

Introduction to Interferometry

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Radio interferometry is the technique of combining the response of multiple telescopes to detect radiation from astronomical sources. One of the main advantages of this technique is the resolution power of an interferometer. The angular resolution of any optical instrument can be approximated as $\theta \sim \frac{\lambda}{D}$. So for any given wavelength of observation, λ , we obtain higher resolution by increasing the size of the aperture of the observing instrument, D . However, this will eventually be limited by engineering constraints to ~ 100 metres for fully-steerable radio telescopes. With an interferometer, D , is the separation *between* the radio telescopes and can be as large as hundreds or even thousands of kilometres. In the following derivation, I will be providing a broad overview of the main concepts that interferometry is based upon.

1 Electromagnetic waves incident on an array

The overall scheme of radio interferometry is represented in Figure 1. The incoming radiation from a natural astronomical source \mathbf{S} can be considered as a noise signal. Over a small time, δt , the radiation can be considered to be sinusoidal. Such an oscillating electromagnetic wave with *quasi-monochromatic* frequency, ν , can be described as

$$\begin{aligned} E(t) &= A \cos(2\pi \nu t + \Phi) \\ &= A \cos(\omega t + \Phi), \quad \text{where } \omega = \frac{2\pi}{T} = 2\pi\nu \end{aligned} \quad (1)$$

Our aim is to mathematically determine the relationship which will ensure that we are measuring the same wavefront of $E(t)$ which is incident upon antennas V_1 and V_2 . We do this by determining the *geometric delay* τ_g , which is the time difference between the plane wave arriving at V_1 and V_2 .

$$\begin{aligned} \tau_g &= \frac{\text{Distance of the projection of } \vec{\mathbf{b}} \text{ in the direction of } \hat{\mathbf{s}}}{\text{Speed of light}} \\ &= \frac{\vec{\mathbf{b}} \cdot \hat{\mathbf{s}}}{c} \end{aligned} \quad (2)$$

In Equation (2), $\hat{\mathbf{s}}$ is a unit vector but $\vec{\mathbf{b}}$ is not. This is because we do not know the distance to the source, \mathbf{S} . Having obtained an expression for τ_g , we can represent the electromagnetic radiation incident at the antennas as

$$\begin{aligned} V_1 &= |V_1| \cos(\omega(t - \tau_g)) \\ V_2 &= |V_2| \cos(\omega t) \end{aligned} \quad (3)$$

2 Correlation of a point source

In signal processing *cross-correlation* is the measure of similarity between two series which are offset from each other. In our system described in Equation (3) this offset is $\omega\tau_g$.

We define the cross-correlation of the outputs of antennas V_1 and V_2 as $\langle V_1 V_2 \rangle = \frac{1}{T} \int_0^T V_1 V_2 dt$. It can be thought of as an *averaging process* because you are integrating (adding) over a certain duration and then dividing by the length of that duration.

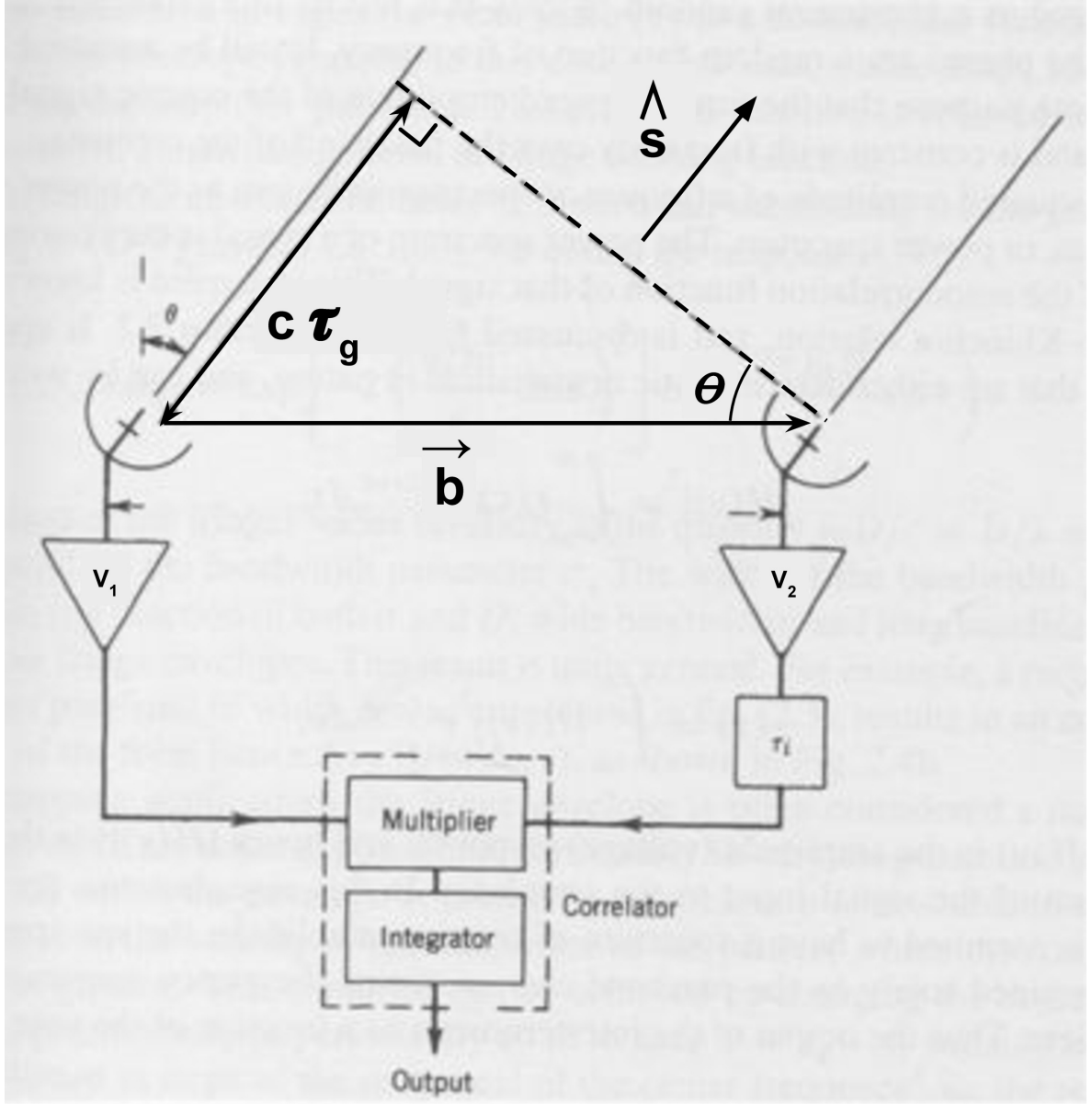


Figure 1: A schematic diagram of interferometry showing the same electromagnetic plane wave (the dotted line -----) from an astronomical source, in the direction, $\hat{\mathbf{s}}$, incident upon antennas V_1 and V_2 . The angle θ corresponds to the direction of $\hat{\mathbf{s}}$ and gives rise to the different times at which this wave arrives at V_1 and V_2 . This time difference is given as τ_g corresponding to a distance of $\mathbf{c} \tau_g$. The antennas V_1 and V_2 are separated along the baseline vector $\vec{\mathbf{b}}$.

Image credit: Interferometry and Synthesis in Radio Astronomy (Thomson, Moran and Swenson, 2017) used under the Creative Commons Attribution International License (<http://creativecommons.org/licenses/by-nc/4.0/>). The bold, black lines and text have been added by me.

$$\begin{aligned}
\langle V_1 V_2 \rangle &= \frac{1}{T} \int_0^T V_1 V_2 dt \\
&= \frac{1}{T} \int_0^T |V_1| \cos(\omega(t - \tau_g)) |V_2| \cos(\omega t) dt \\
&= \frac{1}{T} |V_1 V_2| \int_0^T \cos(\omega(t - \tau_g) + \omega t) + \sin(\omega(t - \tau_g)) \sin(\omega t) dt \\
&= \frac{1}{T} |V_1 V_2| \int_0^T \cos(2\omega t - \omega \tau_g) - \cos(\omega(t - \tau_g)) \cos(\omega t) + \cos(\omega(t - \tau_g) - \omega t) dt \\
&= \frac{1}{T} |V_1 V_2| \int_0^T \cos(2\omega t - \omega \tau_g) - \cos(\omega(t - \tau_g)) \cos(\omega t) + \cos(-\omega \tau_g) dt \\
&= \frac{1}{T} |V_1 V_2| \int_0^T \cos(2\omega t - \omega \tau_g) - \cos(\omega(t - \tau_g)) \cos(\omega t) + \cos(\omega \tau_g) dt \\
&= \frac{1}{T} |V_1 V_2| \int_0^T \cos(\omega \tau_g) + \cos(2\omega t - \omega \tau_g) - \cos(\omega(t - \tau_g)) \cos(\omega t) dt \\
&= \frac{1}{T} |V_1 V_2| \left[\int_0^T \cos(\omega \tau_g) dt + \int_0^T \cos(2\omega t - \omega \tau_g) - \cos(\omega(t - \tau_g)) \cos(\omega t) dt \right]
\end{aligned}$$

In the last step, we have grouped the terms which depend on t in a single integral. In doing so we are attempting to make use of the following property: Over some sufficient duration of $t \in [0, T]$, there will be sufficient complete cycles of the trigonometrical terms which depend on t for the result to average to 0 when integrating over dt .

(Try integrating $\int_0^{2\pi} \sin(t) dt$ or $\int_0^{2\pi} \cos(t) dt$ to get an idea why this happens over complete trigonometric cycles.)

$$\begin{aligned}
\Rightarrow \quad \langle V_1 V_2 \rangle &= \frac{1}{T} |V_1 V_2| \left[\int_0^T \cos(\omega \tau_g) dt + 0 \right] \\
&= \frac{1}{T} |V_1 V_2| \cos(\omega \tau_g) \int_0^T 1 dt \\
&= \frac{1}{T} |V_1 V_2| \cos(\omega \tau_g) [t]_0^T \\
&= \frac{1}{T} |V_1 V_2| \cos(\omega \tau_g) T \\
&= |V_1 V_2| \cos(\omega \tau_g)
\end{aligned} \tag{4}$$

In radio interferometry this process happens in the *correlator*. The term $|V_1 V_2|$ is a multiplication of voltages which is proportional to power (Ohm's Law: $P \propto V\mathcal{I} = V \frac{V}{R} = \frac{V^2}{R}$). I shall refer to $|V_1 V_2|$ as the intensity, I , of the source which is the power per unit area.

$$\begin{aligned}
\therefore \text{Correlator response, } R_c &= \langle V_1 V_2 \rangle \\
&= I \cos(\omega \tau_g)
\end{aligned} \tag{5}$$

The correlator response, R_c , is a function of ω and τ_g . In practice τ_g is difficult to time directly. So our aim now is to represent τ_g in terms of quantities which we can measure easily, like:

1. the baseline length – which **we** can measure from one antenna to another
2. the observation wavelength – dependant on the frequency range of the receivers **we** build
3. the position of the source in the sky – which is where **we** point the interferometer

In expanding τ_g in R_c we get

$$\begin{aligned}
R_c &= I \cos(\omega \tau_g) \\
&= I \cos\left(2\pi \nu \frac{\vec{\mathbf{b}} \cdot \hat{\mathbf{s}}}{c}\right) \\
&= I \cos\left(2\pi \nu \frac{\vec{\mathbf{b}} \cdot \hat{\mathbf{s}}}{\nu \lambda}\right) \\
&= I \cos\left(2\pi \frac{|b| \cos(\frac{\pi}{2} - \theta)}{\lambda}\right), \quad \text{see Figure 1} \\
&= I \cos\left(2\pi \frac{|b| \sin\theta}{\lambda}\right) \\
&= I \cos(2\pi u l), \quad \text{where } u = \frac{|b|}{\lambda}, \quad l = \sin\theta
\end{aligned} \tag{6}$$

Instead of using traditional units of distance, such as the meter, in interferometry it is more convenient to represent the baseline length in units of our observation frequency's corresponding wavelength, i.e. $u = \frac{|b|}{\lambda}$.

2.1 Visualising the correlator output

We can graphically visualise the correlator output R_c by sketching the trigonometrical function $R_c = I \cos(2\pi u l)$. The range of the cosine function $R_c \in [-I, I]$ for all values of u and l . For this example let the baseline be 3 wavelengths long $\implies u = \frac{|b|}{\lambda} = 3$ and our field-of-view extend from horizon-to-horizon $\implies -\frac{\pi}{2} \leq \theta \leq \frac{\pi}{2}$.

n	Maxima	Minima
-3	-90.0°	-56.4°
-2	-41.8°	-30.0°
-1	-19.5°	-9.6°
0	0.0°	9.6°
1	19.5°	30.0°
2	41.8°	56.4°
3	90.0°	

Table 1: Maxima and minima of R_c when $u = \frac{|b|}{\lambda} = 3$.

$$\begin{aligned}
\text{Maxima when } R_c &= I \\
\cos(2\pi u l) &= 1 \\
\therefore 2\pi u l &= 2n\pi \\
\frac{|b|}{\lambda} \sin\theta &= n \\
\theta &= \sin^{-1}\left(\frac{n}{3}\right), \quad \text{where } n \in \mathbb{Z}, |n| \leq 3
\end{aligned}$$

$$\begin{aligned}
\text{Minima when } R_c &= -I \\
\cos(2\pi u l) &= -1 \\
\therefore 2\pi u l &= (2n+1)\pi \\
\frac{|b|}{\lambda} \sin\theta &= \frac{2n+1}{2} \\
\theta &= \sin^{-1}\left(\frac{2n+1}{3(2)}\right), \quad \text{where } n \in \mathbb{Z}, -3 \leq n < 3
\end{aligned}$$

I show the maxima, minima of R_c in Table 1, and the corresponding curve in Figure 2. This curve is known as the interferometer's *fringe pattern*, and you can think of it as the pattern which governs regions of the interferometer's sensitivity to the sky. In our example this would be the pattern for a dipole array. For interferometers such as MeerKAT, the fringes are shaped by the receivers such that the central lobe dominates. You can use `01-corr-resp.py` to determine the maxima, minima and fringe patterns for arbitrary values of u and l .

At this point we can see why the interferometer's resolution power, which I described at the beginning as $\theta \sim \frac{\lambda}{D}$ is what it is. We can do this by considering the separation of one maxima to the next in

$$\begin{aligned}
\theta_{n+1} - \theta_n &= \sin^{-1}\left(\frac{n+1}{u}\right) - \sin^{-1}\left(\frac{n}{u}\right) \\
&\sim \left(\frac{n+1}{u}\right) - \left(\frac{n}{u}\right), \quad \text{using the small angle approximation} \\
&= \frac{1}{u} \\
&= \frac{\lambda}{b}
\end{aligned}$$

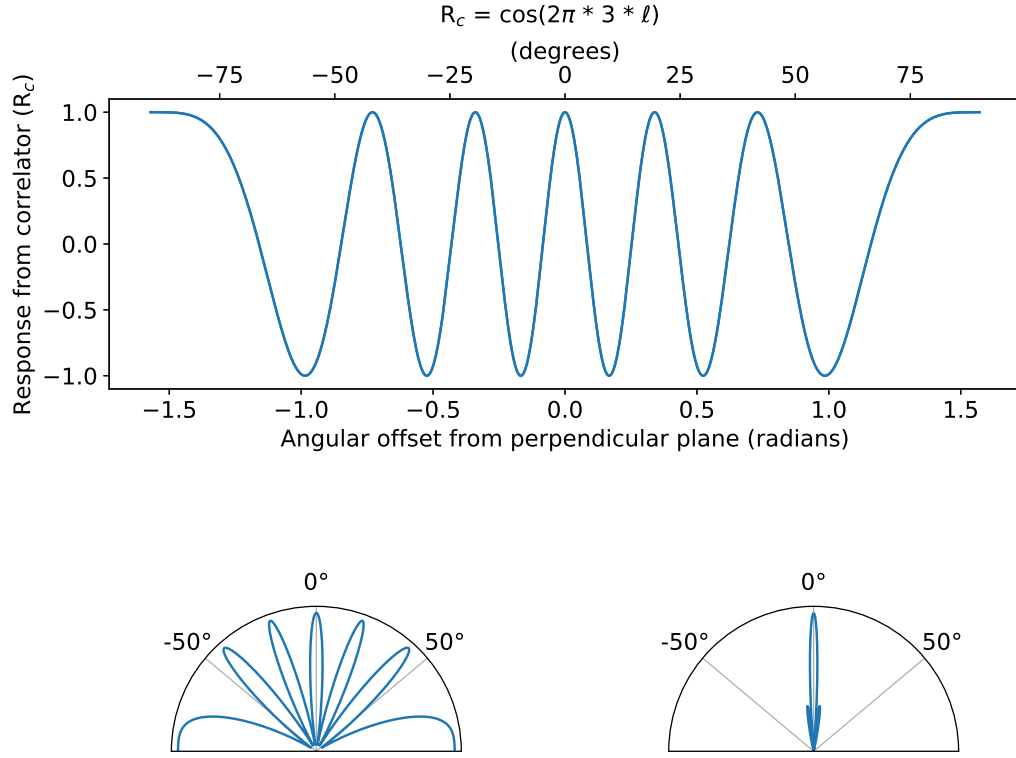


Figure 2: The correlator response, R_c in rectangular (top) coordinates. The maxima and minima of R_c are stated in Table 1. The polar plot (bottom left) shows what the fringes look like from horizon-to-horizon. The previous two plots resemble the response of a di-pole baseline. In contrast, the polar plot (bottom right) resembles what you might expect from an interferometer like MeerKAT where there is a dominant central fringe and diminished sidelobes.

3 Correlation of a non-point (finite) source

Up to this point, we have considered the correlation of a point source. However, a real astronomical source occupies a 2D region on the *celestial sphere*¹. For such an extended source we need to sum each of the antenna responses – V_1 and V_2 – over a 2D solid angle region, $d\Omega$, in the sky.

$$\therefore \langle V_1 V_2 \rangle = \left\langle \iint V_1 d\Omega \iint V_2 d\Omega \right\rangle$$

If and only if the emission is *spatially incoherent* then

$$\langle V_1 V_2 \rangle = \left\langle \iint V_1 V_2 d\Omega \right\rangle$$

(Spatial incoherence is required as in general $\int g(x) h(x) dx \neq \int g(x) dx \times \int h(x) dx$.
Typically we need to use *integration by parts* in these cases.
For example try to integrate $\int x \cos(x) dx$ v.s. $\int x dx \times \int \cos(x) dx$.)

$$\begin{aligned} &= \frac{1}{T} \int_0^T \iint V_1 V_2 d\Omega dt \\ &= \iint \frac{1}{T} \int_0^T V_1 V_2 dt d\Omega \\ &= \iint I \cos(2\pi u l) d\Omega \end{aligned} \tag{7}$$

Note that the $\cos(2\pi u l)$ term in Equation (7) has a deficiency since the \cos function is an *even* function and is therefore only sensitive to the even component of the signal. We also need a way to detect/measure/sample the *odd* component. This is because, for some general function

$$I(x) = I_{\text{even}}(x) + I_{\text{odd}}(x), \quad \text{for } -k \leq x \leq k$$

The odd and even components have the property that

¹The celestial sphere is an imaginary sphere which is concentric to Earth.

$$\begin{aligned}
I_{\text{even}}(-x) &= -I_{\text{even}}(x) \\
I_{\text{odd}}(-x) &= I_{\text{odd}}(x)
\end{aligned}$$

$$\begin{aligned}
\text{Giving } R_c &= \iint I \cos(2\pi u l) d\Omega \\
&= \iint (I_{\text{even}} + I_{\text{odd}}) \cos(2\pi u l) d\Omega \\
&= \iint I_{\text{even}} \cos(2\pi u l) + I_{\text{odd}} \cos(2\pi u l) d\Omega \\
&= \iint I_{\text{even}} \cos(2\pi u l) d\Omega + \iint I_{\text{odd}} \cos(2\pi u l) d\Omega \\
&= \iint I_{\text{even}} \cos(2\pi u l) d\Omega + \iint 0 d\Omega \\
&= \iint I_{\text{even}} \cos(2\pi u l) d\Omega
\end{aligned}$$

$\left(\text{To see why } \iint I_{\text{even}} \cos(2\pi u l) d\Omega = 0, \text{ try integrating } \int_0^{2\pi} \cos(x) \cos(x) dx \text{ v.s. } \int_0^{2\pi} \sin(x) \cos(x) dx \text{ or } \int_{-1}^1 |x| \cos(x) dx \text{ v.s. } \int_{-1}^1 x \cos(x) dx. \right)$

3.1 Odd and Even Components of the Correlator Output

So, if we wish to keep the odd component of the emission, we need to sample the *sin* component, which can be accomplished by generating a 90° phase shift in one of the signal paths from the pair of Equation (3)

$$\begin{aligned}
V_1 &= |V_1| \cos(\omega(t - \tau_g)) \\
V_2 &= |V_2| \cos(\omega t - \frac{\pi}{2})
\end{aligned} \tag{8}$$

In cross-correlating these antenna outputs (as we did to obtain Equation (4))

$$\begin{aligned}
\langle V_1 V_2 \rangle &= \frac{1}{T} \int_0^T V_1 V_2 dt \\
&= \frac{1}{T} \int_0^T |V_1| \cos(\omega(t - \tau_g)) |V_2| \cos(\omega t - \frac{\pi}{2}) dt \\
&= \frac{1}{T} |V_1 V_2| \int_0^T \cos(\omega(t - \tau_g) + \omega t - \frac{\pi}{2}) + \sin(\omega(t - \tau_g)) \sin(\omega t - \frac{\pi}{2}) dt \\
&= \frac{1}{T} |V_1 V_2| \int_0^T \cos(2\omega t - \omega \tau_g - \frac{\pi}{2}) - \cos(\omega(t - \tau_g)) \cos(\omega t - \frac{\pi}{2}) + \dots \\
&\quad \dots \cos\left[\omega(t - \tau_g) - (\omega t - \frac{\pi}{2})\right] dt \\
&= \frac{1}{T} |V_1 V_2| \int_0^T \cos(2\omega t - \omega \tau_g - \frac{\pi}{2}) - \cos(\omega(t - \tau_g)) \cos(\omega t - \frac{\pi}{2}) + \cos(\omega \tau_g - \frac{\pi}{2}) dt \\
&= \frac{1}{T} |V_1 V_2| \left[\int_0^T \cos(\omega \tau_g - \frac{\pi}{2}) dt + \dots \right. \\
&\quad \left. \dots \int_0^T \cos(2\omega t - \omega \tau_g - \frac{\pi}{2}) - \cos(\omega(t - \tau_g)) \cos(\omega t - \frac{\pi}{2}) dt \right]
\end{aligned}$$

Again we have the rapidly varying terms averaging to 0 and we substitute I for $|V_1 V_2|$.

$$\begin{aligned}
\Rightarrow \quad \langle V_1 V_2 \rangle &= \frac{1}{T} I \left[\int_0^T \cos(\omega \tau_g - \frac{\pi}{2}) dt + 0 \right] \\
&= I \cos(\omega \tau_g - \frac{\pi}{2}) \\
&= I \left[\cos(\omega \tau_g) \cos(\frac{\pi}{2}) + \sin(\omega \tau_g) \sin(\frac{\pi}{2}) \right] \\
&= I \sin(\omega \tau_g)
\end{aligned}$$

$$\therefore R_s = I \sin(\omega \tau_g)$$

The next natural step is to define a complex function in terms of the *sin* and *cos* components of the correlator output.

$$\begin{aligned}
\therefore \langle V_1 V_2 \rangle &= R_c - i R_s, \quad \text{where } i = \sqrt{-1} \\
&= I \cos(\omega \tau_g) - i I \sin(\omega \tau_g) \\
&= I e^{-i \omega \tau_g} \\
&= I e^{-i 2\pi u l}, \quad \text{see Equation (6) for converting } \omega \tau_g \text{ to } 2\pi u l
\end{aligned}$$

(We use $R_c - i R_s$ instead of $R_c + i R_s$ to maintain the convention of the Fourier Transform) relationship which will be revealed in Equation (11).

In conjunction with Equation (7) the correlator output is now called the *complex visibility* and is given by

$$\begin{aligned}
\mathcal{V}(\vec{b}) &= \iint I_\nu(\hat{s}) e^{-i 2\pi \nu \frac{\vec{b} \cdot \hat{s}}{c}} d\Omega \\
&= \iint I_\nu(\hat{s}) e^{-i 2\pi u l} d\Omega
\end{aligned} \tag{9}$$

I have replaced I with, $I_\nu(\hat{s})$ which is now the source brightness intensity at observing frequency ν and in the direction \hat{s} . At this stage, we are beginning to link something that we can measure, $\mathcal{V}(\vec{b}) = \langle V_1 V_2 \rangle$, with something that we want to obtain, $I_\nu(\hat{s})$.

3.2 Visualising the result

Again, let's try to get an intuitive feel for what $\mathcal{V}(\vec{b})$ is. In the following example, consider the simplest possible case where we have a 1 D baseline such that $\mathcal{V}(\vec{b}) = \mathcal{V}(u)$ which samples the source intensity along 1 D such that $I_\nu(\hat{s}) = I_\nu(l)$. $d\Omega$ now reduces to dl . Furthermore, let us assume that the source intensity can be modelled as a *Dirac δ -function* $\implies I_\nu(l) = \delta(l - l_0)$. $\mathcal{V}(\vec{b})$ can now be expressed as

$$\begin{aligned}
\mathcal{V}(u) &= \int \delta(l - l_0) e^{-i 2\pi u l} dl \\
&= \int \delta(l - l_0) e^{-i 2\pi u l} dl \Big|_{l < l_0} + \int \delta(l - l_0) e^{-i 2\pi u l} dl \Big|_{l = l_0} + \int \delta(l - l_0) e^{-i 2\pi u l} dl \Big|_{l > l_0} \\
&= \int 0 \times e^{-i 2\pi u l} dl + \int \delta(0) e^{-i 2\pi u l_0} dl + \int 0 \times e^{-i 2\pi u l} dl \\
&= e^{-i 2\pi u l_0} \int \delta(0) dl \\
&= e^{-i 2\pi u l_0} [1] \\
&= \cos(2\pi u l_0) - i \sin(2\pi u l_0)
\end{aligned}$$

This means that the output of the correlator, $\mathcal{V}(u)$, is just a complex number. In Figure 3, we see $\mathcal{V}(u)$ displayed in both trigonometrical form (middle column) and amplitude & phase form (right column), where the amplitude $= \sqrt{\cos^2(2\pi u l_0) - \sin^2(2\pi u l_0)} = 1$ and the phase $= \tan^{-1} \left[-\frac{\sin(2\pi u l_0)}{\cos(2\pi u l_0)} \right]$.

In Figure 4, we add a level of complexity by replacing the $I_\nu(l)$ in Equation (9) with a Gaussian function instead. Both these figures can be replicated using `03-dirac-vis.py` and `04-gauss-vis.py` for arbitrary source offsets, as well as source width for the latter case.

4 Finite bandwidth

The bandwidth has a dramatic effect on the correlator response. In the following computation, I am ignoring the effect of source structure.

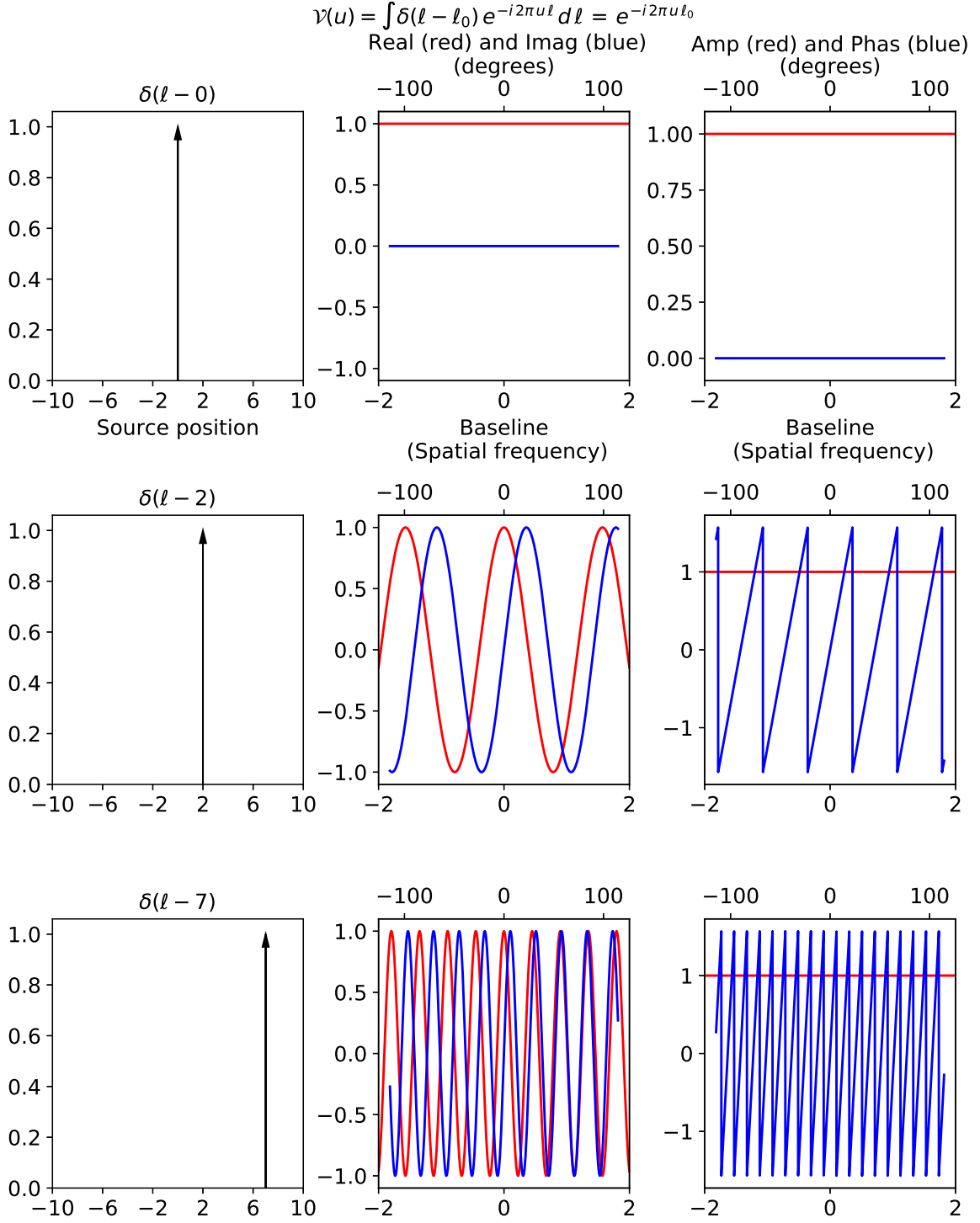


Figure 3: The output from the correlator, \mathcal{V} is always a complex number. Here we see \mathcal{V} in trigonometric (middle column) and amplitude & phase (right column) for the simplest case where $\mathcal{V}(\vec{b}) = \mathcal{V}(u)$ and a source $I_\nu(\hat{s}) = I_\nu(l) = \delta(l - l_0)$, which is offset from the phase centre by $l = 0, 2$ & 7 units (left column). The phase in the right column $\in [-\frac{\pi}{2}, \frac{\pi}{2}]$ radians.

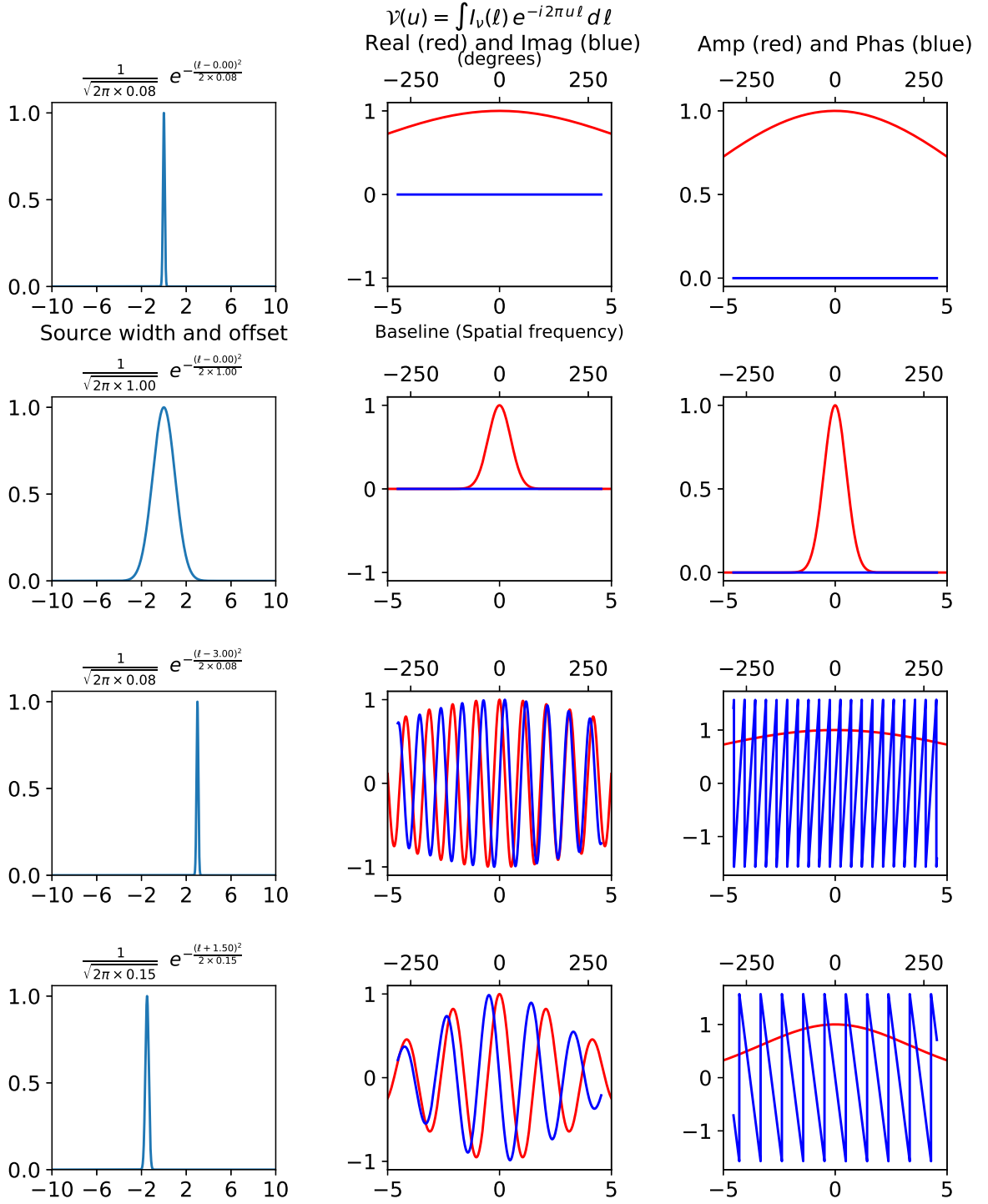


Figure 4: The correlator response to different Gaussian functions.

$$\begin{aligned}
\mathcal{V}(\vec{b}) &= \iiint I_\nu(\hat{s}) e^{-i 2\pi \nu \frac{\vec{b} \cdot \hat{s}}{c}} d\Omega \quad \leftarrow \text{quasi-monochromatic, i.e. infinitesimal bandwidth} \\
&= \iiint I_\nu(\hat{s}) \frac{1}{\Delta\nu} \int_{\nu_0 + \frac{\Delta\nu}{2}}^{\nu_0 + \frac{\Delta\nu}{2}} e^{-i 2\pi \nu \frac{\vec{b} \cdot \hat{s}}{c}} d\nu d\Omega \quad \leftarrow \text{finite bandwidth} \\
&= \iiint I_\nu(\hat{s}) \frac{1}{\Delta\nu} \int_{\nu_0 + \frac{\Delta\nu}{2}}^{\nu_0 + \frac{\Delta\nu}{2}} e^{-i 2\pi \nu \tau_g} d\nu d\Omega \\
&= \iiint I_\nu(\hat{s}) \frac{1}{\Delta\nu} \int_{\nu_0 + \frac{\Delta\nu}{2}}^{\nu_0 + \frac{\Delta\nu}{2}} \cos(2\pi \nu \tau_g) - i \sin(2\pi \nu \tau_g) d\nu d\Omega \\
&= \iiint I_\nu(\hat{s}) \frac{1}{\Delta\nu} \left[\frac{\sin(2\pi \nu \tau_g)}{2\pi \tau_g} + i \frac{\cos(2\pi \nu \tau_g)}{2\pi \tau_g} \right]_{\nu_0 + \frac{\Delta\nu}{2}}^{\nu_0 + \frac{\Delta\nu}{2}} d\Omega \\
&= \iiint I_\nu(\hat{s}) \frac{1}{\Delta\nu 2\pi \tau_g} \left[\sin(2\pi \nu \tau_g) + i \cos(2\pi \nu \tau_g) \right]_{\nu_0 + \frac{\Delta\nu}{2}}^{\nu_0 + \frac{\Delta\nu}{2}} d\Omega \\
&= \iiint I_\nu(\hat{s}) \frac{1}{2\pi \tau_g \Delta\nu} \left[\sin\left[2\pi \left(\nu_0 + \frac{\Delta\nu}{2}\right) \tau_g\right] + i \cos\left[2\pi \left(\nu_0 + \frac{\Delta\nu}{2}\right) \tau_g\right] \dots \right. \\
&\quad \left. \dots - \sin\left[2\pi \left(\nu_0 - \frac{\Delta\nu}{2}\right) \tau_g\right] - i \cos\left[2\pi \left(\nu_0 - \frac{\Delta\nu}{2}\right) \tau_g\right] \right] d\Omega \\
&= \iiint I_\nu(\hat{s}) \frac{1}{2\pi \tau_g \Delta\nu} \left[\sin\left(2\pi \nu_0 \tau_g + 2\pi \frac{\Delta\nu}{2} \tau_g\right) + i \cos\left(2\pi \nu_0 \tau_g + 2\pi \frac{\Delta\nu}{2} \tau_g\right) \dots \right. \\
&\quad \left. \dots - \sin\left(2\pi \nu_0 \tau_g - 2\pi \frac{\Delta\nu}{2} \tau_g\right) - i \cos\left(2\pi \nu_0 \tau_g - 2\pi \frac{\Delta\nu}{2} \tau_g\right) \right] d\Omega \\
&= \iiint I_\nu(\hat{s}) \frac{1}{2\pi \tau_g \Delta\nu} \left[\sin(2\pi \nu_0 \tau_g) \cos\left(2\pi \frac{\Delta\nu}{2} \tau_g\right) + \cos(2\pi \nu_0 \tau_g) \sin\left(2\pi \frac{\Delta\nu}{2} \tau_g\right) \dots \right. \\
&\quad \dots + i \cos(2\pi \nu_0 \tau_g) \cos\left(2\pi \frac{\Delta\nu}{2} \tau_g\right) - i \sin(2\pi \nu_0 \tau_g) \sin\left(2\pi \frac{\Delta\nu}{2} \tau_g\right) \dots \\
&\quad \dots - \sin(2\pi \nu_0 \tau_g) \cos\left(2\pi \frac{\Delta\nu}{2} \tau_g\right) + \cos(2\pi \nu_0 \tau_g) \sin\left(2\pi \frac{\Delta\nu}{2} \tau_g\right) \dots \\
&\quad \left. \dots - i \cos(2\pi \nu_0 \tau_g) \cos\left(2\pi \frac{\Delta\nu}{2} \tau_g\right) - i \sin(2\pi \nu_0 \tau_g) \sin\left(2\pi \frac{\Delta\nu}{2} \tau_g\right) \right] d\Omega
\end{aligned}$$

$$\begin{aligned}
&= \iint I_\nu(\hat{s}) \frac{1}{2\pi \tau_g \Delta\nu} \left[\cancel{\sin(2\pi \nu_0 \tau_g) \cos\left(2\pi \frac{\Delta\nu}{2} \tau_g\right)} + \cos(2\pi \nu_0 \tau_g) \sin\left(2\pi \frac{\Delta\nu}{2} \tau_g\right) \dots \right. \\
&\quad \dots + i \cos(2\pi \nu_0 \tau_g) \cos\left(2\pi \frac{\Delta\nu}{2} \tau_g\right) - i \sin(2\pi \nu_0 \tau_g) \sin\left(2\pi \frac{\Delta\nu}{2} \tau_g\right) \dots \\
&\quad \dots - \cancel{\sin(2\pi \nu_0 \tau_g) \cos\left(2\pi \frac{\Delta\nu}{2} \tau_g\right)} + \cos(2\pi \nu_0 \tau_g) \sin\left(2\pi \frac{\Delta\nu}{2} \tau_g\right) \dots \\
&\quad \left. \dots - i \cos(2\pi \nu_0 \tau_g) \cos\left(2\pi \frac{\Delta\nu}{2} \tau_g\right) - i \sin(2\pi \nu_0 \tau_g) \sin\left(2\pi \frac{\Delta\nu}{2} \tau_g\right) \right] d\Omega \\
&= \iint I_\nu(\hat{s}) \frac{1}{2\pi \tau_g \Delta\nu} \left[2 \cos(2\pi \nu_0 \tau_g) \sin\left(2\pi \frac{\Delta\nu}{2} \tau_g\right) - 2i \sin(2\pi \nu_0 \tau_g) \sin\left(2\pi \frac{\Delta\nu}{2} \tau_g\right) \right] d\Omega \\
&= \iint I_\nu(\hat{s}) \frac{2 \sin\left(2\pi \frac{\Delta\nu}{2} \tau_g\right)}{2\pi \tau_g \Delta\nu} \left[\cos(2\pi \nu_0 \tau_g) - i \sin(2\pi \nu_0 \tau_g) \right] d\Omega \\
&= \iint I_\nu(\hat{s}) \frac{\sin(\pi \Delta\nu \tau_g)}{\pi \tau_g \Delta\nu} \left[\cos(2\pi \nu_0 \tau_g) - i \sin(2\pi \nu_0 \tau_g) \right] d\Omega \\
&= \iint I_\nu(\hat{s}) \frac{\sin(\pi \Delta\nu \tau_g)}{\pi \tau_g \Delta\nu} e^{-i 2\pi \nu_0 \tau_g} d\Omega \tag{10}
\end{aligned}$$

What this tells us is that the correlator response $e^{-i 2\pi \nu_0 \tau_g}$ is modulated by the normalised *sinc* function $\frac{\sin(\pi \Delta\nu \tau_g)}{\pi \tau_g \Delta\nu}$ as shown in Figure 5. From Equation (10), we clearly have nulls in the envelope – where the response falls to 0 – when the numerator in $\frac{\sin(\pi \Delta\nu \tau_g)}{\pi \tau_g \Delta\nu}$ is 0.

$$\begin{aligned}
\text{i.e. when } \quad \sin(\pi \Delta\nu \tau_g) &= 0 \\
\implies \quad \Delta\nu \tau_g &= 1, \quad \text{for the first null} \\
\Delta\nu \frac{\vec{b} \cdot \hat{s}}{c} &= 1 \\
\Delta\nu b \sin\theta &= c \\
\therefore \sin\theta &= \frac{c}{b \Delta\nu} = \frac{\frac{\lambda}{b}}{\frac{\Delta\nu}{\nu}}
\end{aligned}$$

The number of fringes between the peak and this first null can be approximated by

$$\begin{aligned}
N &\sim \frac{\text{Distance between peak and first null}}{\text{Distance between fringes}} \\
&= \frac{\frac{c}{b \Delta\nu}}{\frac{\lambda}{b}} \\
&= \frac{c}{b \Delta\nu} \cdot \frac{b}{\lambda} = \frac{\nu}{\Delta\nu}
\end{aligned}$$

i.e. something like observing frequency divided by the bandwidth. The *sinc* envelope can also be expressed in a way more useful to the interferometer's characteristics rather than τ_g since $\sin(\pi \tau_g \Delta\nu) = \text{sinc}\left(\pi \frac{b}{\lambda} \frac{\Delta\nu}{\nu} \sin\theta\right) = \text{sinc}\left(\pi \frac{b \Delta\nu}{c} \sin\theta\right)$.

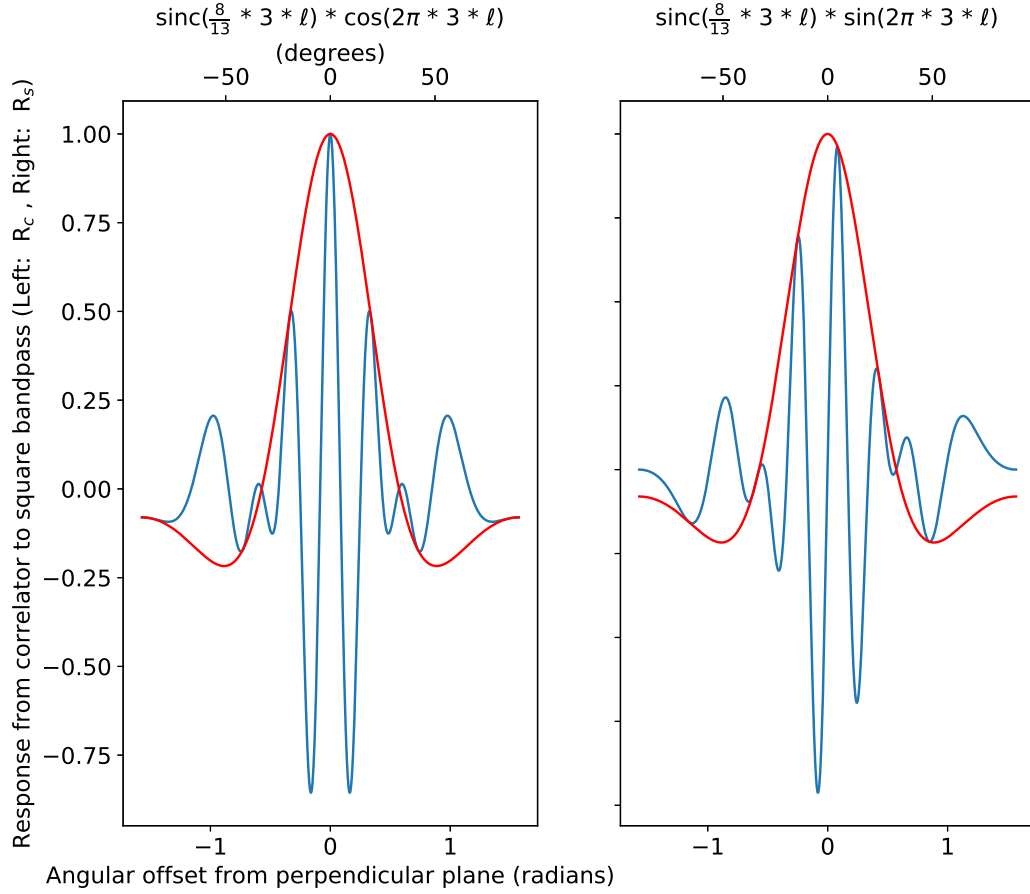


Figure 5: The correlator *cosine* (left) and *sine* (right) responses (blue lines) when we have a finite-bandwidth of $\Delta\nu = 8$ units (e.g. GHz) and an observing frequency of 13 units for $u = \frac{|b|}{\lambda} = 3$ and l extends from horizon-to-horizon $\in [-\frac{\pi}{2}, \frac{\pi}{2}]$. The red line is the *sinc* envelope due to the bandwidth $\Delta\nu$. This is analogous to Figure 2, except we assume an infinitesimal bandwidth and only a *cosine* response in that previous figure. Use `05-fin-band.py` to replicate this figure.

5 Coordinate systems

We need a formal coordinate framework in order to proceed further with the mathematics. This is described in Figure 6. Let the baseline have components, $\frac{\vec{b}}{\lambda} = u \hat{i} + v \hat{j} + w \hat{k}$. Let the source have components, $\hat{s} = l \hat{i} + m \hat{j} + n \hat{k}$.

$$\begin{aligned} \frac{\vec{b}}{\lambda} \cdot \hat{s} &= \frac{\vec{b} \cdot \hat{s}}{\lambda} = (u \hat{i} + v \hat{j} + w \hat{k}) \cdot (l \hat{i} + m \hat{j} + n \hat{k}) \\ &= (u l + v m + w n) \end{aligned}$$

$$\left(\begin{array}{l} \text{Note that } \hat{s} \text{ is a unit vector, } |\hat{s}| = 1. \\ \implies \quad \quad \quad l^2 + m^2 + n^2 = 1 \\ \quad \quad \quad n = \sqrt{1 - l^2 - m^2} \end{array} \right)$$

At this point, we employ a simplification to assume that the antennas lie on a plane such that $w = 0$.

$$\therefore \frac{\vec{b} \cdot \hat{s}}{\lambda} = (u l + v m)$$

Equation (9), $\mathcal{V}(\vec{b}) = \iint I_\nu(\hat{s}) e^{i 2\pi \nu \frac{\vec{b} \cdot \hat{s}}{c}} d\Omega$, now becomes

$$\begin{aligned} \mathcal{V}(\vec{b}) &= \iint I_\nu(\hat{s}) e^{-i 2\pi \nu \frac{\vec{b} \cdot \hat{s}}{c}} d\Omega \\ \mathcal{V}(\vec{b}) &= \iint I_\nu(\hat{s}) e^{-i 2\pi \nu \frac{\vec{b} \cdot \hat{s}}{\nu \lambda}} d\Omega \\ \mathcal{V}(\vec{b}) &= \iint I_\nu(\hat{s}) e^{-i 2\pi \frac{\vec{b} \cdot \hat{s}}{\lambda}} d\Omega \\ \mathcal{V}(u, v) &= \iint I_\nu(l, m) e^{-i 2\pi (u l + v m)} dl dm \end{aligned}$$

In the final step, we replace the vectors \vec{b} and \hat{s} with their corresponding vector components $u - v$ and $l - m$. You can think of the area $d\Omega \approx dl dm$ (see the shaded area $d\Omega$ in Figure 6). We now have a 2D Fourier transform between the visibility, $\mathcal{V}(u, v)$, which we can measure, and the source brightness distribution, $I_\nu(l, m)$, which we want.

$$\begin{aligned} \mathcal{V}(u, v) &= \iint I_\nu(l, m) e^{-i 2\pi (u l + v m)} dl dm \\ I_\nu(l, m) &= \iint \mathcal{V}(u, v) e^{i 2\pi (u l + v m)} du dv \end{aligned} \tag{11}$$

This is known as the *Van Cittert–Zernicke Theorem* and $I_\nu(l, m)$ is the true sky brightness distribution and $\mathcal{V}(u, v)$ is the true sky visibility.

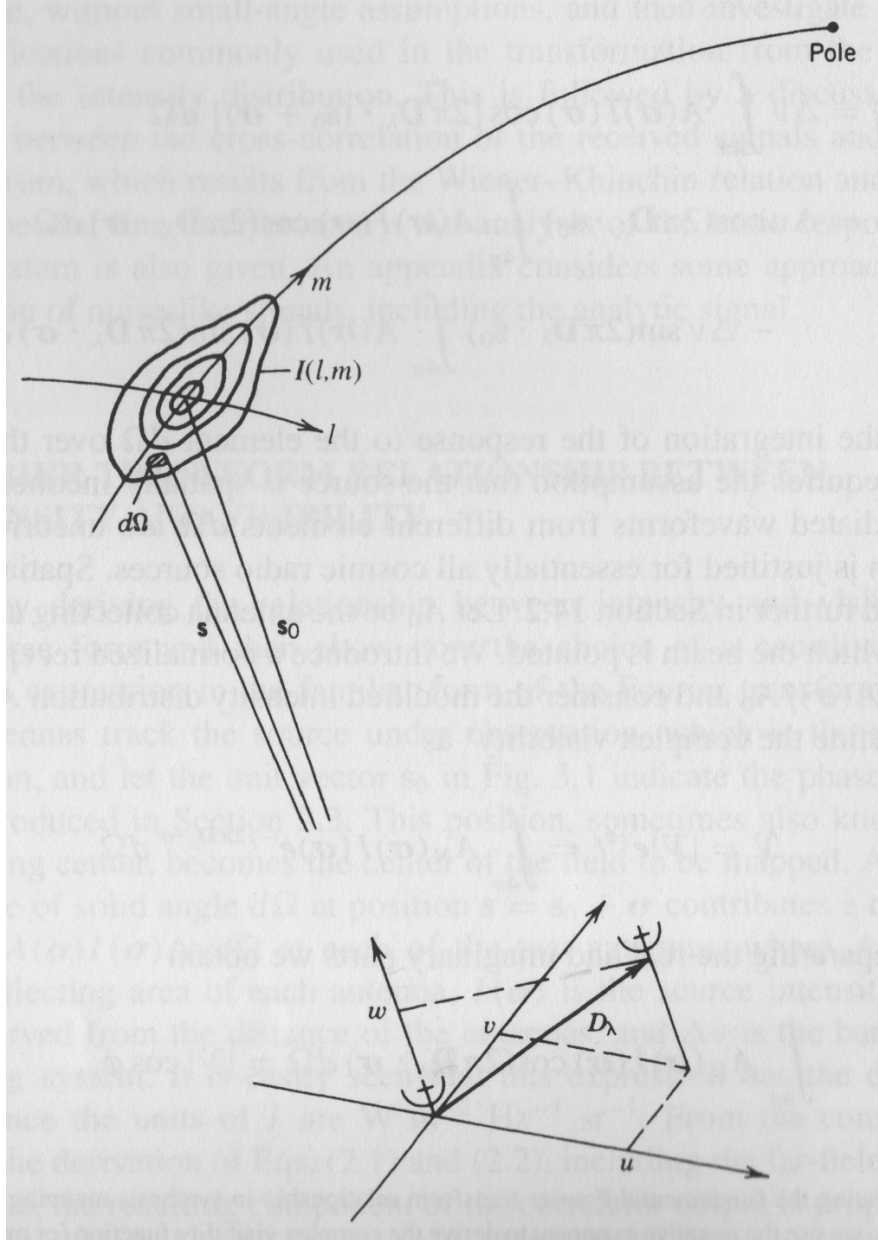


Figure 6: There are two coordinate frames which we use in interferometry. The source is in the $l - m$ plane and the interferometer is in the $u - v$ plane.

Image credit: Interferometry and Synthesis in Radio Astronomy (Thomson, Moran and Swenson, 2017) used under the Creative Commons Attribution International License (<http://creativecommons.org/licenses/by-nc/4.0/>).

6 UV sampling

According to equation (11), the source intensity $I_\nu(l, m)$ can be perfectly described by $\mathcal{V}(u, v)$. However, in practice this is not the case as the (u, v) sampling is baseline dependent. Figure 7a (left panel) shows a hypothetical triangular array with elements A , B and C . With respect to the centre $(0, 0)$ of this array these have corresponding position vectors \vec{OA} , \vec{OB} and \vec{OC} . Since the $u - v$ space is derived from the baseline vector, we need to determine the individual baseline coordinates \vec{AB} , \vec{AC} and \vec{BC} (and corresponding degenerate baselines \vec{BA} , \vec{CA} and \vec{CB}). For example,

$$\begin{aligned}\vec{AB} &= \vec{AO} + \vec{OB} \\ &= -\vec{OA} + \vec{OB} \\ &= -(0.5 \hat{i} - 0.5 \hat{j}) + (0 \hat{i} + 0.5 \hat{j}) \\ &= -0.5 \hat{i} + 1 \hat{j}\end{aligned}$$

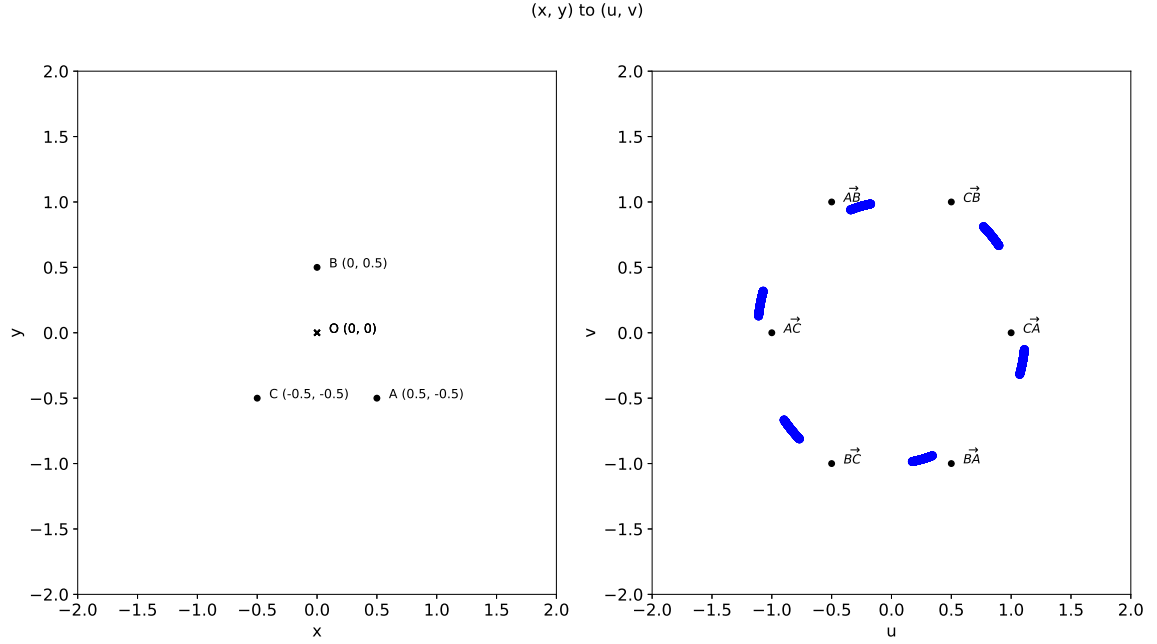
The right panel of Figure 7a is a graphical representation of what is known as the *sampling pattern*, $S(u, v)$, because it is where $\mathcal{V}(u, v)$ is sampled such that we have the *dirty image*

$$\begin{aligned}\text{True image, } I_\nu(l, m) &= \iint \mathcal{V}(u, v) e^{i 2\pi (ul + vm)} du dv \quad \text{from Equation (11)} \\ \text{Dirty image, } I_D(l, m) &= \iint S(u, v) \mathcal{V}(u, v) e^{i 2\pi (ul + vm)} du dv\end{aligned}\tag{12}$$

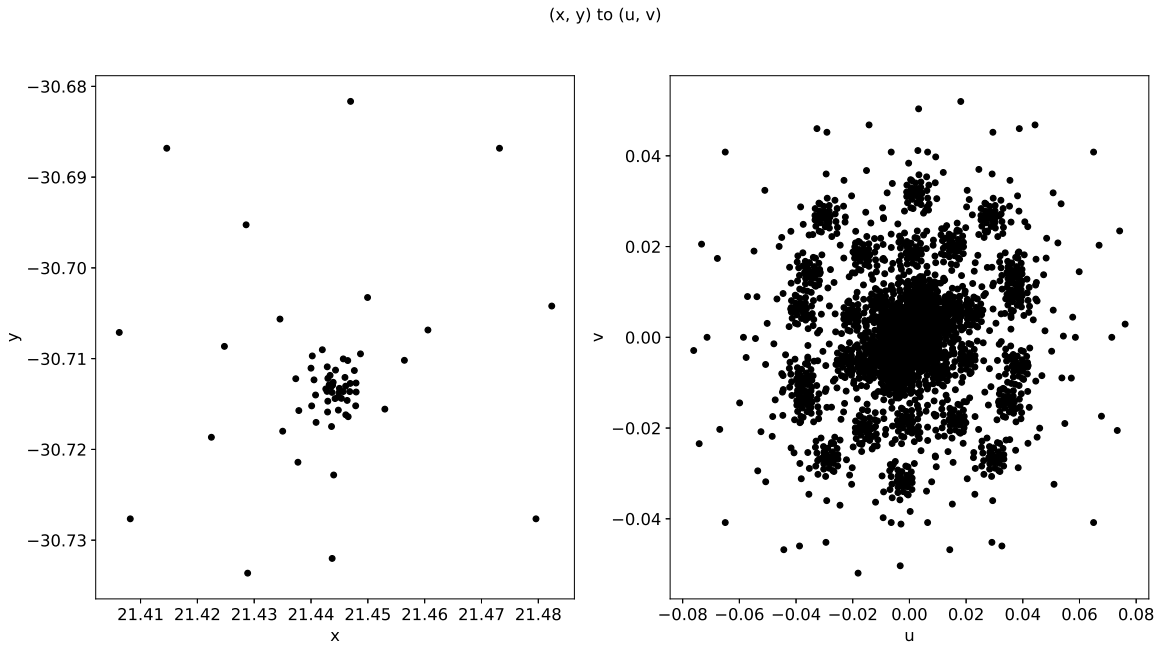
We know that a product of two functions in the $u - v$ Fourier domain corresponds to a convolution of two signals in the $l - m$ domain such that $I_D(l, m) = I_\nu(l, m) * B(l, m)$, where $B(l, m)$ is known as the *dirty beam* and is the Fourier Transform of $S(u, v)$

$$B(l, m) = \iint S(u, v) e^{i 2\pi (ul + vm)} du dv$$

I have summarised the relationship between $I_\nu(l, m)$, $\mathcal{V}(u, v)$, $B(l, m)$, $S(u, v)$, $I_D(l, m)$ and $\mathcal{V}(u, v) S(u, v)$ in Table 2. Recovering $I_\nu(l, m)$, $\mathcal{V}(u, v)$ from $I_D(l, m)$ is a deconvolution problem which can be accomplished using CLEAN.



(a) The black circles represent antenna positions of a hypothetical array (left panel) on the Earth's surface (x, y) and corresponding uv -coordinates (right panel). The blue arcs trace the loci of these $u-v$ points which are sampled as the Earth rotates. The offset between the start of the blue points and the black points is due to a coordinate rotation which I have applied in keeping with convention. You can use `07-array2uv-loci.py` to reproduce this figure.



(b) The black circles represent MeerKAT's antenna positions (left panel) on the Earth's surface (x, y) and corresponding uv -coordinates (right panel).

Figure 7

l – m (image) domain	$I_\nu(l, m)$ True image <i>(Desired)</i>	*	$B(l, m)$ Dirty beam	=	$I_D(l, m)$ Dirty image
$\mathcal{F.T.}$	\Updownarrow		\Updownarrow		\Updownarrow
u – v (Fourier) domain	$\mathcal{V}(u, v)$ True visibility	×	$S(u, v)$ Sampling pattern <i>(Known)</i>	=	$\mathcal{V}(u, v) S(u, v)$ <i>(Measured)</i>

Table 2: A summary of the relationship between quantities in the image and Fourier domains.