

Attaining quantum limited precision of localizing an object in passive imagingAqil Sajjad,^{1,*} Michael R. Grace^{1,†}, Quntao Zhuang,^{2,1} and Saikat Guha^{1,2,‡}¹*James C. Wyant College of Optical Sciences, University of Arizona, Tucson, Arizona 85721, USA*²*Department of Electrical and Computer Engineering, University of Arizona, Tucson, Arizona 85721, USA*

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We investigate our ability to determine the mean position, or centroid, of a linear array of equally bright incoherent point sources of light whose continuum limit is the problem of estimating the center of a uniformly radiating object. We consider two receivers, an image-plane ideal direct-detection imager and a receiver that employs Hermite-Gaussian (HG) spatial-mode demultiplexing in the image plane, prior to shot-noise-limited photon detection. We compare the Fisher information (FI) for estimating the centroid achieved by these two receivers, which quantifies the information-accrual rate per photon, and compare those with the quantum Fisher information (QFI): the maximum attainable FI by any choice of measurement on the collected light allowed by physics. We find that focal-plane direct imaging is strictly suboptimal, although not by a large margin. We also find that not only is the HG mode sorter, which is the optimal measurement for estimating the separation between point sources (or the length of a line object), suboptimal, but it performs worse than direct imaging. We study the scaling behavior of the QFI and direct imaging's FI for a continuous uniformly bright object in terms of its length and find that both are inversely proportional to the object's length when it is sufficiently larger than the Rayleigh length. Finally, we propose a two-stage adaptive modal receiver design that attains the QFI for centroid estimation.

DOI: [10.1103/PhysRevA.104.022410](https://doi.org/10.1103/PhysRevA.104.022410)**I. INTRODUCTION**

Rayleigh's criterion for the resolution of two incoherent light sources [1] remains one of the most important results in optical imaging. Based on diffraction effects, it tells us that we cannot resolve two objects whose angular separation is less than λ/D , where λ is the wavelength and D is the size of the receiver's aperture. This relies on a somewhat heuristic argument. A more rigorous estimate for the maximum estimation precision for the separation between the two pointlike sources, when imaged by an ideal direct-detection focal plane array, can be obtained from the classical Cramér-Rao bound [2], expressed in terms of the Fisher information (FI). This analysis still shows that the FI, whose inverse gives a lower bound on the variance of any unbiased estimator and therefore serves as a measure for precision, sharply drops and approaches zero as the separation between the two sources falls below λ/D , approaching zero. In qualitative terms, both Rayleigh's criterion and the Cramér-Rao bound are essentially telling us the same common sense thing: If two objects are too close to each other, then it is hard to tell them apart, and hence determine their separation, if their *images*, blurred by the point-spread function (PSF) of the aperture, overlap so much that it is hard to tell if the image is that of a single point source or that of two closely spaced sources. This also translates into our inability to resolve any features of the object that are

too small compared to λ/D . Obtaining relevant information about objects in the sub-Rayleigh regime has therefore been a major topic of interest for a wide variety of fields ranging from astronomy [3,4] to biological imaging [5,6].

A. Superresolution imaging using predetection mode sorting

Recent findings based on quantum estimation theory show that it is possible to build new imaging devices that surpass Rayleigh's limit. One useful tool in quantum estimation theory is the quantum Cramér-Rao bound, introduced by Helstrom [7], expressed in terms of the quantum Fisher information (QFI): The QFI is an upper bound to the maximum FI attainable with any physically allowed measurement scheme. Thus, if we find that the FI for a given measurement (the physical device that detects the information bearing light producing an electrical signal) is equal to the QFI, then we know that the said measurement is optimal and no other measurement in its place will generate an estimate of the parameter of interest with a lower variance. The inverse statement also holds for estimating a single scalar parameter: If we find that there is a gap between the QFI and the FI of a specific measurement, then it means that we can do better by employing some other measurement whose FI is equal to the QFI.

It turns out that for estimating the separation between two incoherent point sources, the QFI remains constant instead of shrinking to zero as the separation reduces to zero [8], thereby proving that Rayleigh's criterion, in its commonly stated form, is an artifact of *direct*, i.e., intensity, detection in the image plane (corrupted by the fundamental Poisson shot noise), which discards valuable information in the phase

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of the field. For the case of a Gaussian PSF, Tsang *et al.* showed that the QFI-attaining (optimal) measurement can be realized by an image-plane Hermite-Gaussian (HG) spatial-mode demultiplexer (SPADE), followed by shot-noise-limited photon detection on those sorted modes. Along the same lines, Kerviche *et al.* [9] and Řeháček *et al.* [10] independently showed that the optimal measurement for a hard aperture pupil (sinc-function PSF) is realizable with an image-plane sinc-Bessel SPADE followed by photon detection. Moreover, for the same problem with an arbitrary aperture function, both these works propose measuring in a basis comprising the PSF and its orthogonalized derivatives, with the latter proving that such a measurement attains the quantum-optimal performance provided the PSF is an even function. In [11], Dutton *et al.* generalized the work of Tsang *et al.* to the case of estimating the angular extent of $M > 1$ equidistant equally bright point sources. In the limit where the number of points goes to infinity, this equates to the problem of estimating the length of a continuous line-shaped object with uniform brightness. It was shown that an image-plane HG SPADE is again the optimal measurement for estimating the length of such a uniform-brightness line object [11]. In [12], the results of Tsang *et al.* were extended to two dimensions, showing that the two-dimensional HG basis is quantum optimal. A three-dimensional generalization was the subject of [13], where the so-called Zernike basis functions were shown to attain the QFI.

These mode sorters, however, need to be pointed exactly at the centroid of the incoherent point sources in order to obtain an FI for estimating the separation that equals the QFI. In fact, Tsang *et al.* showed that even a somewhat small misalignment of the SPADE can result in a large drop in its performance, especially in the regime where the separation is much smaller than the Rayleigh limit. In this limit, it is even possible to simultaneously estimate the separation and centroid of two (not necessarily equal brightness) point sources optimally; however, such a measurement likewise needs to be spatially aligned with respect to the intensity-weighted centroid of the sources in order to avoid a significant loss of performance [14]. This means that if the centroid is not perfectly known *a priori*, we first need to estimate it before we carry out the SPADE measurement to determine the separation. Based on this intuition, a two-stage optimization scheme was proposed by Grace *et al.* in [15]. This receiver first estimates the centroid, a nuisance parameter, using direct detection, and once a good enough estimate has been obtained, the system then switches to the SPADE for finding the separation. In addition to the above motivation for estimating the centroid first when we are mainly interested in estimating the separation, there is also the fact that finding the location of an object is an important problem in its own right. We can be interested, for instance, in locating a known object whose size and features we already know.

All the above-mentioned works on estimating the separation generally assume that the centroid can be determined fairly accurately from direct imaging. This intuition in part stems from the fact that Helstrom showed that for a single point source, ideal image-plane direct detection is the optimal measurement to estimate its position [7]. Moreover, Tsang *et al.* showed that the Fisher information for the cen-

troid for two equally bright sources, while less than the QFI, is not suboptimal by a very large difference.

B. Main results

Since SPADE-like measurements are so sensitive to misalignment, even small improvements in estimating the centroid can be beneficial. With that in mind, in this paper we present a thorough study of the optimal measurement to estimate the centroid for two or more equally bright incoherent point sources, assuming a Gaussian PSF, including the case of an infinite number of equally bright equally spaced point sources in a straight line of sub-Rayleigh length. The infinite case is naturally of special interest because we want to be able to estimate the position of continuum objects, such as localizing a star or planet (in astronomical imaging) or localizing a cellular structure (in biological imaging).

We investigate the performance of both direct imaging and the HG SPADE for centroid estimation. The interest in the latter arises from it being the optimal measure for the separation, and it is therefore worth studying whether it can also be useful for finding the centroid. This is also of interest from the perspective of a two-stage detection scheme such as the one proposed in [15], where first the centroid would be determined using direct imaging or some other more optimal measurement, and once a reasonable estimate has been made for it, the device would switch automatically to the HG SPADE measurement for finding the separation. We show that for two or more sources with equal separation, direct imaging is in fact *not* an optimal measurement for estimating the centroid. We also find that the HG SPADE is more suboptimal than direct imaging. Moreover, the better the HG SPADE is aligned with the centroid, the worse its ability to determine the centroid.

Finally, we present a two-stage adaptive modal measurement strategy that achieves the QFI for estimating the centroid of a constellation of n equally bright equally spaced point sources. The strategy we present applies to finding the QFI-attaining receiver measurement for any n -point constellation.

C. Organization of the paper

In Sec. II we introduce the overall setup for centroid estimation and the underlying assumptions of the model, give a brief overview of the classical and quantum Cramér-Rao bounds, and summarize the key findings of [8,11] that are most relevant for our study of centroid estimation. In Sec. III we present our results on centroid estimation for different numbers of equidistant uniformly bright emitters placed in a single line, including the infinite case of a uniformly bright object. We compare the performance of direct imaging and the HG SPADE with the QFI and show that direct imaging is suboptimal for centroid estimation, but not by a substantial amount. We then go on to discuss the performance of the HG SPADE and show that it is mostly worse than direct imaging for locating an object, even though it gives us the QFI-attaining measurement for finding the size or end-to-end diameter. Finally, we present a two-stage measurement scheme that attains the QFI for centroid estimation in the limit of large integration time. In Sec. IV we present a summary and our conclusions. We also include several Appendixes, proving

important results and explaining known calculation methods but with additional details and clarifications that have been generally skipped in the literature and may be helpful for the reader.

II. PHYSICAL SETUP

For the setup and our basic assumptions about the physics, we closely follow the framework laid out in [8], except that we generalize it to more than two incoherently radiating point sources. For simplicity, we assume that the object and image planes are one dimensional with unit magnification and that our light sources emit nearly monochromatic light with paraxial waves [1]. We also make the standard assumption, valid for optical-frequency radiation, that the average number of photons ϵ per temporal mode arriving at the image plane is much less than one, requiring many photons to be detected over a large number of temporal modes, to extract any useful information [16–20].

A. Quantum model for an imaging scene made up of incoherently radiating point emitters

Let λ be the center wavelength, W (measured in hertz) the spectral bandwidth of the collected light (around λ), and T (measured in seconds) the integration time. In that time-bandwidth window, there are roughly $M \approx WT$ mutually orthogonal temporal modes. We take $N = M\epsilon$ to be the mean number of photons received over the integration time, where ϵ is the mean number of photons collected per temporal mode. We now write the density operator of the photon field in a single temporal mode over the infinite number of mutually orthogonal spatial modes spanning the receiver telescope’s entrance pupil’s spatial extent. In the (conventional) image plane, this density operator can be written as

$$\rho = (1 - \epsilon)\rho_0 + \epsilon\rho_1 + O(\epsilon^2), \tag{1}$$

where $\rho_0 = |\mathbf{0}\rangle\langle\mathbf{0}|$ is the zero-photon or vacuum state and ρ_1 is a single-photon quantum state, both of a single temporal mode over some infinite-basis of spatial modes. In addition, $O(\epsilon^2)$ denotes higher-order terms in ϵ , which we will ignore, since at visible frequencies, $\epsilon \ll 1$. The quantum state of all the collected light during the integration is given by $\rho^{\otimes M}$.

The one-photon state ρ_1 is a mixed state: an incoherent mixture of states $|\psi_s\rangle$, a pure state of one photon, spread over an infinite basis of spatial modes, of the image-plane field, corresponding to the s th point source making up the overall scene. For a scene comprised of n equally bright point sources,

$$\rho_1 = \frac{1}{n} \sum_{s=1}^n |\psi_s\rangle\langle\psi_s|, \tag{2}$$

with

$$|\psi_s\rangle = \int_{-\infty}^{\infty} dx \psi(x - x_s)|x\rangle, \tag{3}$$

where $\psi(x)$ is our coherent PSF, x_s is the position of the s th point source, and $|x\rangle = \hat{a}^\dagger(x)|\mathbf{0}\rangle$ is the (unphysical) state of one photon localized exactly at the spatial point x in the image plane, where the annihilation and creation operators obey the δ -function commutator $[\hat{a}(x), \hat{a}^\dagger(x')] = \delta(x - x')$. We will

consider a Gaussian PSF

$$\psi(x) = \frac{1}{(2\pi\sigma^2)^{1/4}} \exp(-x^2/4\sigma^2), \tag{4}$$

where $\sigma = 1/2\Delta k = \lambda/2\pi\mathcal{N}$, with $\Delta k^2 \equiv \int_{-\infty}^{\infty} [\partial\psi(x)/\partial x]^2 dx$, λ the center wavelength, and \mathcal{N} the effective numerical aperture.

Now let us consider the model for ideal direct imaging in the traditional image plane: an infinite-size continuum active surface (i.e., infinitely small detector pixels with unity fill factor), where each of those pixels is a unity quantum efficiency shot-noise-limited photon-number-resolving detector with an infinite bandwidth, no read noise, and no dead time. This ideal continuum detector array generates a spatiotemporal photocurrent process that, for the aforesaid model of collected light, is characterized by a space-time Poisson point process with a rate given by the squared-magnitude photon-unit field in the image plane. We remind the reader that, per our model, for each of the M temporal modes, at most one photon can be detected, since $\epsilon \ll 1$. Given there is a photon in a particular temporal mode, the spatial probability density of its detection is given by

$$\Lambda(x) = \frac{1}{n} \sum_{s=1}^n |\langle x|\psi_s\rangle|^2 = \frac{1}{n} \sum_{s=1}^n |\psi(x - x_s)|^2. \tag{5}$$

Therefore, over a single temporal mode, the expectation value of the number of photons being detected in a region of width dx around x is then given by a Poisson distribution with the mean of $\epsilon\Lambda(x)$ [18,21–23]. Over M temporal modes, the average number of photons detected (over the entire detector array) becomes $N = M\epsilon$, with an average of $N\Lambda(x)dx$ photons in a region of width dx around x in the image plane.

In general, if we instead measure the received optical field by some other receiver (e.g., homodyne detection), which is associated with an observable $\hat{\mathcal{Y}}$, then the probability distribution of measurement outcomes is given by $P(\mathcal{Y}) = \langle \mathcal{Y} | \rho_1 | \mathcal{Y} \rangle$, where $\mathcal{Y} \in \mathbb{R}$ is a particular value of the observable and $|\mathcal{Y}\rangle$ is the eigenket associated with that value.

B. Quantum Fisher information and the Cramér-Rao bound

Let us say we are presented with N copies of a quantum state, i.e., $\rho^{\otimes N}$, and we wish to estimate a set of parameters $\{\theta_\mu\}$ embedded in ρ by measuring $\hat{\mathcal{Y}}$ on each copy of ρ . In other words, we have the classical estimation theory problem, wherein we wish to estimate parameters $\{\theta_\mu\}$ embedded in a random variable \mathcal{Y} , by N independent and identically distributed samples of \mathcal{Y} , each drawn from the distribution $P(\mathcal{Y}) = \langle \mathcal{Y} | \rho | \mathcal{Y} \rangle$. Consider a set of estimators $\tilde{\theta}_\mu(\mathcal{Y})$ and the error covariance matrix

$$\Sigma_{\mu\nu} \equiv \int d\mathcal{Y} P(\mathcal{Y}) [\tilde{\theta}_\mu(\mathcal{Y}) - \theta_\mu][\tilde{\theta}_\nu(\mathcal{Y}) - \theta_\nu]. \tag{6}$$

If $\tilde{\theta}_\mu(\mathcal{Y})$ is an unbiased estimator, then it obeys the Cramér-Rao bound on the covariance matrix

$$\Sigma \geq \mathcal{J}^{-1}, \tag{7}$$

where for N measurements

$$\mathcal{J}_{\mu\nu} \equiv N \int d\mathcal{Y} \frac{1}{P(\mathcal{Y})} \frac{\partial P(\mathcal{Y})}{\partial \theta_\mu} \frac{\partial P(\mathcal{Y})}{\partial \theta_\nu} \tag{8}$$

is the Fisher information matrix [2] associated with this specifically chosen receiver measurement. Moreover, if $\tilde{\theta}_\mu(\mathcal{Y})$ is the maximum-likelihood estimator, then we saturate the inequality in (7) for large N . The FI thus quantifies the performance of a measurement in determining the parameter we are interested in. Therefore, ideally, we want to choose a measurement that maximizes the FI.

The quantum Cramér-Rao bound gives us the maximum possible FI that *any* physically permissible measurement scheme could achieve. In other words,

$$\Sigma \geq \mathcal{J}^{-1} \geq \mathcal{K}^{-1}, \quad (9)$$

where \mathcal{K} is the quantum Fisher information matrix [7]. For $\rho^{\otimes N}$ encoding parameters of interest $\{\theta_\mu\}$, the QFI matrix is given by

$$\mathcal{K}_{\mu\nu}(\rho^{\otimes N}) \equiv N \text{tr}[\rho\{\mathcal{L}_\mu(\rho), \mathcal{L}_\nu(\rho)\}], \quad (10)$$

where $\mathcal{L}_\mu(\rho)$ is the symmetric logarithmic derivative (SLD) of ρ with respect to the parameter θ_μ . It is a Hermitian operator that is defined by the relation

$$\frac{\partial \rho}{\partial \theta_\mu} = \frac{1}{2}[\rho \mathcal{L}_\mu(\rho) + \mathcal{L}_\mu(\rho) \rho]. \quad (11)$$

If $\rho = \sum_j D_j |e_j\rangle\langle e_j|$ is the decomposition of ρ in terms of its eigenvalues D_j and eigenvectors $|e_j\rangle$, then the SLD is given by

$$\mathcal{L}_\mu(\rho) = \sum_{j,k:D_j+D_k \neq 0} \frac{2}{D_j + D_k} \langle e_j | \frac{\partial \rho}{\partial \theta_\mu} | e_k \rangle | e_j \rangle \langle e_k |. \quad (12)$$

Note that the QFI matrix does not depend on a particular choice of measurement, but is a property of the quantum state ρ . If the FI for a chosen measurement scheme is equal to the QFI, then we know that it is the best possible way to estimate the parameter(s) of interest.

C. Estimating geometrical parameters of a linear point source constellation

Where possible, it is convenient to work in terms of parameters that give diagonal QFI and FI. For our physical system of a linear constellation of equidistant uniformly bright light sources, it turns out that the QFI and the FI for direct imaging are both diagonal in terms of the centroid

$$\theta_1 = \frac{\sum_{s=1}^n x_s}{n} \quad (13)$$

and the separation between the first and the last point source

$$\theta_2 = x_n - x_1. \quad (14)$$

In terms of these two parameters, the individual positions of the point sources are given as

$$x_s = \theta_1 - \frac{\theta_2}{2} + \frac{(s-1)\theta_2}{n-1}, \quad 1 \leq s \leq n. \quad (15)$$

This diagonality of the QFI and the direct imaging FI in terms of θ_1 and θ_2 arises from the symmetry of our physical setup around the centroid, and we prove this in Appendix A. It is also worth noting that due to the physical symmetry around the centroid, the direct imaging FI and QFI matrices will be independent of θ_1 .

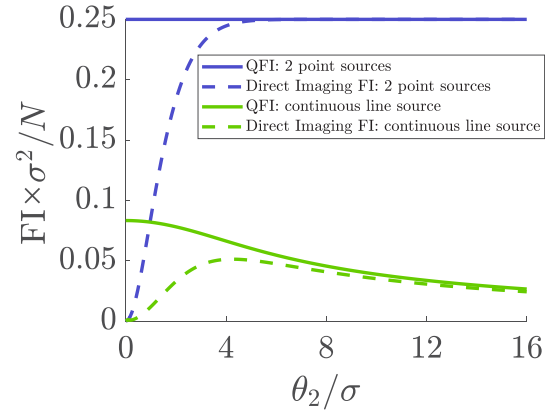


FIG. 1. QFI (solid lines) and direct-imaging FI (dashed lines) for estimating the separation θ_2 between n point sources, plotted as a function of θ_2 . We show two sets of plots: one for $n = 2$ point sources (dark blue) and the other corresponding to $n \rightarrow \infty$ (light green), which corresponds to a continuous line-shaped object. A Gaussian PSF with width σ is assumed.

Tsang *et al.* studied the problem of estimating the separation between two equally bright point sources, i.e., $n = 2$ in the above notation, assuming the centroid θ_1 is known *a priori* [8]. The Fisher information for estimating the separation using direct imaging approaches zero as the separation goes to zero (see Fig. 1). This is a manifestation of the so-called Rayleigh's curse, since the two sources become unresolvable when they are very close to each other, as their image-plane fields have a width comparable to their separation. However, Tsang *et al.* showed that the QFI for estimating θ_2 is a nonzero constant even when the separation approaches zero, the same constant the direct-imaging FI approaches when $\theta_2 \rightarrow \infty$ (see Fig. 1). This means that the so-called Rayleigh's curse is only an artifact of direct imaging rather than being a fundamental limit imposed by the PSF of the imaging system. They go on to show that if we carry out a measurement of the image-plane field using a Hermite Gaussian basis spatial-mode demultiplexer that is aligned perfectly with the centroid, and assuming that the centroid itself is perfectly known *a priori*, the FI attained by this measurement for estimating the separation θ_2 equals the QFI, hence being a quantum-optimal measurement scheme for estimating the separation.

These results were generalized by Dutton *et al.* to the problem of finding the end-to-end separation for an arbitrary number of point sources ($n \geq 2$) in [11], including the $n \rightarrow \infty$ case of a line-shaped object of length θ_2 . They again found that the QFI does not fall to zero even as the separation becomes small and that an HG SPADE aligned with the centroid attains the QFI.

We show these results in Fig. 1, where we reproduce the plots reported by Tsang *et al.* for the QFI and the FI for direct imaging for finding the separation between two incoherent emitters [8]. We also show the same quantities for the continuous line source, reproducing the results of Dutton *et al.* [11]. The QFI curves in both cases also represent the FI for the HG SPADE, since it attains the QFI. Additionally, we note that a binary SPADE measurement, in which only the zeroth (or the first) image-plane HG mode is detected by separating

it from the rest of the light (which is also detected using a bucket detector), attains the QFI in the limit of $\theta_2 \rightarrow 0$. This was shown for $n = 2$ in [8,9] and for $n \geq 2$ in [11].

It is however important to emphasize that all these results are strongly contingent on the alignment of the SPADE with the centroid. In fact, in the limit where $\theta_2 \rightarrow 0$ and $\theta_1 \gg \theta_2$, where the SPADE is aligned to the position $x = 0$, the FI for the HG SPADE drops all the way from attaining the QFI to being 0. This is discussed, for instance, in Appendix D of [8] as well as in [15].

Grace *et al.* studied the performance of a binary SPADE measurement to estimate the separation of two point sources when their centroid is not known *a priori* [15]. They considered a two-stage adaptive receiver, where image-plane direct imaging is employed in the first segment of the optical integration time to obtain an estimate of the centroid and the receiver then dynamically switches over to a second stage where a binary HG SPADE is employed whose center is aligned with respect of the (noisy) estimate of the centroid obtained from the first stage. Grace *et al.* developed an algorithm for that dynamic switching and the ensuing parameter estimation, which would enable a tenfold to 100-fold reduction in the integration time needed to obtain a desired (small) mean-square error in estimating θ_2 despite no prior information of θ_1 is assumed, compared to the scenario when image-plane direct detection is used for the entire integration time [15].

The choice of image-plane direct detection to obtain a preestimate of the centroid during the first stage of the aforesaid adaptive receiver was driven by intuition. Image-plane direct imaging is quantum optimal (attains QFI) for localizing a single point source [7], but suboptimal when it comes to estimating the centroid of two point sources [8]. Given the performance of SPADE-like measurements is extremely sensitive to misalignment, even small improvements in estimating the centroid can be very beneficial. This is our motivating reason to study the problem of centroid estimation.

In this paper we investigate the performance of both direct imaging and the HG SPADE for estimating the centroid θ_1 of $n \geq 2$ point sources in a line spanning an angular length of θ_2 , the former because it is the simplest and the most obvious measurement and the latter because it is worth asking if the SPADE can again outperform direct imaging in some region of parameter space for finding the centroid, just as it did for the separation. We also calculate the quantum limit (QFI) of centroid estimation to quantify the gaps to the FIs attained by the two aforesaid measurements. Finally, we describe an adaptive two-stage receiver design that would attain that QFI in the limit of long integration time.

III. QUANTUM LIMIT OF LOCALIZING AN OBJECT IN PASSIVE IMAGING

A. The QFI and FI of direct imaging for a linear constellation of point sources

We now consider the behavior of the direct imaging FI and the QFI for estimating the centroid θ_1 of a linear array of $n \geq 2$ equally spaced point emitters spanning a total angular extent θ_2 . First, it is worth noting that the QFI and direct imaging FI should both be independent of the value of the centroid θ_1 due

to the assumption of a linear shift-invariant physical imaging system. In the calculation of the FI for direct imaging, i.e., from the samples drawn from the spatial probability density $\Lambda(x)$ of photon clicks as in Eq. (5), this appears in the form of the shift symmetry of the variable of integration in (21) from x to $x - \theta_1$, which removes θ_1 from the integrand.

In the case of a single light source, Helstrom showed that direct imaging is quantum optimal [7]. The QFI and direct imaging FI for this case can be calculated easily as we show in Appendix B, and we find that they are both equal for any arbitrary PSF

$$\mathcal{K}_{1\text{-pt}} = \mathcal{J}_{1\text{-pt}} = 4N\Delta k^2, \quad (16)$$

where

$$\Delta k^2 \equiv \int_{-\infty}^{\infty} dx \left[\frac{\partial \psi(x)}{\partial x} \right]^2. \quad (17)$$

For our Gaussian PSF defined in (4), this yields N/σ^2 for the QFI and the direct imaging FI. This is a result that we will regularly use throughout the rest of this paper since all the cases involving two or more points also have special limiting points where the FI and the QFI will approach this value.

For two point sources, the QFI has been worked out analytically by Tsang *et al.* [8]. We describe their calculation in Appendix C and simply state the result here. For the diagonal component of the QFI matrix involving the centroid, i.e., the QFI for estimating the centroid, they obtained

$$\mathcal{K}_{11} = 4N(\Delta k^2 - \gamma^2), \quad (18)$$

where Δk^2 was defined in (17) and

$$\gamma = \int_{-\infty}^{\infty} dx \frac{\partial \psi(x)}{\partial x} \psi(x - \theta_2). \quad (19)$$

It is worth noting that γ goes to zero when θ_2 goes to zero or infinity for any symmetric PSF. When θ_2 goes to zero, the derivative of $\psi(x)$ is antisymmetric, so the integral in (19) tends to zero. On the other hand, when θ_2 becomes large, then $\psi(x - \theta_2)$ and $\partial \psi(x)/\partial x$ overlap very little with each other for PSFs $\psi(x)$ that fall off to zero away from the origin. Therefore, again, γ goes to zero. Consequently, the QFI approaches $4N\Delta k^2$ in these two limits, which is the result for the single-point-source case. In between, however, there is a region where γ is not zero and we get a QFI smaller than that for a single point source. It is this regime where direct imaging is unable to attain the QFI for estimating the centroid.

The physical explanation for this behavior is that when the separation is very small, the impulse response of two point sources, i.e., their aperture-blurred fields in the image plane, each of width σ , look like the impulse response of a single point source at origin. Therefore, the problem of centroid estimation in this limit should reduce to that of finding the location of a single point source for which image-plane direct detection is known to be quantum optimal [7]. For a slightly larger separation, the images of two point sources no longer overlap as much and the effect of diffraction is to cause a decrease in the QFI. However, when the separation becomes significantly larger than the width of the PSF, the images of the two point sources fully separate with no overlap, in which regime their individual positions can be estimated

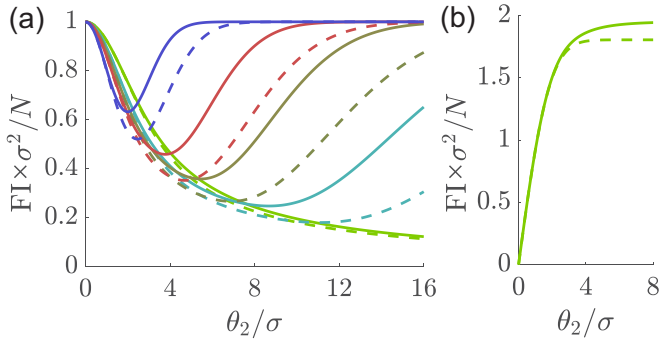


FIG. 2. (a) QFI (solid lines) and FI of direct imaging (dashed lines) for estimating the centroid of n point sources (dark to light: blue, $n = 2$; gold, $n = 3$; red, $n = 4$; cyan, $n = 6$; and green, continuous line source) with separation θ_2 given a Gaussian PSF with width σ . (b) QFI (solid line) and FI of direct imaging (dashed line) for a continuous line source with a constant photon flux per unit length of the source.

separately (again quantum optimally using image-plane direct detection), treating the two as single point sources. Each point source now emits half the light and therefore the QFI for its location is $2N\Delta k^2$, but the total sum is still $4N\Delta k^2$. For the specific case of our Gaussian PSF (4), we get

$$\mathcal{K}_{11} = \frac{N}{\sigma^2} - \frac{N\theta_2^2}{4\sigma^4} \exp\left(-\frac{\theta_2^2}{4\sigma^2}\right), \quad (20)$$

which approaches the single-point-source result of N/σ^2 in the $\theta_2 \rightarrow 0$ and $\theta_2 \rightarrow \infty$ limits with a dip in between as discussed above [see Fig. 2(a)].

The FI of the centroid from direct imaging for the two-point-source case has a somewhat similar qualitative behavior with the same physical intuition, except that its dip between the two limiting cases of $\theta \rightarrow 0$ and $\theta_2 \rightarrow \infty$ is deeper. It is given by

$$\mathcal{J}_{11} = N \int dx \frac{1}{\Lambda(x)} \left(\frac{\partial \Lambda(x)}{\partial \theta_1} \right)^2, \quad (21)$$

where $\Lambda(x)$ is the probability density given in (5). We are unable to do this integral analytically and therefore use numerical integration. The result was plotted along with the QFI in [8] and we reproduce it in Fig. 2 along with our results for when $n > 2$ point emitters constitute the scene. We see that the direct imaging FI approaches the QFI for small and large separation, as expected from the aforesaid intuitive explanation, for all $n \geq 2$. However, there is a gap in the region between these two limiting regimes. This gap is not too large, especially deep in the sub-Rayleigh regime. In particular, in the small- θ_2 regime for $n = 2$, we can see this explicitly by Taylor expanding the QFI (20) and also Taylor expanding the integrand of the direct imaging FI (21) in the $\theta_2 \rightarrow 0$ limit and integrating term by term. The resulting limiting behaviors for the QFI and direct imaging FI are given by

$$\mathcal{K}_{11} = \frac{N}{\sigma^2} - \frac{N\theta_2^2}{4\sigma^2} + \frac{N\theta_2^4}{16\sigma^6} - \frac{N\theta_2^6}{128\sigma^8} + O(\theta_2^8) \quad (22)$$

and

$$\mathcal{J}_{11} = \frac{N}{\sigma^2} - \frac{N\theta_2^2}{4\sigma^2} + \frac{N\theta_2^4}{16\sigma^6} - \frac{N\theta_2^6}{64\sigma^8} + O(\theta_2^8), \quad (23)$$

from which we see that the two quantities vary only in the sixth order in θ_2 . Tsang *et al.* argued [8] that we should be able to obtain a reasonable estimate for the centroid from direct imaging in order to correctly align the SPADE for estimating the separation, an intuition that was validated in the adaptive two-stage receiver designed and analyzed by Grace *et al.* [15].

Importantly, the above discussion using FI and QFI as performance benchmarking tools does not address the fact that if we are not in the regime $\theta_2/\sigma \ll 1$, the maximum-likelihood estimator of θ_1 , with either direct detection or the quantum-optimal measurement as the receiver choice, would in general also depend on the true (*a priori*-unknown) values of θ_1 and θ_2 . The fact that with θ_1 known *a priori* the optimal measurement and estimator to estimate θ_2 is the HG SPADE and is independent of the estimate of θ_2 was a happy coincidence.

We now generalize the result in [8] for the two-point-source case to a general number of incoherent point sources. For $n = 3$ or more emitters, calculating the QFI becomes increasingly complicated as it involves diagonalizing larger and larger matrices. We therefore perform these diagonalizations numerically. The detailed procedure we employ for this purpose is described in Appendix D and here we focus on the results. Figure 2(a) shows the plots of QFI and direct imaging FI against the end-to-end separation θ_2 for $n = 2, 3, 4$, and 7 emitters, as well as for a continuous line ($n \rightarrow \infty$). We see that as one would expect, both the QFI and direct imaging FI go to N/σ^2 when θ_2 approaches zero, for any n as well as the continuous line case.

As discussed above for two sources, even for $n \geq 2$ sources, as θ_2 increases from 0, the QFI and direct imaging FI fall from N/σ^2 due to the diffraction-induced overlap among nearby point sources, and hence our ability to estimate the centroid decreases. However, the performance of direct imaging falls more rapidly than the QFI and we see a small gap between the two. For any finite n , as θ_2 increases to the extent that the n point sources no longer significantly overlap, they essentially all become totally separate point sources and their locations can be estimated individually as totally separate single emitters, just as we argued for two points, in which regime the QFI is attainable with direct imaging. As a result, the QFI and direct imaging FI rise back towards the N/σ^2 value for a single emitter as θ_2 becomes sufficiently large. However, as n the number of point sources keeps increasing, θ_2 must increase further for the points to become “totally separate.” Therefore, we see that the QFI for three sources has a minimum at a larger θ_2 compared to that for the two-source case before it starts increasing again; the direct imaging FI behaves the same way. Further increasing the number of point sources augments this effect, with QFI and direct imaging FI having their minima at even larger values of θ_2 and requiring more and more separation for the QFI and FI to rise back toward the respective values for totally separated points. For a continuous line source, i.e., $n = \infty$, since there is an infinite number of points next to each other, the QFI and direct imaging FI both monotonically decrease as we increase the length because the constituent point sources comprising the

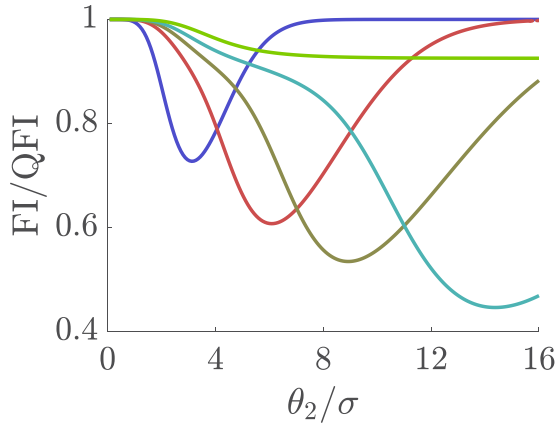


FIG. 3. Ratio between the QFI and the direct imaging FI for estimating the centroid of n point sources (dark to light: blue, $n = 2$; gold: $n = 3$; red, $n = 4$; cyan, $n = 6$; and green, continuous line) with end-to-end separation θ_2 given a Gaussian PSF with width σ .

uniformly radiant object can never be totally separated. In this case, increasing θ_2 only makes it more and more difficult to estimate the location of the centroid for a given total mean integrated photon number N .

In fact, it turns out that both the QFI and direct imaging FI for the continuous line scale as $1/\theta_2$ for large θ_2 . This can be seen as follows. Instead of assuming a fixed total number of photons N , let us consider the case when the total brightness of the object is proportional to its length. In other words, the total number of photons is $N\theta_2$, which amounts to simply multiplying the QFI and direct imaging FI values for N photons by θ_2 . These results are shown in Fig. 2(b) as solid and dashed curves, respectively. We see that they asymptote to constant values of about $1.95N/\sigma^2$ and $1.80N/\sigma^2$ for the QFI and direct imaging FI, respectively, which means that (1) the scaling behavior for constant total brightness (that does not scale with length) is indeed $1/\theta_2$ and (2) there is a constant-factor gap between direct imaging FI with QFI in the large- θ_2 limit. It is also possible to see this scaling behavior analytically, even though we cannot carry out the full calculations for the QFI and direct imaging FI analytically and have to resort to numerical methods. We describe this in Appendix E, where we outline the calculation for the continuum case where the sums over the infinite number of emitters, as in (2) and (5), are replaced by integrals.

To get a clearer picture about how the performance of direct imaging compares with the quantum-optimal measurement for estimating the centroid, we consider the ratio of the direct imaging FI to the QFI. We plot this ratio against the end-to-end separation θ_2 for a different number of point sources as well as for the continuous line in Fig. 3. We see that overall the ratio is generally not too low, though in some places it goes down below 50%. The minima are about 73%, 60%, 53%, and 45% for $n = 2, 3, 4$, and 6 points, respectively. For the continuous line, the ratio continuously falls, asymptoting to a constant value of around 92.5%.

All these results mean that while direct imaging is suboptimal, it is not significantly worse than the QFI. Therefore, it is not possible to get a significant improvement over direct imaging by using any other measurement. However, the fact

that it falls down to about 72% for two point sources and even slightly below 50% for a few more points suggests that there is some room for improvement, especially in a situation where simply collecting more photons to get the same improvement is not the most desirable option. It is also worth mentioning that since the gap between direct imaging and QFI is generally smaller for the continuum case than for a small finite number of emitters, direct imaging is closer to the optimal scheme for locating a full uniformly bright object rather than one that has internal structure and/or sparsity. Finally, these results and our methods outlined above apply to any two- or three-dimensional point-source constellations, and it is possible that the gap between QFI and direct-imaging FI is higher for other more general constellations of point emitters or continuous objects.

B. Comparison with the HG SPADE performance and an interesting duality

We now consider an image-plane HG mode SPADE measurement [8,24] but for centroid estimation. Let us take the central position of this SPADE to be $x = 0$ such that the optical axis of the SPADE defines the one-dimensional Cartesian coordinate system. The question we will consider is whether this can allow us to determine the centroid θ_1 more efficiently than direct imaging. Let us begin with writing the quantum state of a single temporal mode of the collected image-plane field with exactly one photon in the q th HG mode. In other words, a single-photon Fock state of the q th HG mode is given by

$$|\phi_q\rangle = \int_{-\infty}^{\infty} dx \phi_q(x)|x\rangle, \tag{24}$$

with $q = 0, 1, \dots$ and

$$\phi_q(x) = \frac{(2\pi\sigma^2)^{-1/4}}{\sqrt{2^q q!}} H_q\left(\frac{x}{\sqrt{2}\sigma}\right) \exp\left(-\frac{x^2}{4\sigma^2}\right), \tag{25}$$

where H_q are the Hermite polynomials and $|x\rangle$ is the unphysical one-photon Fock state of the perfectly localized (δ -function) spatial mode at position x . Now, recalling (2), it is straightforward to see that if we pass the image-plane field through an HG mode sorter and detect photons on each mode, the probability of finding the photon in mode q is

$$P(q) = \frac{1}{n} \sum_{s=1}^n P_s(q), \tag{26}$$

where $P_s(q)$ is the probability for a photon from source s to be found in mode q :

$$P_s(q) \equiv |\langle \phi_q | \psi_s \rangle|^2 = \left| \int_{-\infty}^{\infty} dx \phi_q(x) \psi(x - x_s) \right|^2. \tag{27}$$

For our Gaussian PSF defined in (4), this gives

$$P_s(q) = \exp(-Q_s) \frac{Q_s^q}{q!}, \tag{28}$$

where

$$Q_s = \frac{x_s^2}{4\sigma^2}. \tag{29}$$

It is now a straightforward exercise to obtain the Fisher information

$$\begin{aligned} \mathcal{J}_{\text{HG},11} &= \sum_{q=0}^{\infty} \frac{N}{P(q)} \left(\frac{\partial P(q)}{\partial \theta_1} \right)^2 \\ &= \sum_{q=0}^{\infty} \frac{N}{n\sigma^2 q! \sum_i e^{-Q_i} Q_i^q} \left(\sum_i Q_i (qQ_i^{q-1} - Q_i^q) e^{-2Q_i} \right. \\ &\quad \left. + \sum_{i \neq j} 2Q_{ij} (qQ_i^{q-1} - Q_i^q) (qQ_j^{q-1} - Q_j^q) e^{-Q_i - Q_j} \right), \end{aligned} \quad (30)$$

where $Q_{ij} \equiv x_i x_j / 4\sigma^2 = \sqrt{Q_i Q_j}$.

For the continuous line case of the number of points becoming infinite, we replace the sum in (26) by an integral. We show in Appendix F that

$$\mathcal{J}_{\text{HG},l,11} = N \sum_{q=0}^{\infty} \frac{(\exp(-Q_+) \frac{Q_+^q}{q!} - \exp(-Q_-) \frac{Q_-^q}{q!})^2}{\theta_2 \int_{y=-\theta_2/2}^{\theta_2/2} \exp[-Q(y)] \frac{Q(y)^q}{q!} dy}, \quad (31)$$

where

$$Q_{\pm} = \frac{(\theta_1 \pm \theta_2/2)^2}{4\sigma^2}. \quad (32)$$

Note that this clearly has a dependence on θ_2 other than the $1/\theta_2$ factor and therefore does not obey the same type of scaling behavior for large θ_2 that we found for the QFI and the direct imaging FI.

It is not clear how to do the sum over the FI contributions for the individual HG modes in (30) or (31) analytically, so we have to do this numerically. However, the series sum does simplify nicely for a few special cases.

First is the case when θ_2 approaches zero but θ_1 does not. This is essentially the limiting case where all the emitters effectively merge into a single one, but the SPADE is misaligned with the centroid by a constant amount. For this, the FI approaches the N/σ^2 value, the QFI for a single emitter as discussed in Appendix B 3. However then, for this $\theta_2 \rightarrow 0$ case, we also know that the performance of direct imaging approaches the QFI [7].

Second is the case when θ_1 approaches zero but θ_2 does not. This is the case when the SPADE is almost perfectly aligned with the centroid and the end-to-end separation is a constant. In this case, we get zero for the FI. This means that the HG SPADE yields tending-to-zero information about the centroid as the SPADE's alignment with the true centroid approaches near perfect.

To see this, recall (26), (28), and (29) and consider the partial derivative of $P(q)$:

$$\frac{\partial P(q)}{\partial \theta_1} = \frac{1}{n} \sum_{s=1}^n \frac{\partial P_s(q)}{\partial \theta_1} = \sum_{s=1}^n \frac{x_s}{2n\sigma^2} \exp(-Q_s) \left(\frac{qQ_s^{q-1}}{q!} - \frac{Q_s^q}{q!} \right). \quad (33)$$

Now, if we have an even number of emitters, then all their locations come in pairs of the form $x_s = \theta_1 \pm |c_s| \theta_2/2$, where c_s is a constant factor whose exact value depends on s . When $\theta_1 = 0$, these become $x_s = \pm |c_s| \theta_2/2$ and the corresponding Q values (which are proportional to x_s^2) are then equal for each

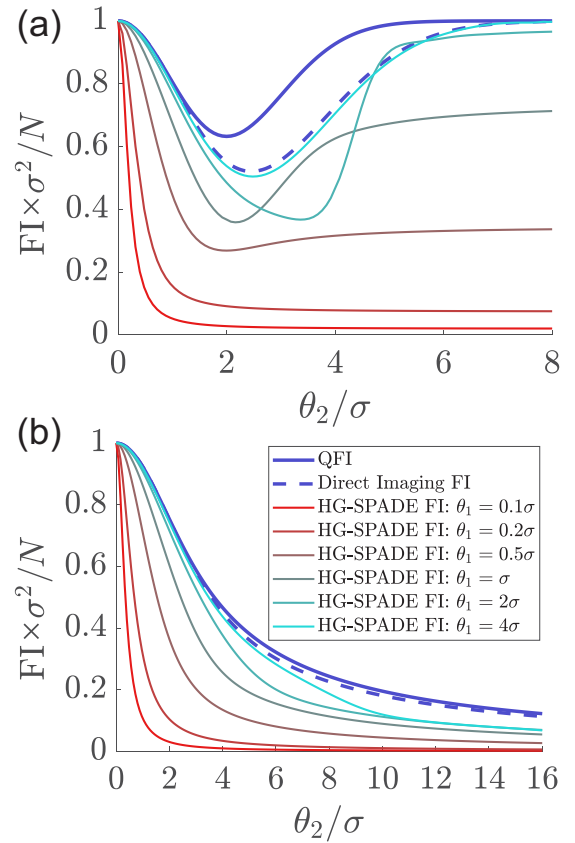


FIG. 4. (a) QFI, direct detection FI, and HG SPADE FI for estimating the centroid of two point sources with separation θ_2 and a centroid of θ_1 from the origin (i.e., the SPADE alignment axis), where red to cyan (dark to light) coloring indicates the shift from low misalignment to high misalignment. (b) Same as in (a) but for a continuous line source.

pair. The sum in (33) then becomes zero since the contributions from the two points in each pair cancel due to the x_s in front. If we have an odd number of sources, then all of them are in similar pairs, except the middle one at $x = \theta_1$. However, this becomes zero for $\theta_1 = 0$ and therefore we still get zero for the sum in (33).

This result has important implications for the two-stage setup of the kind being proposed in Ref. [15], in which we first obtain an estimate of the centroid from direct imaging in order to align an HG mode sorter in the second stage for estimating the separation or object size. Once we switch to the HG SPADE with a reasonably decent alignment close to the centroid, we will not be able to get any improvement in our estimate for the centroid. Our entire estimation precision of the centroid will therefore be based on the first measurement stage alone.

We now compare the performance of the HG SPADE with direct imaging and the QFI for centroid estimation. For this purpose, we focus on the two-emitter and the continuum cases as our two examples that illustrate the overall pattern. For two sources, the plots of the QFI, the direct imaging FI, and the HG SPADE for different fixed values of the misalignment are given in Fig. 4(a). The same comparison for the continuum case is shown in Fig. 4(b). The QFI and direct imaging curves in these figures are of course the same as

those shown in Fig. 2(a), and for the two-emitter case, the QFI and direct imaging curves are the same as those shown in Ref. [8], but that reference does not compare these with the HG SPADE's FI. We see that the performance of the HG SPADE is mostly worse than direct imaging for these graphs. It does however tend to converge with direct imaging from below in the $\theta_2 \rightarrow \infty$ limit. However, it is never higher than the direct imaging curve, except in a very small region for the $\theta_1 = 2\sigma$ curve for the two-emitter case. This is a special region between the $\theta_2 \rightarrow 0$ and large θ_2 extremes where the direct imaging FI drops sufficiently and the FI for the HG SPADE rises enough to achieve a very small amount of superiority. It is worth noting that this is a region where the HG SPADE has enough misalignment with the centroid and is also pointed sufficiently away from either of the two emitters. We can also find some other such regions where this happens and the SPADE performs better than direct imaging, but the improvement is very small, and for most of the parameter space, the latter outperforms the former by a bigger margin.

Therefore, the overall conclusion is that for all practical purposes, direct imaging is better than the HG SPADE for estimating the centroid and there is a small gap between the HG SPADE's performance and the ultimate quantum limit in the intermediary range of the object's length.

This highlights an interesting duality and complementarity between direct imaging and the HG SPADE. Unless the separation is large, direct imaging performs poorly for estimating the separation, especially in the sub-Rayleigh regime, whereas the HG SPADE, if aligned perfectly with the centroid, allows us to attain the QFI. However, the SPADE generally does not perform very well for estimating the centroid, whereas direct imaging comes closer to attaining the QFI. We should however qualify this statement by pointing out that this complementarity is not totally perfect; it certainly holds when the SPADE points exactly at the centroid, but there are regions of the parameter space for nonzero alignment where this relationship is no longer true. The small region we saw in the two-emitter case where the SPADE with a misalignment from the centroid of 2σ slightly outperforms direct imaging is an example of this. There is also a duality in the sense that a perfectly aligned SPADE attains the QFI for separation estimation but completely fails in determining the centroid, whereas a misaligned SPADE generally tends to be better for estimating the centroid rather than the separation.

C. Quantum-optimal measurement scheme for estimating the centroid

Having found that direct imaging is not optimal for centroid estimation in general and the HG SPADE's performance is mostly worse than direct imaging, we now discuss a scheme for surpassing it and asymptotically attaining the QFI. In general, the projective quantum measurement given by the eigenbasis of the SLD for a given quantum state defines a physically allowed measurement that achieves the QFI [25–27]. This SLD measurement would translate to a SPADE (not the HG SPADE), followed by photon detection on the sorted spatial modes. This is because any projective measurement on a quantum state of one photon in many (spatial) modes, which is the case for the quantum description of the state of a single temporal mode of collected light in our

problem, is always realizable by a passive linear optical transformation followed by photon detection.

However, since in this case the SLD depends on the true value of the centroid itself, this eigenbasis, and hence the aforesaid quantum-optimal SPADE to detect each temporal mode of the collected field, also depends upon the centroid. Therefore, we cannot carry out a measurement in this basis unless we already know the centroid, which is the variable we are trying to estimate in the first place. To get around this problem, we use a two-stage adaptive measurement scheme proposed in [26], applied to our problem.

(1) Recall that $N = M\epsilon$ is the mean photon number of the total collected field, where M is the number of temporal modes and $\epsilon \ll 1$ is the mean photon number per mode. The receiver's first stage uses a small proportion of the integration time, worth N^α mean photon number, with any $0 < \alpha < 1$, to obtain an initial maximum-likelihood estimate $\hat{\theta}_1$ for the unknown centroid parameter θ_1 . The measurement used for this stage can be direct imaging and does not have to be an optimal choice in any sense. The only requirement on this measurement is that it has a nonzero FI for estimating the centroid θ_1 .

(2) Based on the estimate $\hat{\theta}_1$ obtained during the first stage, we carry out a measurement on each temporal mode of the remaining collected photons (of mean photon number $N - N^\alpha$) using the eigenbasis of the SLD of θ_1 , evaluated at $\hat{\theta}_1$. Based on this measurement, we obtain a maximum-likelihood estimate for θ_1 .

We would like to mention that using any other QFI-attaining measurement in place of the SLD eigenbasis in the second stage will give the same performance in theory. Since the SLD basis is not necessarily the only measurement that attains the QFI, there may be other alternatives too; for example, the linear interferometric approach put forward in [28] is worth investigating for this purpose.

The above procedure prescribes a measurement that asymptotically reaches the efficiency of the QFI as N tends to infinity. This two-stage scheme is also described in [29–31], with the specific choice of $\alpha = \frac{1}{2}$ and with the condition that the FI for the first stage should be nonzero.

While we refer the reader to the above references for a detailed derivation of why this two-stage approach should attain the efficiency of the QFI in the large- N limit even though the SLD depends upon the *a priori*-unknown parameter, here is a short summary of the argument. If N is sufficiently large, N^α or \sqrt{N} in particular will also be large. Therefore, the variance of the estimate $\hat{\theta}_1$ scales as $1/\mathcal{J}_1 N^\alpha$, where $\mathcal{J}_1 > 0$ is the FI of this stage-1 measurement, hence approaching zero as $N \rightarrow \infty$, around the true (*a priori*-unknown) value of θ_1 . Now the Fisher information in stage 2 should be $(N - N^\alpha)\mathcal{K}$ if we measure in the eigenbasis of the SLD based on the exactly true value of θ_1 . However, in reality, since we will carry out this measurement at the estimated value $\hat{\theta}_1 = \theta_1 + \theta_{1,\text{err}}$, we must replace \mathcal{K} by the FI for the SLD eigenbasis measurement evaluated at this value rather than the true θ_1 . We can express this FI as a Taylor expansion around the true value θ_1 as

$$\mathcal{J}_{\text{SLD}}(\theta_1 + \theta_{1,\text{err}}) = \mathcal{J}_{\text{SLD}}(\theta_1) + \theta_{1,\text{err}}^2 \frac{\partial^2 \mathcal{J}_{\text{SLD}}(\theta_1)}{\partial \theta_1^2} + \dots \tag{34}$$

Here we do not have a first derivative term because \mathcal{J}_{SLD} has a maximum at θ_1 equal to \mathcal{K} , so the first derivative must be zero. Moreover, since this is a maximum, the second derivative will be a negative constant with respect to $\theta_{1,\text{err}}$. Therefore, we can rewrite the stage-2 FI as

$$\mathcal{J}_{\text{SLD}}(\theta_1 + \theta_{1,\text{err}}) = \mathcal{K}(1 - O(\theta_{1,\text{err}}^2)) = \mathcal{K}\left(1 - O\left(\frac{1}{N^\alpha \mathcal{J}_1}\right)\right), \quad (35)$$

where in the last step we have used the fact that the mean-square error of the initial centroid estimate is approximately equal to the inverse of the FI for stage 1. The total FI accumulated over stage 2 is therefore $(N - N^\alpha)\mathcal{K}(1 - O(\frac{1}{N^\alpha \mathcal{J}_1}))$. Furthermore, since when N is large

$$\frac{1}{(N - N^\alpha)\mathcal{K}(1 - O(\frac{1}{N^\alpha \mathcal{J}_1}))} \approx \frac{1}{N\mathcal{K}}, \quad (36)$$

the variance approaches that of the optimal measurement. Based on the choice of the stage-1 measurement and its FI \mathcal{J}_1 , one could optimize the choice of α such that the overall FI attained at the end of stage 2 is maximized for a given fixed N . A multistage adaptive quantum estimation algorithm is given in [32] where the result of each stage is used as input for the next one, which could lead to further improved performance in this nonasymptotic setting. However, finding the quantum-optimal measurement for finite N is being left for future work. Note that in the context of this adaptive measurement, N indicates the number of photons that are dedicated to estimating the centroid. If the ultimate goal is to estimate the separation, there also will be photons set aside for the estimation of the separation (e.g., using a SPADE aligned to the centroid estimate) [15].

We can therefore apply this two-step procedure for estimating the centroid if we can calculate the eigenvectors of the SLD. For two point sources of light, the nonzero entries of the SLD are given in Appendix C. For more than two light sources, including the case of a continuous line, we can obtain the SLD numerically, calculating it in the basis of HG-basis one-photon Fock states, as described in Appendix D. It is important to note here that the SLDs depend not only on the centroid, but also on the separation (or the object length, in the case of a continuous line). Therefore, if we already know the separation, then we only need to estimate θ_1 in stage 1 of our two-step adaptive scheme. However, if we do not know the separation, then we also need to extract an initial estimate $\hat{\theta}_2$ for the separation from the measurement in the first stage. We can then switch to the second stage where we calculate the eigenbasis of the SLD in terms of $\hat{\theta}_1$ and $\hat{\theta}_2$ to get a good estimate for the centroid whose quality approaches the QFI for a large number of integrated photons.

IV. CONCLUSION

We have carried out a detailed systematic study of our ability to estimate the centroid of a linear array of incoherent light sources as well as a line-shaped object with uniform brightness. Our approach is easily extensible to estimating the centroid of more complex objects. We calculated the QFI for estimating the centroid and compared it with the FI for direct imaging as well as an image-plane HG SPADE. We described

a two-stage readily realizable measurement that would attain the QFI of centroid estimation. Our key conclusions can be summarized as follows.

Direct imaging in the image plane is strictly suboptimal compared to the QFI for centroid estimation, though the ratio of the QFI to the direct imaging FI is less than an order of magnitude. Therefore, direct imaging should offer a fairly good estimate of the location of an object, as intuitively expected.

The gap between the performance of direct imaging and QFI is generally less for the continuum objects than for a constellation involving a small number of pointlike emitters. This suggests that direct imaging performs closer to the optimal scheme for locating a full uniformly bright object rather than one that has more internal features.

The performance of the HG SPADE is mostly worse than direct imaging for centroid estimation, except for some special limiting regimes where its FI approaches the direct imaging FI from below or some small regions of parameter space where it marginally surpasses direct imaging. However, these regions where it slightly outperforms direct imaging are negligible, and the performance improvement is also too little to be of any practical significance.

We have also found that the HG mode performs very poorly when it is nearly aligned with the centroid. This means that if we employ a two-stage procedure for determining θ_1 and θ_2 , where we first estimate θ_1 and then use it to align the SPADE for determining θ_2 in stage 2 as in [15], then we would not be able to extract much additional information about the centroid from the SPADE measurement in stage 2.

There is a complementarity between direct imaging and the HG SPADE. Direct imaging has a fairly good performance for centroid estimation, even though it is not the optimal measurement. However, it performs very poorly for determining the separation in the sub-Rayleigh regime, and the Fisher information for the separation goes to zero when the separation approaches zero. The HG SPADE, on the other hand, is the optimal measurement for finding the separation, but it performs very poorly for determining the centroid, when it is nearly aligned with the centroid.

We have found an interesting scaling behavior for the QFI and direct imaging for the continuous line for large θ_2 , with a constant-factor gap. Specifically, we have found that the QFI and the direct imaging FI both scale as $1/\theta_2$ in this region. This makes very good intuitive sense: The larger the length of a continuous object, the greater the portion of the object that has spatially constant irradiance, and hence fewer information-bearing photons are available to estimate the centroid.

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APPENDIX A: DIAGONALITY OF THE FI AND QFI FOR OUR CENTROID AND SEPARATION PARAMETERS

1. The FI for direct imaging

First, consider direct imaging. For our choice of a Gaussian PSF, the density probability function $\Lambda(x)$ defined in (5) is an even function around the centroid $x = \theta_1$. It is a straightforward exercise to see that $\frac{\partial \Lambda(x)}{\partial \theta_1}$ is an odd function around the centroid, whereas $\frac{\partial \Lambda(x)}{\partial \theta_2}$ is even. Their product is therefore an odd function around θ_1 , and the integral over x from $-\infty$ to ∞ is therefore zero.

To see why $\partial \Lambda(x)/\partial \theta_1$ is odd and $\partial \Lambda(x)/\partial \theta_2$ is even, note that if we have an even number of points, they come in pairs of the form $x_s = \theta_1 \pm c_s \theta_2$, in which c_s is a factor that only depends on s . If we have an odd number of points, then we have a point in the middle at $x = \theta_1$ and all the other points again come in such pairs with the same distance on either side of the centroid. The partial derivative of $\Lambda(x)$ will therefore also contain pairs with contributions of the form

$$\begin{aligned} & \frac{\partial |\psi(x - \theta_1 \pm c_s \theta_2)|^2}{\partial \theta_1} \\ &= -2 |\psi(x - \theta_1 \pm c_s \theta_2)| \frac{\partial |\psi(x - \theta_1 \pm c_s \theta_2)|}{\partial (x - \theta_1 \pm c_s \theta_2)}. \end{aligned} \quad (\text{A1})$$

Now, since $|\psi(x - \theta_1 \pm c_s \theta_2)|$ is symmetric around $\theta_1 \mp c_s$, $\frac{\partial |\psi(x - \theta_1 \pm c_s \theta_2)|}{\partial (x - \theta_1 \pm c_s \theta_2)}$ is antisymmetric. Therefore, the sum over both elements of the pair with $\pm c_s \theta_2$ is an antisymmetric function around θ_1 . In case the number of points is odd, then the partial derivative of the middle point will also be an antisymmetric function. So $\partial \Lambda(x)/\partial \theta_1$ is odd around the centroid.

In contrast,

$$\begin{aligned} & \frac{|\psi(x - \theta_1 \pm \theta_2)|^2}{\partial \theta_2} \\ &= \pm 2c_s |\psi(x - \theta_1 \pm c_s \theta_2)| \frac{\partial |\psi(x - \theta_1 \pm c_s \theta_2)|}{\partial (x - \theta_1 \pm c_s \theta_2)}. \end{aligned} \quad (\text{A2})$$

So now the two members of the pair have opposite signs. This, along with the fact that $\frac{\partial |\psi(x - \theta_1 \pm c_s \theta_2)|}{\partial (x - \theta_1 \pm c_s \theta_2)}$ is antisymmetric around $\theta_1 \mp c_s \theta_2$, means that the two members of each pair combine to give a symmetric function around θ_1 . If the total number of points is odd, then there is also an additional point located in the middle at the centroid. However, since its position does not depend on θ_2 , it does not contribute to $\partial \Lambda(x)/\partial \theta_2$. Therefore, we conclude that $\partial \Lambda(x)/\partial \theta_2$ is an even function around θ_1 .

2. The QFI

For the QFI, the argument is somewhat similar, but now we need to think in terms of the density operator and its partial derivatives. Recalling (2), the one-photon part of the density operator is

$$\rho_1 = \frac{1}{n} \sum_{s=1}^n |\psi_s\rangle \langle \psi_s| = \frac{1}{n} \sum_{s=1}^n |\psi(x - x_s)\rangle \langle \psi(x - x_s)|. \quad (\text{A3})$$

This again comes in pairs, and the density operator is symmetric in each pair. Now consider the partial derivative

$$\begin{aligned} \frac{\partial \rho_1}{\partial \theta_1} &= \frac{1}{n} \sum_{s=1}^n \left(\frac{\partial |\psi(x - x_s)\rangle}{\partial \theta_1} \langle \psi(x - x_s)| \right. \\ &\quad \left. + |\psi(x - x_s)\rangle \frac{\partial \langle \psi(x - x_s)|}{\partial \theta_1} \right). \end{aligned} \quad (\text{A4})$$

However, $|\psi(x - x_s)\rangle = \int dx \psi(x - x_s) |x\rangle$ and therefore it has the same even-odd parity as $\psi(x - x_s)$. Likewise, the partial derivatives of this ket will have the same even-odd parity as the partial derivatives of the function $\psi(x - x_s)$. Therefore, like $\partial \Lambda(x)/\partial \theta_1$ and $\partial \Lambda(x)/\partial \theta_2$, $\partial \rho_1/\partial \theta_1$ and $\partial \rho_1/\partial \theta_2$ will be odd and even, respectively, and therefore the off-diagonal entries of the QFI evaluated in terms of the parameters θ_1 and θ_2 will be zero.

3. The HG SPADE

For the HG mode sorter, our physical setup is not symmetric unless the SPADE is perfectly aligned with the centroid. Therefore, the above-mentioned symmetry arguments no longer hold and the FI for a measurement in the HG basis is not diagonal in general. For the particular case of the SPADE being perfectly aligned with the centroid, we saw in Sec. III B that $\partial P(q)/\partial \theta_1$ becomes zero due to cancellations of the contributions from each member of the symmetric pair. Therefore, not only is the FI diagonal, but all its entries other than the diagonal one corresponding to θ_2 are zero.

However, when the SPADE is misaligned, we will not get a diagonal QFI matrix in general. To see this, consider the specific example of the two-point case. The probability function is

$$P(q) = \frac{1}{2q!} [\exp(-Q_1) Q_1^q + \exp(-Q_2) Q_2^q], \quad (\text{A5})$$

where $Q_1 = \frac{(\theta_1 + \theta_2/2)^2}{4\sigma^2}$ and $Q_2 = \frac{(\theta_1 - \theta_2/2)^2}{4\sigma^2}$. The two terms are clearly not symmetric or antisymmetric, since they have different Gaussian decay factors as well as $(\theta_1 \pm \theta_2)^q$, which will be different for both points.

APPENDIX B: THE SINGLE-POINT CASE

1. The FI for direct imaging

For a single point, the probability function $\Lambda(x)$ defined in (5) becomes

$$\Lambda_1(x) = |\psi(x - \theta_1)|^2. \quad (\text{B1})$$

The Fisher information for direct imaging is then

$$\begin{aligned} \mathcal{J}_{1\text{-pt}} &= \int dx \frac{N}{\Lambda(x)} \left(\frac{\partial \Lambda(x)}{\partial \theta_1} \right)^2 \\ &= 4N \int dx \left(\frac{\partial |\psi(x - \theta_1)|}{\partial \theta_1} \right)^2 \\ &= 4N \Delta k^2. \end{aligned} \quad (\text{B2})$$

For our Gaussian PSF defined in (4), the result is

$$\mathcal{J}_{1\text{-pt}} = \frac{N}{\sigma^2}. \quad (\text{B3})$$

2. The QFI

When we only have one point, the single-photon part of the density matrix is simply one dimensional

$$\rho_1 = |\psi(x - \theta_1)\rangle\langle\psi(x - \theta_1)|. \quad (\text{B4})$$

The partial derivative of this is

$$\begin{aligned} \frac{\partial \rho_1}{\partial \theta_1} &= \frac{\partial |\psi(x - \theta_1)\rangle}{\partial \theta_1} \langle\psi(x - \theta_1)| \\ &+ |\psi(x - \theta_1)\rangle \frac{\partial \langle\psi(x - \theta_1)|}{\partial \theta_1} \\ &= -|\psi'(x - \theta_1)\rangle\langle\psi(x - \theta_1)| - |\psi(x - \theta_1)\rangle\langle\psi'(x - \theta_1)| \end{aligned} \quad (\text{B5})$$

since $\partial \psi(x - \theta_1)/\partial \theta_1 = -\psi'(x - \theta_1)$, where ψ' is the derivative of ψ . It is straightforward to see that $|\psi(x - \theta_1)\rangle$ and $|\psi'(x - \theta_1)\rangle$ are orthogonal states for any symmetric $\psi(x)$:

$$\langle\psi(x - \theta_1)|\psi'(x - \theta_1)\rangle = \int dx \psi(x - \theta_1) \frac{\partial \psi(x - \theta_1)}{\partial \theta_1} = 0. \quad (\text{B6})$$

We thus have an orthogonal basis and only need to normalize $\partial |\psi(x - \theta_1)\rangle/\partial \theta_1$. Our orthonormal basis is thus

$$|e_1\rangle \equiv |\psi(x - \theta_1)\rangle, \quad (\text{B7})$$

$$|e_2\rangle \equiv \frac{1}{\Delta k} |\psi'(x - \theta_1)\rangle, \quad (\text{B8})$$

where $\Delta k = \sqrt{\int dx |\psi'(x - \theta_1)|^2}$ is a normalization factor and is equal to the square root of Δk^2 , defined in (17). The density operator in this eigenbasis is simply $\rho_1 = |e_1\rangle\langle e_1|$ with eigenvalues $D_1 = 1$ and $D_2 = 0$ corresponding to $|e_1\rangle$ and $|e_2\rangle$. Recalling (12) and (B5), the symmetric logarithmic derivative is then

$$\mathcal{L} = 2\Delta k(|e_2\rangle\langle e_1| + |e_1\rangle\langle e_2|). \quad (\text{B9})$$

The QFI is then

$$\mathcal{K}_{1\text{-pt}} = N \text{tr}(\rho \mathcal{L}^2) = 4N\Delta k^2. \quad (\text{B10})$$

We see that this is equal to the FI for direct imaging in (B2) for any PSF.

3. The HG SPADE FI

The probability function for the q th HG mode for light coming from a single point with a Gaussian PSF is given by

$$P_q = \exp(-Q) \frac{Q^q}{q!}, \quad (\text{B11})$$

where $Q = \frac{\theta_1^2}{4\sigma^2}$. The FI for the q th mode is then

$$\mathcal{J}_{q,\text{HG},1\text{-pt}} = \frac{N}{P_q} \left(\frac{\partial P_q}{\partial \theta_1} \right)^2. \quad (\text{B12})$$

It is a straightforward exercise to calculate this and carry out the sum over the whole series in q ; the result is

$$\mathcal{J}_{\text{HG},1\text{-pt}} = \frac{N}{\sigma^2}, \quad (\text{B13})$$

which is equal to the QFI as well as the FI for direct imaging.

APPENDIX C: THE QFI FOR TWO POINTS

This calculation has been explained by Tsang *et al.* in their paper. Therefore, we will only summarize their method while clarifying one or two points.

From (2), the single-photon part of the density matrix for the two-point case is

$$\rho_1 = \frac{1}{2}(|\psi_1\rangle\langle\psi_1| + |\psi_2\rangle\langle\psi_2|). \quad (\text{C1})$$

However, for the QFI, we need to work in an orthonormal eigenbasis that spans the whole space spanned by $|\psi_1\rangle$ and $|\psi_2\rangle$ as well as their partial derivatives with respect to θ_1 . It turns out that while $|\psi_1\rangle$ and $|\psi_2\rangle$ individually have norm-1, they are not mutually orthogonal in general

$$\delta \equiv \langle\psi_1|\psi_2\rangle = \langle\psi_2|\psi_1\rangle \neq 0 \quad (\text{C2})$$

for a real valued $\psi(x)$. Therefore, we first need to express ρ_1 in an orthonormal basis

$$\rho_1 = D_1|e_1\rangle\langle e_1| + D_2|e_2\rangle\langle e_2|. \quad (\text{C3})$$

To find the eigenvalues D_i and the eigenstates $|\psi_i\rangle$, we write down a 2×2 matrix of the inner products $\langle\psi_i|\rho|\psi_j\rangle$,

$$\begin{pmatrix} 1 & \delta \\ \delta & 1 \end{pmatrix}. \quad (\text{C4})$$

The normalized eigenvectors of this matrix give us an orthogonal set of functions, and the square roots of the eigenvalues give us the normalization factors. We find that the eigenvalues are $1 \pm \delta$, with the eigenvectors $\frac{1}{\sqrt{2}}(1, \pm 1)$. Therefore, our orthonormal basis of states spanning $|\psi_1\rangle$ and $|\psi_2\rangle$ is

$$|e_1\rangle = \frac{1}{\sqrt{2(1-\delta)}}(|\psi_1\rangle - |\psi_2\rangle), \quad (\text{C5})$$

$$|e_2\rangle = \frac{1}{\sqrt{2(1+\delta)}}(|\psi_1\rangle + |\psi_2\rangle). \quad (\text{C6})$$

It is a straightforward exercise to see that these are also the eigenstates of our density operator ρ_1 and that the eigenvalues D_i of ρ_1 are the corresponding eigenvalues of the matrix of inner products (C4) divided by 2:

$$D_1 = \frac{1-\delta}{2}, \quad (\text{C7})$$

$$D_2 = \frac{1+\delta}{2}. \quad (\text{C8})$$

This division by 2 is simply the $\frac{1}{2}$ factor in front of $|\psi_1\rangle\langle\psi_1| + |\psi_2\rangle\langle\psi_2|$.

However, $\frac{\partial \rho_1}{\partial \theta_1}$ also contains the derivatives of $|\psi_1\rangle$ and $|\psi_2\rangle$. Therefore, we need to extend our eigenbasis to span these states too. We therefore include the derivatives of $|\psi_i\rangle$ and carry out an orthogonalization procedure. This gives us the additional states

$$|e_3\rangle = \frac{1}{c_3} \left[\frac{\Delta k}{\sqrt{2}}(|\psi_{11}\rangle + |\psi_{22}\rangle) - \frac{\gamma}{\sqrt{1-\delta}}|e_1\rangle \right], \quad (\text{C9})$$

$$|e_4\rangle = \frac{1}{c_4} \left[\frac{\Delta k}{\sqrt{2}}(|\psi_{11}\rangle - |\psi_{22}\rangle) + \frac{\gamma}{\sqrt{1+\delta}}|e_2\rangle \right], \quad (\text{C10})$$

where Δk^2 and γ were defined in (17) and (19). The other quantities defined here are

$$|\psi_{11}\rangle \equiv \frac{1}{\Delta k} \int dx \frac{\partial \psi(x-x_1)}{\partial x_1} |x\rangle, \quad (\text{C11})$$

$$|\psi_{22}\rangle \equiv \frac{1}{\Delta k} \int dx \frac{\partial \psi(x-x_2)}{\partial x_2} |x\rangle, \quad (\text{C12})$$

$$c_3 \equiv \left(\Delta k^2 + b^2 - \frac{\gamma^2}{1-\delta} \right)^{1/2}, \quad (\text{C13})$$

$$c_4 \equiv \left(\Delta k^2 - b^2 - \frac{\gamma^2}{1+\delta} \right)^{1/2}, \quad (\text{C14})$$

$$b^2 \equiv \int dx \frac{\partial \psi(x-x_1)}{\partial x_1} \frac{\partial \psi(x-x_2)}{\partial x_2}, \quad (\text{C15})$$

and δ was defined in (C2).

Since $\rho_1 = D_1|e_1\rangle\langle e_1| + D_2|e_2\rangle\langle e_2|$, we get

$$D_3 = D_4 = 0. \quad (\text{C16})$$

Having found all the eigenbasis states and the eigenvalues of ρ_1 , it is now a simple exercise to use the formula (12) for the SLD and compute the QFI, with the results given in (C24) and (C25). For reference, the nonzero entries of the SLD are with respect to the centroid in the $|e_i\rangle$ ($i = 1, \dots, 4$) basis. The SLD with respect to the centroid has the nonzero entries are as follows:

$$\mathcal{L}_{1,12} = \frac{2\gamma\delta}{\sqrt{1-\delta^2}}, \quad (\text{C17})$$

$$\mathcal{L}_{1,14} = \frac{2c_4}{\sqrt{1-\delta}}, \quad (\text{C18})$$

$$\mathcal{L}_{1,23} = \frac{2c_3}{\sqrt{1+\delta}}. \quad (\text{C19})$$

The nonzero entries of the SLD with respect to the separation are

$$\mathcal{L}_{2,11} = -\frac{\gamma}{1-\delta}, \quad (\text{C20})$$

$$\mathcal{L}_{2,13} = -\frac{c_3}{\sqrt{1-\delta}}, \quad (\text{C21})$$

$$\mathcal{L}_{2,22} = \frac{\gamma}{1+\delta}, \quad (\text{C22})$$

$$\mathcal{L}_{2,24} = -\frac{c_4}{\sqrt{1+\delta}}. \quad (\text{C23})$$

From these it is a straightforward exercise to calculate the QFI. For the centroid we obtain

$$\mathcal{K}_{11} = 4N(\Delta k^2 - \gamma^2) \quad (\text{C24})$$

and for the separation

$$k_{22} = N\Delta k^2. \quad (\text{C25})$$

APPENDIX D: CALCULATING THE QFI FOR MORE THAN TWO POINTS

Calculating the QFI analytically for more than two points by employing the procedure of Tsang *et al.* for two points becomes a very complicated process, since it requires diagonalizing larger and larger matrices as the number of points is increased. Even doing this numerically is a very involved process, and in fact we soon start running into floating point

errors when we go to about ten or so points. It also does not allow us to calculate the QFI for a continuous line. We therefore follow the more efficient numerical approach employed in [11]. The idea is that instead of working with states $|\psi_i\rangle$ and their derivatives and carrying out a laborious diagonalization process, we work in the HG basis. We only need to consider the first few HG modes, since higher-order modes have diminishing contributions. The HG SPADE calculations in this paper have been carried out with 50 HG modes, and we have checked that this is more than enough for the results to converge for the parameters being considered.

Specifically, we express our states $\psi(x-x_s)$ in the HG basis, which gives

$$\psi(x-x_s) = \sum_{q=0}^{\infty} \exp\left(-\frac{x_s^2}{8\sigma^2}\right) \frac{x_s^q}{\sqrt{q!}} \phi_q(x), \quad (\text{D1})$$

where $\phi_q(x)$ are the HG functions. We then express our density matrix ρ_1 and its derivative $\partial\rho_1/\partial\theta_1$ in this basis. Since $\phi_q(x)$ does not depend on the location of the individual points or the centroid, the partial derivatives do not change the basis and therefore we do not have to carry out any orthogonalization procedure to find our additional basis states. We simply numerically calculate the eigenvectors and eigenvalues for the ρ_1 in the basis of the first 50 HG modes (or whatever other number we decide to consider). We then use the formula (12) to obtain the SLD and calculate the QFI from $N \text{tr}(\rho\mathcal{L}^2)$.

APPENDIX E: THE QFI AND DIRECT IMAGING FI FOR THE CONTINUUM CASE AND ITS SCALING BEHAVIOR

Here we show why the QFI and direct imaging FI for the centroid scale as $1/\theta_2$ for large θ_2 in the continuum case of an infinite number of emitters. First, consider direct imaging and recall the definition of the probability function $\Lambda(x)$ in (5). This is an average over m points, and for a continuous line we replace it by an integral

$$\Lambda_{\text{line}}(x) = \frac{\int_{y=-\theta_2/2}^{\theta_2/2} |\psi(x-\theta_1-y)|^2 dy}{\theta_2}. \quad (\text{E1})$$

For θ_2 sufficiently large compared to σ , this, being an integral of a sharply peaked but smooth function, should give a nearly flat function that is constant over the length of the line and zero elsewhere, but with smooth edges with width of order σ . For our Gaussian PSF we get

$$\Lambda_{\text{line}}(x) = \frac{1}{2\theta_2} \left[\text{erf}\left(\frac{x-\theta_1+\frac{\theta_2}{2}}{\sqrt{2}\sigma}\right) - \text{erf}\left(\frac{x-\theta_1-\frac{\theta_2}{2}}{\sqrt{2}\sigma}\right) \right], \quad (\text{E2})$$

which is indeed nearly $1/\theta_2$ over the line and almost zero elsewhere, but has smoothly falling edges. The other ingredient we need for the FI is the partial derivative of this with respect to θ_1 , which can easily be computed using the fundamental theorem of calculus. Since $\frac{\partial |\psi(x-\theta_1-y)|^2}{\partial \theta_1} = \frac{\partial |\psi(x-\theta_1-y)|^2}{\partial y}$, we

obtain $\frac{\partial \Lambda_{\text{line}}(x)}{\partial \theta_1}$ by removing the integral and evaluating $|\psi(x - \theta_1 - y)|^2$ at the y values of the end points:

$$\frac{\partial \Lambda_{\text{line}}(x)}{\partial \theta_1} = \frac{1}{\theta_2} \left[\left| \psi \left(x - \theta_1 - \frac{\theta_2}{2} \right) \right|^2 - \left| \psi \left(x - \theta_1 + \frac{\theta_2}{2} \right) \right|^2 \right]. \quad (\text{E3})$$

If θ_2 is sufficiently large compared to σ , this is the difference between two nonoverlapping sharply peaked but smooth functions at the edges of the line. The square of this will therefore be a sum of two even more sharply peaked functions at the edges of the line. The integral over x to evaluate FI therefore only gets noticeable contributions near the edges of the line, and hence the distance between the edges does not have any bearing on it. This leaves the $1/\theta_2$ factor in front as the main θ_2 dependent part in (E3). This, along with the $1/\theta_2$ scaling of $\Lambda(x)$ in (E2), means that the FI should scale as $1/\theta_2$. It is worth noting that this reasoning should hold equally for any other PSF that has a high peak in the center and quickly but smoothly falls to zero away from it.

A somewhat similar argument can be made for the scaling of the QFI in the large- θ_2 region. The sum in (2) for the density matrix gets replaced by an integral over the line. Recalling the definition of $|\psi_s\rangle$ in (3), this integral is

$$\rho_1 = \frac{1}{\theta_2} \int dx dx' \int_{y=-\theta_2/2}^{\theta_2/2} dy \psi(x - \theta_1 - y) \times \psi(x' - \theta_1 - y) |x\rangle \langle x'|. \quad (\text{E4})$$

The partial derivative of this with respect to θ_1 can again be obtained from the fundamental theorem of calculus by removing the integral over y and evaluating this at the end points:

$$\rho_1 = \int dx dx' \frac{1}{\theta_2} \left[\psi \left(x - \theta_1 - \frac{\theta_2}{2} \right) \psi \left(x' - \theta_1 - \frac{\theta_2}{2} \right) - \psi \left(x - \theta_1 + \frac{\theta_2}{2} \right) \psi \left(x' - \theta_1 + \frac{\theta_2}{2} \right) \right] |x\rangle \langle x'|. \quad (\text{E5})$$

For θ_2 sufficiently larger than σ , the two terms will sharply peak when both x and x' are simultaneously equal to the end point locations of the line at $\theta_1 + \theta_2/2$ and $\theta_1 - \theta_2/2$. Elsewhere they will be nearly zero. The height and width of these peaks will not depend on θ_2 , and therefore the scaling of $\partial \rho / \partial \theta_1$ in terms of θ_2 arises almost entirely from the $1/\theta_2$ factor in front.

As for the scaling of ρ , we can return to (E4). Again, focusing on θ_2 sufficiently larger than σ , we note that with $\psi(x - y)$ and $\psi(x' - y)$ sharply peaked at $y = x$ and $y = x'$, performing the integral over y will give us a function that

sharply peaks at $x = x'$ provided x and x' lie somewhere on our line between $-\theta_2/2$ and $\theta_2/2$. For our Gaussian point spread function, we get

$$\rho_1 = \frac{1}{\theta_2} \int dx dx' \frac{\exp\left(-\frac{(x-x')^2}{8\sigma^2}\right)}{2} \times \left[\text{erf}\left(\frac{\theta + x + x'}{2^{3/2}\sigma}\right) - \text{erf}\left(\frac{-\theta + x + x'}{2^{3/2}\sigma}\right) \right], \quad (\text{E6})$$

in which we indeed have a sharply peaked Gaussian involving $x - x'$ and the erf functions are simply step functions with smooth edges and hence do not contribute to the scaling in terms of θ_2 . This leaves the $1/\theta_2$ in front as the only factor that contributes to the scaling. Since the QFI is given as $K = \text{tr}(\rho \mathcal{L}^2)$ and satisfies the relation (11), it is clear that the overall scaling is essentially of two powers of $\partial \rho / \partial \theta_1$ and an inverse power of ρ , so overall we get $1/\theta_2$. Like the argument for the direct imaging FI, this reasoning for the scaling of the QFI should also hold for any other smooth PSF that has a sufficiently sharp peak.

APPENDIX F: THE HG SPADE FI FOR THE CONTINUUM CASE

For the continuum case, we replace the sum in (26) by an integral. Recalling (28), we get

$$P_l(q) = \frac{1}{\theta_2} \int_{y=-\theta_2/2}^{\theta_2/2} dy \exp[-Q(y)] \frac{Q(y)^q}{q!}, \quad (\text{F1})$$

where

$$Q(y) = \frac{\theta_1 + y}{4\sigma^2}. \quad (\text{F2})$$

In addition, the partial derivative with respect to θ_1 can be obtained, according to the fundamental theorem of calculus, by just removing the integral over y and evaluating at the end points,

$$\frac{\partial P_l(q)}{\partial \theta_1} = \frac{1}{\theta_2} \left(\exp(-Q_+) \frac{Q_+^q}{q!} - \exp(-Q_-) \frac{Q_-^q}{q!} \right), \quad (\text{F3})$$

where

$$Q_{\pm} = \frac{(\theta_1 \pm \frac{\theta_2}{2})^2}{4\sigma^2}. \quad (\text{F4})$$

It is now straightforward to write down the FI contribution for each HG mode, and summing over all the modes gives us the total FI

$$\mathcal{J}_{\text{HG},l,11} = N \sum_{q=0}^{\infty} \frac{\left(\exp(-Q_+) \frac{Q_+^q}{q!} - \exp(-Q_-) \frac{Q_-^q}{q!} \right)^2}{\theta_2 \int_{y=-\theta_2/2}^{\theta_2/2} dy \exp[-Q(y)] \frac{Q(y)^q}{q!}}. \quad (\text{F5})$$

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