choose the number of basis functions we need to take the roughness of the covariance kernel into account (top: smooth SE kernel, bottom: rough RQ kernel). For rougher processes, too few basis functions result in biased realization samples that do not predict the ground truth kernel well. (left to right,  $M = 10, 100, 1000$ , ground truth).

than, for example, the squared exponential kernel. Finally, one has to be able to sample from the spectral density of the chosen covariance kernel. This is not always trivial to do, and we provide procedures for the stationary kernels used in this work in Section 2.1 of the supplemental.

## 3.2 Practical function-space sampling strategies

In Section 4.2 of the main text we discussed our function-space sampling strategy and left off with two practical issues to solve in an implementation: Determining scene bounds along rays and limiting the number of correlated values that have to be sampled jointly. Here we go into more detail on how we handle both of these issues.

**Determining scene bounds.** One simple and practical solution to define scene bounds is to prescribe them *a priori*. For example, we can simply intersect the ray with a box or a sphere that we define as the limits of the scene, resulting in only having to sample a realization over a finite ray segment. A more principled option is to consider that many scenes are naturally finite in extent. In our case, this occurs when  $\mu(\mathbf{x}) \rightarrow \infty$  for large  $\mathbf{x}$ , e.g. when the mean surface is modeled using an SDF. Then, for large  $\mathbf{x}$ ,  $P(f(\mathbf{x}) \leq 0) \rightarrow 0$ , i.e. the probability of sampling a value that we see as “inside” a GPIS goes towards 0. We can define the bounds of the scene as the region for which  $P(f(\mathbf{x}) \leq 0) > \epsilon$  for some small epsilon. That is, we discard regions of space for which we are very unlikely to sample a zero crossing. Evaluating this “point-wise occupancy” is cheap as it is just the CDF of a mono-variate normal distribution. We can intersect the ray with the level set  $\{\mathbf{x} \mid P(f(\mathbf{x}) \leq 0) = \epsilon\}$  using ray marching. Accuracy here is not critical as long as we are conservative. This gives us a finite ray segment to work with as long as our scene is finite. The only case this does not cover is if we have an *infinite* scene. The most common example of this would be a scene filled with an infinite volume-type GPIS. Here we will have to rely fully on the second level of our strategy.

**Progressive ray segment sampling.** There are two cases where we can’t simply distribute sample points along a ray segment: We have an infinitely large scene, or the ray segment is long compared to the length scale of the covariance kernel (in which case the  $O(n^3)$  scaling of function-space sampling would make sampling a sufficiently dense set of points too slow). In practice, we found that taking anything more than 256 correlated samples at a time significantly impacts render times. To overcome this, we apply the same strategy we applied in Section 4 to path segments to ray segments. That is, we sample realizations  $f(\mathbf{x}_{t_i}, \mathbf{x}_{t_{i+1}}) \sim \text{GP}(\mathbf{x}_{t_i}, \mathbf{x}_{t_{i+1}}) | \zeta \wedge \zeta'$  with  $t_0 = 0$  and  $t_{i+1} = t_i + n \cdot \Delta t$ , where  $\Delta t$  is a step size chosen based on the covariance kernel, and  $\zeta'$  the condition based on the chosen memory model. We repeat this until  $\mathbb{I}(f(\mathbf{x}_{t_i}, \mathbf{x}_{t_{i+1}})) = 0$ , i.e.  $f(\mathbf{x}_{t_i}, \mathbf{x}_{t_{i+1}})$  contains a zero crossing. We then proceed as before, sampling a normal followed by an outgoing direction. This process is illustrated in Fig. 8.

### 3.3 Thoughts on next-event estimation

Our function-space sampling approach enables us to employ next-event estimation to a much greater degree than the global weight-space sampling approach. When determining a global realization before tracing a path, we fix not only the realization’s values but also its gradients and, hence, normals. Then, during tracing, there is exactly one possible normal at each intersection point. This means that, when using a mirror micro-BSDF, we cannot perform next-event estimation since, with the normal at the intersection point being deterministic, the outgoing ray direction is completely determined. In function-space sampling, the normal at an intersection is not determined ahead of time. Instead, we sample it from a distribution conditioned on the realization values we saw along the incoming ray. This realization only fixes the directional derivative of the GP at the intersection point but leaves us two additional degrees of freedom to choose a gradient and, hence, a normal. Assuming non-degenerate covariance kernels (i.e. volume-type GPISes, that do not assume perfect correlation along any axis), the resulting gradient distribution is non-degenerate and assigns some density to the whole hemisphere of normals facing the ray. Hence, even when using a mirror micro-BSDF, we can always find a normal such that the reflected ray points into the desired NEE direction. Computing the density of this distribution for a given normal is not trivial. The conditioned distribution of *gradients* is a 3D multivariate Gaussian distribution (no matter how complex the conditioning is). To compute the (unnormalized) density of sampling a given normal, we simply need to integrate over all possible gradients that normalize to that normal as

$$p(\mathbf{n}) = \int_0^\infty f(\mathbf{n} * t) dt, \quad (27)$$

where  $f$  is the pdf of the conditioned gradient distribution at the intersection point. Our preliminary investigation has shown that the integral in Eq. (27) is tractable analytically, and we should be able to compute the normalization constant to turn  $p(\mathbf{n})$  into a true pdf. This would then allow us to evaluate the conditioned visible normal distribution, enabling (single scattering) NEE for mirror micro-BSDFs. We leave this for future work but still apply classical NEE after sampling a normal when the micro-BSDF is diffuse.

### 3.4 Implementation

We implemented the sampling strategies described in this section in the Tungsten renderer [Bitterli 2018] with a focus on correctness over performance. A GPIS is treated as a participating medium that allows for sampling free-flight distances and computing transmittance, and provides a phase function at scattering locations. Light and camera positions are uncorrelated from the GPIS and we often place uncorrelated, non-GPIS surfaces in the scene alongside the GPIS we are investigating. We could, of course, represent these as uncorrelated GPIS as well, but that would unnecessarily increase rendering times without aiding in the validation or understanding of our method. We use standard data structures such as OpenVDB grids to store volumetric scene data (variance, length scale, and mean). For the mean, we also support using a mesh directly and computing SDFs on the fly.

## 4 UNCERTAINTY QUANTIFICATION USING DOUBLE MONTE CARLO SAMPLING

We have a model  $M^\Phi(x)$  with uncertain parameters  $\Phi \sim p_\Phi$ . We would like to compute the moments of the distribution of model outputs  $q_x(y_x = M^\Phi(x))$ . This problem is known as *uncertainty quantification*. If we have access to a deterministic method of evaluation  $M^\Phi(x)$ , we can compute

$$\mathbb{E}_{\Phi \sim p_\Phi} [M^\Phi(x)] = \int M^\Phi(x) dp_\Phi(\Phi) \quad (28)$$

$$\mathbb{V}_{\Phi \sim p_\Phi} [M^\Phi(x)] = \mathbb{E}_{\Phi \sim p_\Phi} [M^\Phi(x)^2] - \mathbb{E}_{\Phi \sim p_\Phi} [M^\Phi(x)]^2 \quad (29)$$

$$= \int M^\Phi(x)^2 dp_\Phi(\Phi) - \left( \int M^\Phi(x) dp_\Phi(\Phi) \right)^2 \quad (30)$$

and use Monte Carlo integration and sample variance to compute an estimate of the mean and variance, respectively as

$$\mathbb{E}_{\Phi \sim p_\Phi} [\widehat{M^\Phi(x)}]_N = \frac{1}{N} \sum_{i=0}^N M^{\Phi_i}(x) \quad \Phi_i \sim p_\Phi \quad (31)$$

$$\mathbb{V}_{\Phi \sim p_\Phi} [\widehat{M^\Phi(x)}]_N = \frac{1}{N} \sum_{i=0}^N M^{\Phi_i}(x)^2 - \left( \frac{1}{N} \sum_{i=0}^N M^{\Phi_i}(x) \right)^2 \quad \Phi_i \sim p_\Phi \quad (32)$$

Note that due to the square in the second term and Jensen’s inequality, we need a relatively large number of samples to get an unbiased estimate of variance. In practice, this is not a big issue since we tend to have the samples available anyway if we want to compute a relatively converged estimate of the mean. That is, if we have enough samples to estimate the mean, we tend also to have enough samples to estimate variance.

Unfortunately, in many graphics applications like rendering, the model itself is often in the form of a complex integral equation, such as

$$M^\Phi(x) = \int m^\Phi(x, y) dy \quad (33)$$

and has to be estimated using Monte Carlo techniques. That is, we only have access to an estimate

$$\widehat{M^\Phi(x)}_N = \sum_{i=0}^N \frac{m^\Phi(x, y_i)}{p_y(y_i)} \quad y_i \sim p_y. \quad (34)$$

This is not an issue when computing the mean of our model predictions. Since a one-sample Monte Carlo estimator is unbiased, that is

$$M^\Phi(x) = \mathbb{E} [\widehat{M^\Phi(x)}_1] = \mathbb{E}_{y \sim p_y} \left[ \frac{m^\Phi(x, y)}{p_y(y)} \right], \quad (35)$$

we can write

$$\mathbb{E}_{\Phi \sim p_\Phi} [M^\Phi(x)] = \mathbb{E}_{\Phi \sim p_\Phi} \left[ \mathbb{E} [\widehat{M^\Phi(x)}_1 \mid \Phi] \right] \quad (36)$$

$$= \mathbb{E}_{\Phi \sim p_\Phi} \left[ \mathbb{E}_{y \sim p_y} \left[ \frac{m^\Phi(x, y)}{p_y(y)} \mid \Phi \right] \right] \quad (37)$$

$$= \mathbb{E}_{\Phi \sim p_\Phi, y \sim p_y} \left[ \frac{m^\Phi(x, y)}{p_y(y)} \right] \quad (38)$$

and then use Monte Carlo integration to estimate the expectation as

$$\mathbb{E}_{\Phi \sim p_\Phi, y \sim p_y} \left[ \frac{m^\Phi(x, y)}{p_y(y)} \right] \approx \frac{1}{N} \sum_{i=0}^N \frac{m^{\Phi_i}(x, y_i)}{p_y(y_i)} \quad \Phi_i \sim p_\Phi, y_i \sim p_y \quad (39)$$

Note that even though we are sampling two random variables now, we can still just average  $N$  independent evaluations of  $m^{\Phi_i}(x, y_i)$ . This is one of the central benefits of Monte Carlo integration; its convergence does not depend on the dimensionality of the problem.

But for the variance, we have

$$\mathbb{V}_{\Phi \sim p_\Phi} \left[ M^\Phi(x) \right] \neq \mathbb{V}_{\Phi \sim p_\Phi, y \sim p_y} \left[ \frac{m^\Phi(x, y)}{p_y(y)} \right] \quad (40)$$

Intuitively, this is because  $\mathbb{V} \left[ \widehat{M^\Phi(x)}_N \mid \Phi \right] \neq 0$  for  $N < \infty$  and we have, according to the law of total variance

$$\begin{aligned} \mathbb{V}_{\Phi \sim p_\Phi} \left[ \widehat{M^\Phi(x)}_N \right] &= \\ &\mathbb{E}_{y_1, \dots, y_N \sim p_y} \left[ \mathbb{V}_{\Phi \sim p_\Phi} \left[ \widehat{M^\Phi(x)}_N \mid y_1, \dots, y_N \right] \right] \\ &+ \mathbb{V}_{y_1, \dots, y_N \sim p_y} \left[ \mathbb{E}_{\Phi \sim p_\Phi} \left[ \widehat{M^\Phi(x)}_N \mid y_1, \dots, y_N \right] \right]. \quad (41) \end{aligned}$$

Here, we can see that for finite  $N$ , the total variance is comprised of the expected value of the variance of our model due to model parameters and the variance due to the Monte Carlo estimation of our model.

We only recover the ground variance of the model due to the model parameters as  $N \rightarrow \infty$ :

$$\lim_{N \rightarrow \infty} \mathbb{E}_{y_1, \dots, y_N \sim p_y} \left[ \mathbb{V}_{\Phi \sim p_\Phi} \left[ \widehat{M^\Phi(x)}_N \mid y_1, \dots, y_N \right] \right] = \mathbb{V}_{\Phi \sim p_\Phi} \left[ M^\Phi(x) \right] \quad (42)$$

$$\lim_{N \rightarrow \infty} \mathbb{V}_{y_1, \dots, y_N \sim p_y} \left[ \mathbb{E}_{\Phi \sim p_\Phi} \left[ \widehat{M^\Phi(x)}_N \mid y_1, \dots, y_N \right] \right] = 0 \quad (43)$$

Intuitively, now that means that if we want to compute the sample variance of our model due to the model parameters unaffected by variance due to Monte Carlo estimation of the model, we need to sample many  $y_j \sim p_y$  for each  $\Phi_i \sim p_\Phi$ . Even if we reuse the same set of  $y$ s, we still need to evaluate  $\widehat{M^{\Phi_i}(x)}_N$  separately for each  $\Phi_i$  and compute sample variance as

$$\begin{aligned} \mathbb{V}_{\Phi \sim p_\Phi} \left[ \widehat{M^\Phi(x)}_N \right]_{N,K} &= \frac{1}{N} \sum_{i=0}^N \left( \frac{1}{K} \sum_{j=0}^K \frac{m^{\Phi_i}(x, y_j)}{p_y(y_j)} \right)^2 \\ &- \left( \frac{1}{NK} \sum_{i=0}^N \sum_{j=0}^K \frac{m^{\Phi_i}(x, y_j)}{p_y(y_j)} \right)^2 \quad \Phi_i \sim p_\Phi, y_j \sim p_y. \quad (44) \end{aligned}$$

This now results in quadratic complexity  $O(NK)$  where  $N$  controls the bias due to the limited number of parameter samples and  $K$  controls the bias due to the additional variance in the Monte Carlo estimator. Hence, while it is certainly possible to do uncertainty quantification via Monte Carlo rendering of GPISes, it is not trivial to do so efficiently and we leave this for future work.

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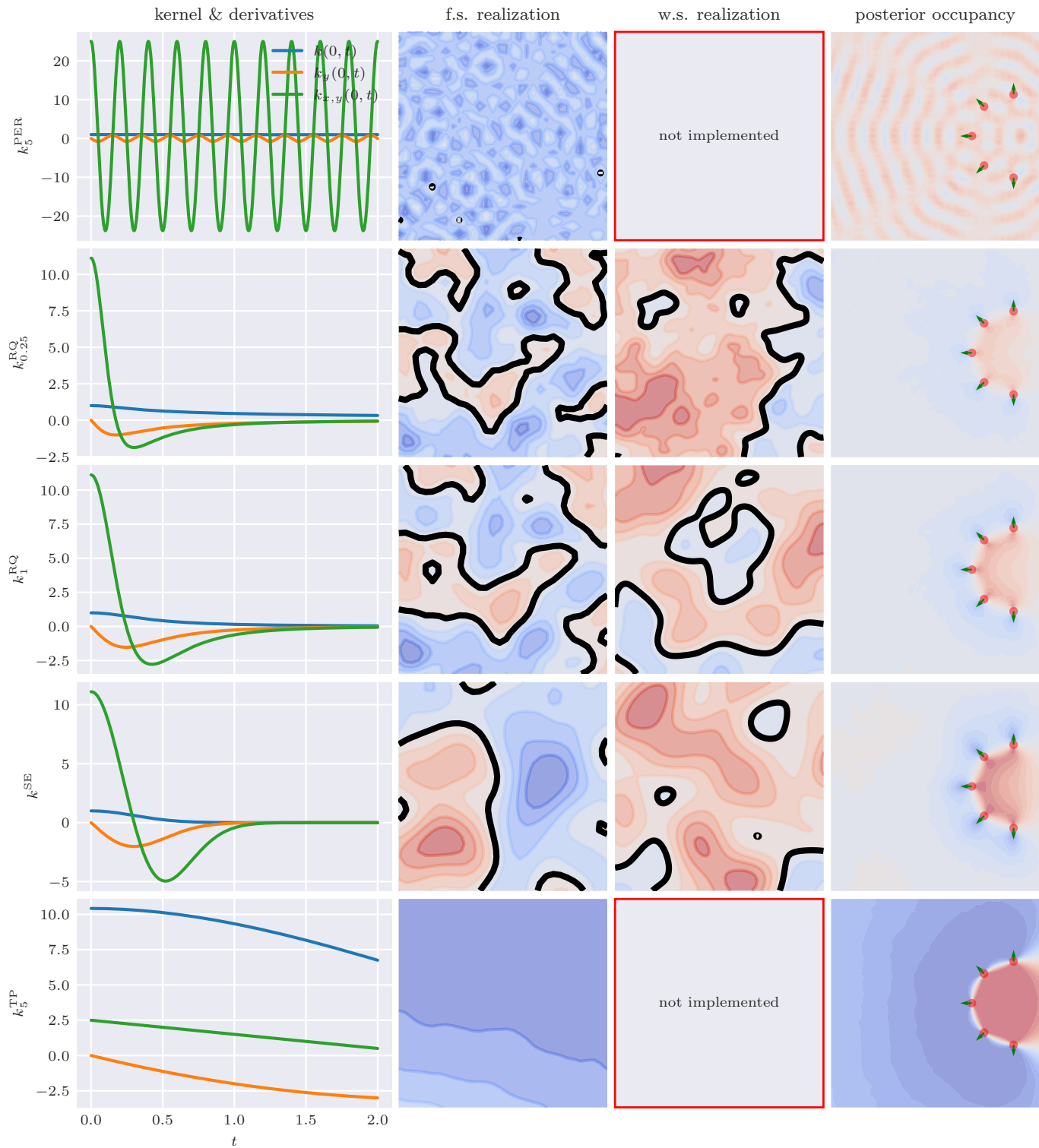


Fig. 3. We show an overview of some of the kernels that we implemented for our method. To give an intuition for the family of GPISes each kernel produces, we show samples from the prior (produced via both function-space and, where applicable, weight-space methods) and posterior occupancy.