



Principles and symmetries of complexity in quantum field theory

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Received: 2 October 2018 / Accepted: 14 January 2019 / Published online: 4 February 2019

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Abstract Based on general and minimal properties of the *discrete* circuit complexity, we define the complexity in *continuous* systems in a geometrical way. We first show that the Finsler metric naturally emerges in the geometry of the complexity in continuous systems. Due to fundamental symmetries of quantum field theories, the Finsler metric is more constrained and consequently, the complexity of $SU(n)$ operators is uniquely determined as a length of a geodesic in the Finsler geometry. Our Finsler metric is bi-invariant contrary to the right-invariance of discrete qubit systems. We clarify why the bi-invariance is relevant in quantum field theoretic systems. After comparing our results with discrete qubit systems we show most results in k -local right-invariant metric can also appear in our framework. Based on the bi-invariance of our formalism, we propose a new interpretation for the Schrödinger's equation in isolated systems – the quantum state evolves by the process of minimizing “computational cost”.

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1 Introduction

Based on the intuition that the classical spacetime geometry encodes information theoretic properties of the dual quantum field theory (QFT) in the context of gauge/gravity duality, many quantum information concepts have been applied to investigations of gravity theories. A notable example is

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the holographic entanglement entropy (EE) of a subregion in a QFT [1]. Even though EE has played a crucial role in understanding the dual gravity, it turned out that EE is not enough [2], in particular, when it comes to the interior of the black hole. In the eternal AdS black hole, an Einstein–Rosen bridge (ERB) connecting two boundaries continues to grow for longer time scale even after thermalization. Because EE quickly saturates at the equilibrium, it cannot explain the growth of the ERB and another quantum information concept, *complexity*, was introduced as a dual to the growth of ERB [3,4]. To ‘geometrize’ the complexity of quantum states in the dual gravity theory, two conjectures were proposed: complexity-volume (CV) conjecture [4] and complexity-action (CA) conjecture [5], which are called *holographic complexity*.¹ See also Refs. [15–17].

However, note that the complexity in information theory is well-defined in *discrete* systems such as quantum circuits [18]. For example, the so-called *circuit complexity* is the minimal number of simple elementary gates required to approximate a target operator in quantum circuit. On the contrary, holographic complexity is supposed to be dual to complexity in a QFT, a *continuous* system. Thus, there may be a mismatch in duality if we try to compare the holographic complexity with the results purely based on the intuition from circuit complexity and it is important to develop the theory of complexity in QFT. Compared with much progress in holographic complexity, the precise meaning of the complexity in QFT is still not complete. In order to define complexity in QFT systematically, we start with the complexity of operator. The complexity between states will be obtained based on the complexity of operator. For the complexity of states we make a brief comment in Sect. 8 and refer to [19] for more detail. Our strategy to define the complexity of operator is: (i) extract minimal and essential axioms for the complexity of operator from the circuit complexity and (ii) define the complexity in *continuous* QFT systems based on that minimal axioms and smoothness (from continuity) (iii) consider general symmetries of QFT to give constrain on the structure of complexity. It will turn out that these steps enable us to determine the complexity of the $SU(n)$ operators uniquely.

We want to emphasize that *not all* properties of circuit complexity survive in the complexity in QFT. The difference between discreteness and continuity makes some essential differences in properties of the complexity. For example, a few basic concepts in “circuit complexity” (computational complexity), such as “gates”, are not well defined in general quantum QFT so they should be modified or abandoned. Thus, we will keep only the most essential properties of the circuits complexity (which will be abstracted into the axioms

G1–G3 in the following section). As another essential ingredient from QFT side, we will take advantages of basic symmetries of QFT, which may not be necessary in the case of quantum circuits or computer science. Because of the effect of this new inputs from QFT, some properties of the complexity in QFT we obtained may be incompatible with quantum circuits or qubit systems but they are more appropriate for QFT.

Our work is also inspired by a geometric approach by Nielsen et al. [20–22], where the discrete circuit complexity for a target operator is identified with the minimal geodesic distance connecting the target operator and the identity in a certain Finsler geometry [23–26], which is just Riemannian geometry without the quadratic restriction. Recently, inspired by this geometric method, Refs. [13,27–30] also investigated the complexity in QFT. However, in these studies, because the Finsler metric can be chosen arbitrarily, there is a shortcoming that the complexity depends on the choice of the metric. In this paper, we show that, for $SU(n)$ operators, the Finsler metric and complexity are uniquely determined based on four general axioms (denoted by **G1–G4**) and the basic symmetries of quantum QFT.

In order to make our logic and claims clear we show a schematic map for the logic structure of this paper in Fig. 1. We want to answer the following questions: (1) for what operators can we define complexity? (2) what are basic properties that complexity should satisfy? (3) for quantum field theory, what symmetries should appear in the complexity? (4) what can we obtain for complexity by the answers of above three questions? (5) what are the similarities and differences compared with previous works?

Form the section perspective, this paper is organized as follows. In Sect. 2, we introduce minimal and basic concepts of the complexity and propose three axioms **G1–G3** for the complexity of operators, which are inspired by the circuit complexity. In Sect. 3, we show how the Finsler metric arises from **G1–G2** and the smoothness of the complexity (**G4**). In Sect. 4, by using fundamental symmetry properties of QFT, we investigate constraints on the Finsler metric and the complexity. In particular we show that the Finsler metric is bi-invariant by several different approaches and is determined uniquely if we take the axiom **G3** into account. We also compare our results with previous researches regarding bi-invariance. In Sect. 5 we derive the explicit form of the Finsler metric of the $SU(n)$ group. Thanks to the bi-invariance, the geodesic in the Finsler space of $SU(n)$ group (so the complexity) is easily computed. In Sect. 6, as one application of the geodesic in the bi-invariance Finsler metric, we propose a “minimal cost principle” as a new interpretation of the Schrödinger’s equation. In Sect. 7 we make a comparison between our complexity and the complexity for K -qubit systems. In Sect. 8 we conclude.

¹ Along these lines, there have been a lot of developments and we refer to some of them, [6–14], for examples.

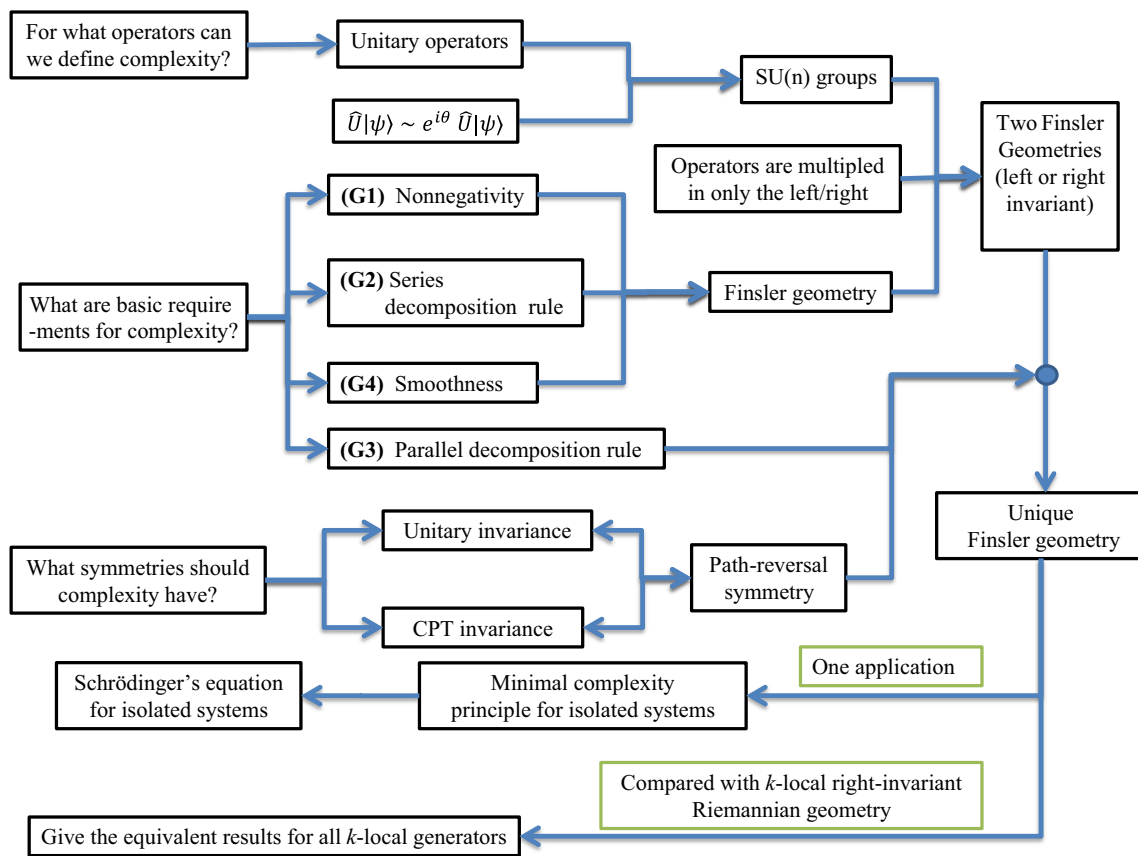


Fig. 1 The logical flows of this paper. By answering three basic questions at the far left side we show that the complexity geometry is determined by a unique bi-invariant Finsler geometry. As an application of our formalism, we show that the Schrödinger's equation for isolated

systems can be obtained from a “minimal cost principle”. Although the bi-invariant geometry looks very different from the right-invariant k -local Riemannian geometry proposed by Ref. [27], we will show that for all k -local operators two theory will give equivalent results

2 Axioms for the complexity of operators

2.1 Why unitary operators?

In order to make a good definition of the “complexity of operator” we first need to clarify what kind of “operator” we intend to deal with in this paper.

Intuitively, the complexity of operator measures how “complex” a physical process is. Thus, the operator should corresponds to a “realizable” physical process. This concept can be formulated as follows. An operator \hat{O} is called **ε -realizable** if there is at least one experimental quantum process ϕ (for example, a quantum circuit) so that the following inequality holds for arbitrary two states $|\psi_1\rangle$ and $|\psi_2\rangle$

$$\left| \langle \psi_2 | \hat{O} | \psi_1 \rangle - \langle \psi_2 | \phi(\psi_1) \rangle \right|^2 \leq \varepsilon, \quad (2.1)$$

with $\varepsilon > 0$. Here $|\phi(\psi_1)\rangle$ is the output state of the quantum process ϕ for an input state $|\psi_1\rangle$. The ε is the tolerance when we use a ϕ to approximate (simulate) the target

operator \hat{O} . Any physical system ϕ satisfying the inequality (2.1) is called an **ε -realization** of operator \hat{O} and denoted by $\phi_{\varepsilon, O}$. All ε -realizable operators form a set $\mathcal{O}_{\varepsilon}$. If an operator is ε -realizable for arbitrary positive ε , then we call it a realizable operator. For example, the identity \hat{I} , which just keeps the input as the output, is one realizable operator. All the realizable operators form the set \mathcal{O} . Quantum system $\phi_O := \lim_{\varepsilon \rightarrow 0^+} \phi_{\varepsilon, O}$ is called a realization of operator \hat{O} .

With the set of input states (S_{in}), the set of output states (S_{out}) by realizable operators ($\hat{O} \in \mathcal{O}$) can be expressed as

$$S_{\text{out}} = \mathcal{O}S_{\text{in}} := \{\hat{O}|\psi\rangle | \forall \hat{O} \in \mathcal{O}, \forall |\psi\rangle \in S_{\text{in}}\}. \quad (2.2)$$

We assume

$$S_{\text{out}} \subseteq S_{\text{in}}, \quad (2.3)$$

which makes it possible that the elements of output set can be used as new inputs. With the assumption (2.3), we can define the products of operators. The product between elements in \mathcal{O} can be defined as

$$(\hat{O}_1 \hat{O}_2) |\psi\rangle := \hat{O}_1 (\hat{O}_2 |\psi\rangle), \quad \forall |\psi\rangle. \quad (2.4)$$

By the definition of realizable operators, it can be shown that $\hat{O}_1 \hat{O}_2$ is realizable if \hat{O}_1 and \hat{O}_2 are both realizable operators. Thus, \mathcal{O} forms a monoid (semigroup with identity).

If we restrict physical processes to quantum mechanical processes, Eq. (2.1) implies that realizable operators are all unitary rather than Hermitian. In other words, our target is a property of the physical process rather than a direct observable. As quantum circuits are quantum mechanical processes and Solovay–Kitaev theorem [31] says that all the unitary operators can be approximated by some quantum circuits with any nonzero tolerance, we can conclude that the realizable operators set is the set of unitary operators. As unitary operators are invertible, the realizable operators set \mathcal{O} forms a (finite dimensional or infinite dimensional) unitary group.²

2.2 Definitions and axioms

Intuitively speaking, the circuit complexity (or computational complexity) of a target operator (or computational task) is defined by the minimal number of required fundamental gates (or fundamental steps) to simulate the target operator (or finish the computational task). Based on this intuitive concept of the complexity in quantum circuits and computations, we propose that the complexity defined in an arbitrary monoid \mathcal{O} should satisfy the following three axioms. We denote a complexity of an operator \hat{x} in an operators set \mathcal{O} by $\mathcal{C}(\hat{x})$.

G1 [Nonnegativity]

$\forall \hat{x} \in \mathcal{O}, \mathcal{C}(\hat{x}) \geq 0$ and the equality holds iff \hat{x} is the identity.

G2 [Series decomposition rule (triangle inequality)]

$\forall \hat{x}, \hat{y} \in \mathcal{O}, \mathcal{C}(\hat{x}\hat{y}) \leq \mathcal{C}(\hat{x}) + \mathcal{C}(\hat{y})$.

G3 [Parallel decomposition rule]

$\forall (\hat{x}_1, \hat{x}_2) \in \mathcal{N} = \mathcal{O}_1 \times \mathcal{O}_2 \subseteq \mathcal{O}, \mathcal{C}((\hat{x}_1, \hat{x}_2)) = \mathcal{C}((\hat{x}_1, \hat{\mathbb{I}}_2)) + \mathcal{C}((\hat{\mathbb{I}}_1, \hat{x}_2))$.

Here, in **G2**, it is possible that the operator $\hat{x}\hat{y}$ is decomposed in different ways, say $\hat{x}'\hat{y}'$. In this case, **G2** can read also as $\mathcal{C}(\hat{x}\hat{y}) = \mathcal{C}(\hat{x}'\hat{y}') \leq \mathcal{C}(\hat{x}') + \mathcal{C}(\hat{y}')$. In **G3**, we consider the case that there is a sub-monoid $\mathcal{N} \subseteq \mathcal{O}$ which can be decomposed into the Cartesian product of two monoids, i.e., $\mathcal{N} = \mathcal{O}_1 \times \mathcal{O}_2$. $\hat{\mathbb{I}}_1$ and $\hat{\mathbb{I}}_2$ are the identities of \mathcal{O}_1 and \mathcal{O}_2 . The Cartesian product of two monoids implies that $(\hat{x}_1, \hat{x}_2)(\hat{y}_1, \hat{y}_2) = (\hat{x}_1\hat{y}_1, \hat{x}_2\hat{y}_2)$ for arbitrary $(\hat{x}_1, \hat{x}_2), (\hat{y}_1, \hat{y}_2) \in \mathcal{N}$.

² Hermit operators, which correspond to observable quantities and are not unitary in general, cannot be approximated by quantum circuits if the tolerance is small enough.

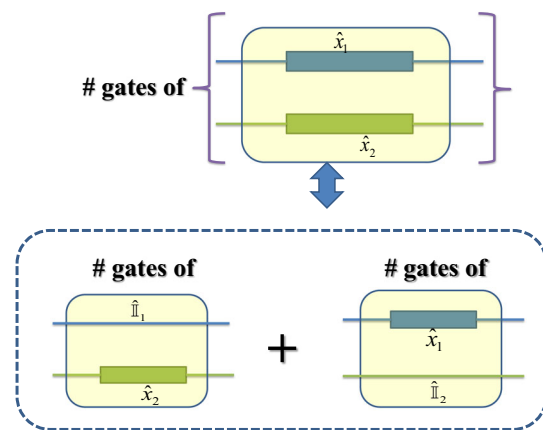


Fig. 2 Schematic diagram for the complexity of the Cartesian product and parallel decomposition rule. As two operators \hat{x}_1 and \hat{x}_2 are simulated independently, the minimally required gates for (\hat{x}_1, \hat{x}_2) is the sum of the minimally required gates for \hat{x}_1 and \hat{x}_2 . Thus, we have $\mathcal{C}((\hat{x}_1, \hat{x}_2)) = \mathcal{C}((\hat{x}_1, \hat{\mathbb{I}}_2)) + \mathcal{C}((\hat{\mathbb{I}}_1, \hat{x}_2))$

The axiom **G1** is obvious by definition. We call the axiom **G2** “series decomposition rule” because the decomposition of the operator $\hat{O} = \hat{x}\hat{y}$ to \hat{x} and \hat{y} is similar to the decomposition of a big circuit into a series of small circuits. Reversely, the ‘product’ of two operators corresponds to a serial connection of two circuits. The axiom **G2** answers a basic question: what is the relationship between the complexities of two operators and the complexity of their products? Because the complexity is a kind of “minimal”, we require the inequality in **G2**.³ This **G2** will lead to the familiar “triangle inequality” in the concept of distance (see **F3** in the Sect. 3) so it is also called “triangle inequality”.

In contrast to **G2** (series decomposition rule), we call the axiom **G3** “parallel decomposition rule”, which is chosen as one of the most basic axioms in defining complexity for the first time in this paper.⁴ It comes from the following fundamental question: if an operator (task) \hat{O} contains two *totally independent* sub-operators (sub-tasks) \hat{x}_1 and \hat{x}_2 , what should be the relationship between the total complexity and the complexities of two sub-operators (sub-tasks)? Here, the *totally independent* means that: (a) \hat{O} accepts two inputs and yields two outputs through \hat{x}_1 and \hat{x}_2 , and (b) the inputs for \hat{x}_1 (or \hat{x}_2) will never affect the outputs of \hat{x}_2 (or \hat{x}_1). See Fig. 2 for this explanation.

Mathematically, the construction of a bigger operator \hat{O} by \hat{x}_1 and \hat{x}_2 under two requirements (a) and (b) corresponds to the *Cartesian product* denoted by $\hat{O} = (\hat{x}_1, \hat{x}_2)$. Note that

³ The reason for axiom **G2** has been explained clearly also in the Nielson’s works [20–22].

⁴ In some cases, the parallel decomposition may be impossible. However, in local computations/gates, the parallel decomposition is permitted, for example see Ref. [32]. Our axiom **G3** does not talk about the possibility of the parallel decomposition, but says what will happen to the complexity if the parallel decomposition is permitted.

the Cartesian product of two monoids does not correspond to the tensor product in a linear representation (i.e., a matrix representation). Instead, it corresponds to the direct sum. For example, if matrixes M_1 and M_2 are two representations of operators \hat{x}_1 and \hat{x}_2 , then the representation of their Cartesian product \hat{O} is $M_1 \oplus M_2$, which is neither $M_1 \otimes M_2$ nor $M_1 M_2$.

In the language of computer science, this “*totally independent*” just means that one task contains two independent *parallel* tasks. Thus, the axiom **G3** tries to answer the following question: if a task contains two parallel sub-tasks, what should be the relationship between the total complexity and the complexities of such sub-tasks? In term of mathematical language, it amounts to asking: what should be the relationship between $\mathcal{C}((\hat{x}_1, \hat{x}_2))$, $\mathcal{C}((\hat{x}_1, \hat{\mathbb{I}}_2))$ and $\mathcal{C}((\hat{\mathbb{I}}_1, \hat{x}_2))$?

G3 answers this question by requiring that the complexity of two parallel tasks is the sum of their complexities, which is very natural. See Fig. 2 for a schematic explanation. In matrix representation, **G3** says, for an operator $M = M_1 \oplus M_1$, $\mathcal{C}(M_1 \oplus M_1) = \mathcal{C}(M_1) + \mathcal{C}(M_2)$. It can be generalized to the direct sum of many operators: for a finite number of matrixes M_1, M_2, \dots, M_k , we have

$$\text{Parallel decomposition rule G3: } \mathcal{C}\left(\bigoplus_{i=1}^k M_i\right) = \sum_{i=1}^k \mathcal{C}(M_i). \quad (2.5)$$

One may worry about the self-consistency between **G2** and **G3** and argue that we can only require $\mathcal{C}((\hat{O}_1, \hat{O}_2)) \leq \mathcal{C}(\hat{O}_1) + \mathcal{C}(\hat{O}_2)$, as there may be other operators $\{\hat{O}_a, \hat{O}'_a, \hat{O}_b, \hat{O}'_b\}$ to satisfy $(\hat{O}_a, \hat{O}'_a)(\hat{O}_b, \hat{O}'_b) = (\hat{O}_1, \hat{O}_2)$ but the total gates is less than $\mathcal{C}(\hat{O}_1) + \mathcal{C}(\hat{O}_2)$. However, this is impossible. One can see that the sum of the minimal gates of $\{\hat{O}_a, \hat{O}'_a, \hat{O}_b, \hat{O}'_b\}$ is $\mathcal{C}(\hat{O}_a) + \mathcal{C}(\hat{O}'_a) + \mathcal{C}(\hat{O}_b) + \mathcal{C}(\hat{O}'_b)$. But according to the fact that $\hat{O}_a \hat{O}'_a = \hat{O}_1$ and $\hat{O}_b \hat{O}'_b = \hat{O}_2$, we find that

$$\mathcal{C}(\hat{O}_a) + \mathcal{C}(\hat{O}'_a) + \mathcal{C}(\hat{O}_b) + \mathcal{C}(\hat{O}'_b) \geq \mathcal{C}(\hat{O}_1) + \mathcal{C}(\hat{O}_2).$$

Thus, $\mathcal{C}(\hat{O}_1) + \mathcal{C}(\hat{O}_2)$ is the minimal gates to obtain (\hat{O}_1, \hat{O}_2) .

The axioms **G1–G3** are satisfied by both circuit complexity and computational complexity. We have expressed the abstract concepts extracted from circuit complexity and computational complexity in terms of mathematical language and will take them as three basic requirements to define complexity also in other systems. The axiom **G1** and **G2** can be satisfied by Nielson’s original works Refs. [20–22] and recent other approaches to complexity such as Refs. [13, 27–30]. However, these works did not take into account the question related to **G3** and broke the requirement in axiom **G3** in general. From the viewpoint of quantum circuits (or computer science), series circuits (or tasks) and parallel circuits (or tasks) are two fundamental manners to decompose a bigger circuits (or tasks) into smaller ones. Thus, the axioms **G3**

should be as important as **G2**. In this paper, we propose the concept of **G3** for the first time and show that it plays a crucial role in determining the form of the complexity of $SU(n)$ operators. We may be able to modify **G3** in somewhat *unnatural* way, which will lead us to another form of the Finsler metric similar to (7.7). This point will be clarified in more detail in Ref. [33].

3 Emergence of the Finsler structure from the axioms for the complexity

In this section, we show that the Finsler metric arises from the minimal and general axioms for the complexity **G1–G3** and the smoothness of the complexity. From here, the group element may represent either an abstract object or a faithful representation, which will be understood by context.

In Sect. 2 we have shown that the realizable operators are unitary operators, so the question now becomes how to define the complexity for unitary operators. As the unitary operators \hat{U} and $e^{i\theta}\hat{U}$ (with $\theta \in (0, 2\pi)$) produce equivalent quantum states, the complexity of \hat{U} and $e^{i\theta}\hat{U}$ should be the same. Thus it is enough to study the complexity for special unitary groups, $SU(n)$ groups. Ultimately, our aim is to investigate the complexity for operators in quantum field theory, of which Hilbert space is infinite dimensional, so we have to deal with the infinite dimensional special unitary groups. However, they involve infinite dimensional manifolds and have not been well-studied even in mathematics so far. As an intermediate step, in this paper, we will first present our whole theory for finite dimensional cases and assume that the results can be generalized into infinite dimensional cases by some suitable limiting procedures. The subtle aspects between finite and infinite dimensional Lie groups are now under investigation [19].

For a given operator $\hat{O} \in SU(n)$, as $SU(n)$ is connected, there is a curve $c(s)$ connecting \hat{O} and identity $\hat{\mathbb{I}}$. The curve may be parameterized by s with $c(0) = \hat{\mathbb{I}}$ and $c(1) = \hat{O}$. See Fig. 3. The tangent of the curve, $\dot{c}(s)$, is assumed to be given by a right generator $H_r(s)$ or a left generator $H_l(s)$:

$$\dot{c}(s) = H_r(s)c(s) \quad \text{or} \quad \dot{c}(s) = c(s)H_l(s). \quad (3.1)$$

This curve can be approximated by discrete forms:

$$\hat{O}_n = c(s_n) \quad (3.2)$$

$$= \delta \hat{O}_n^{(r)} \hat{O}_{n-1} \quad (3.3)$$

$$= \hat{O}_{n-1} \delta \hat{O}_n^{(l)}, \quad (3.4)$$

where $s_n = n/N$, $n = 1, 2, 3, \dots, N$, $\hat{O}_0 = \hat{\mathbb{I}}$ and $\delta \hat{O}_n^{(\alpha)} = \exp[H_\alpha(s_n)\delta s]$ with $\alpha = r$ or l and $\delta s = 1/N$. In general, the two generators $H_r(s)$ and $H_l(s)$ at the same point of the same curve can be different, i.e., $H_r(s) \neq H_l(s)$. In fact, from Eq. (3.1), we see $H_r(s)$ is one adjoint transformation of $H_l(s)$,

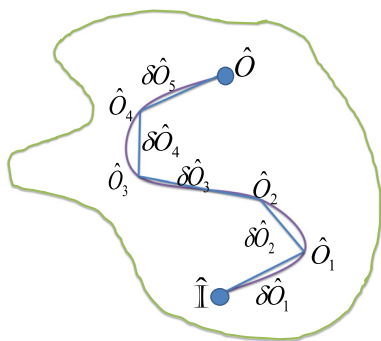


Fig. 3 A curve $c(s)$ connects the identity and a particular operator \hat{O} with the endpoints $c(0) = \hat{\mathbb{I}}$ and $c(1) = \hat{O}$. This curve can be approximated by a discrete form. Every endpoint is also an operator, which is labeled by \hat{O}_n

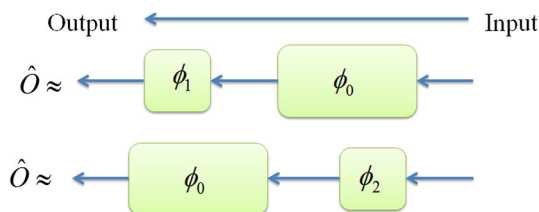


Fig. 4 Schematic diagram for two different generators in quantum circuits. To obtain the some target operator \hat{O} from the quantum circuit ϕ_0 , we have two different ways to add new circuits

$$H_r(s) = c(s)H_l(s)c(s)^{-1}. \quad (3.5)$$

The availability of two different generators can be understood also by a quantum circuit approximation to an operator, say \hat{O} . As shown in Fig. 4, if a quantum circuit ϕ_0 is given, the operator \hat{O} can be constructed in two ways: (i) by adding a new quantum circuit ϕ_1 after the output of ϕ_0 (corresponding to Eq. (3.3)) or (ii) by adding a new quantum circuit ϕ_2 before the input of ϕ_0 (corresponding to Eq. (3.4)). The previous works such as Refs. [20–22, 27] assumed that the new operators/circuits could appear only after the output side of original operators/circuits, which corresponds to Eq. (3.3). This is one mathematically allowed choice but there is no a priori or a physical reason for that particular choice. Eq. (3.4) should be equally acceptable.

The axioms **G1–G3** are suitable for arbitrary monoid, both discrete and continuous ones. Now $SU(n)$ group is a manifold, it is natural to expect that the complexity on it is a smooth function. In fact, it turns out to be enough to assume a weaker form

G4 [Smoothness] The complexity of any infinitesimal operator in $SU(n)$, $\delta\hat{O}^{(\alpha)} = \exp(H_\alpha\delta s)$, is a smooth function of only $H_\alpha \neq 0$ and $\delta s \geq 0$, i.e.,

$$\mathcal{C}(\delta\hat{O}^{(\alpha)}) = \mathcal{C}(\hat{\mathbb{I}}) + \tilde{F}(H_\alpha)\delta s + \mathcal{O}(\delta s^2), \quad (3.6)$$

where $\tilde{F}(H_\alpha) := \partial_{\delta s}\mathcal{C}(\delta\hat{O}^{(\alpha)})|_{\delta s=0}$ and $\mathcal{C}(\hat{\mathbb{I}}) = 0$ by **G1**.

which is our forth axiom. Notice that $\mathcal{C}(\delta\hat{O}^{(r)}) = \mathcal{C}(\delta\hat{O}^{(l)})$ if $\delta\hat{O}^{(r)} = \delta\hat{O}^{(l)}$, which implies that an infinitesimal operator will give the same contribution to the total complexity when it is added to the left-side or right-side.⁵ Thus, the index α is in fact not necessary in this case, but we keep it for notational consistency.

Let us define the *cost* ($L_\alpha[c]$) of a particular curve c , constructed by only $\delta\hat{O}_n^{(r)}$ or only $\delta\hat{O}_n^{(l)}$, as

$$L_\alpha[c] := \sum_{i=1}^N \mathcal{C}(\delta\hat{O}_i^{(\alpha)}) \xrightarrow{N \rightarrow \infty} \int_0^1 \tilde{F}(H_\alpha(s))ds. \quad (3.7)$$

Geometrically, it is the length of the particular curve and $\tilde{F}ds$ looks like a line element in some geometry. Thus, the natural question will be what kind of geometry is allowed for complexity? We will show that it is Finsler geometry, which emerges naturally from our axioms for the complexity.

First, we can prove that \tilde{F} satisfies three properties:

- F1** (Nonnegativity) $\tilde{F}(H_\alpha) \geq 0$ and $\tilde{F}(H_\alpha) = 0$ iff $H_\alpha = 0$
- F2** (Positive homogeneity) $\forall \lambda \in \mathbb{R}^+$, $\tilde{F}(\lambda H_\alpha) = \lambda \tilde{F}(H_\alpha)$
- F3** (Triangle inequality) $\tilde{F}(H_{\alpha,1}) + \tilde{F}(H_{\alpha,2}) \geq \tilde{F}(H_1 + H_2)$

only by using **G1**, **G2** and **G4**! (see Appendix B for a proof.) Note that **F1–F3** may describe some suitable properties that the concept of the ‘norm’ of vectors in a vector space should satisfy. In our case, the vector space is the Lie algebra (the tangent space at the identity) and the generators (H_α) of the algebra are vectors. Indeed, the ‘norm’ satisfying the properties **F1–F3** is called a *Minkowski norm* in mathematical jargon. Once we know $H_\alpha(s)$ we can compute the length of the line element by a Minkowski norm \tilde{F} . (At this stage, we don’t know the explicit form of the Minkowski norm, but we will determine it later.)

For a given \tilde{F} , we have two different natural ways to extend the Minkowski norm \tilde{F} at the identity to every point on the base manifold via arbitrary curves.

$$F_r(c, \dot{c}) := \tilde{F}(H_r) = \tilde{F}(\dot{c}c^{-1}), \text{ or } F_l(c, \dot{c}) := \tilde{F}(H_l) = \tilde{F}(c^{-1}\dot{c}). \quad (3.8)$$

where we introduce a new notation ‘ $F_\alpha(c, \dot{c})$ ’, a standard notation for *Finsler metric* in mathematics. The introduction of ‘ $F_\alpha(c, \dot{c})$ ’ is justified because the Finsler metric is nothing

⁵ This condition will be relaxed in our upcoming work [33], where we will allow different contributions to the total complexity when $\delta\hat{O}$ is added to the left-side or right-side.

but a Minkowskia norm defined at all points on the base manifold and Eq. (3.8) explains how to assign the Minkowskia norm to all the other points. We refer to Refs. [23–26] for an introduction to Minkowski norm and the Finsler geometry.⁶ A brief introduction to the Finsler geometry can be found in Appendix A.

There is an invariant property in the Finsler metrics. $F_r(c, \dot{c})$ is *right-invariant* because H_r is invariant under the right-translation $c \rightarrow c\hat{x}$ for $\forall \hat{x} \in \text{SU}(n)$. Similarly $F_l(c, \dot{c})$ is *left-invariant* because H_l is invariant under the left-translation $c \rightarrow \hat{x}c$ for $\forall \hat{x} \in \text{SU}(n)$. If there is no further restriction on F_α , there are at least two natural Finsler geometries, F_r or F_l , which may give different cost or length.

Finally, the left or right complexity of an operator ($\mathcal{C}_\alpha(\hat{O})$) is identified with the minimal length (or minimal cost) of the curves connecting $\hat{\mathbb{I}}$ and \hat{O} :

$$\mathcal{C}_\alpha(\hat{O}) := \min\{L_\alpha[c] \mid \forall c(s), c(0) = \hat{\mathbb{I}}, c(1) = \hat{O}\}. \quad (3.9)$$

We see that, even if we know the complexity of every infinitesimal operator (Eq. (3.6)), we have at least two different ways (left or right-way) to define the complexity of an operator and there is no a priori preferred choice among them. In order for the complexity of an operator to be a well-defined physical observable, this mathematical ambiguity should disappear naturally by some suitable physical considerations. In the following section we will show how this ambiguity is removed.

4 Symmetries of the complexity inherited from QFT symmetries

In the previous section, we have shown that the complexity can be computed by the minimal length of curves in *Finsler* geometry. We want to emphasize again that in our work the Finsler structure is *not assumed*, but it has been *derived* based on **G1**, **G2** and **G4**. This is a novel feature of our work compared to other works dealing with the Finsler geometry.

However, apart from the defining properties of the Finsler metric **F1–F3**, we don't know anything on $\tilde{F}(H_\alpha)$ so far. In this section, we will show there are constraints on $\tilde{F}(H_\alpha)$ if we take into account some symmetries of QFT. This is another important novel feature of our work compared to others. From here, we do not rely on properties of discrete systems or circuit models, which may be incompatible with QFT so may mislead us. We will directly deal with QFT and its symmetry properties and see what kind of constraints we can impose on $\tilde{F}(H_\alpha)$.

⁶ Strictly speaking, \tilde{F} and F_α are the Minkowski norm and the Finsler metric respectively if we make a further requirement that, in **F3**, the equality holds only when H_1 and H_2 are linearly dependent. However, this subtle mismatch is not important physically.

Note that such symmetry considerations are not necessary if we use “complexity” as a purely mathematical tool, for example, to study the “NP-completeness” and to analyze how complex an algorithm or a quantum circuit is. However, when we use the complexity to study real physical processes and try to treat the complexity as a basic physical variable hiding in physical phenomena, symmetries relevant to physical phenomena will be a necessary requirement.

In Sect. 4.1, by requiring unitary invariance for complexity we find

$$[\text{Independence of left/right generators}] \quad \tilde{F}(H_l) = \tilde{F}(H_r). \quad (4.1)$$

It means that the complexity does not depend on our choice of H_r or H_l . Therefore, we call this property ‘Independence of left/right generators’ of \tilde{F} . Recall that for a given curve, we may have two metrics, either $F_r(c, \dot{c}) = \tilde{F}(H_r)$ or $F_l(c, \dot{c}) = \tilde{F}(H_l)$. It is an inherent ambiguity mathematically but this ambiguity can be removed by imposing physical condition, unitary invariance. To support our result (4.1) we will present three more arguments in Sect. 4.2 and Appendix D. Note that the constraint (4.1) also implies the Finsler geometry is *bi-invariant*, meaning both right and left invariant.

In Sect. 4.3, by requiring the CPT symmetry,⁷ we obtain

$$[\text{reversibility}] \quad \tilde{F}(H_\alpha) = \tilde{F}(-H_\alpha). \quad (4.2)$$

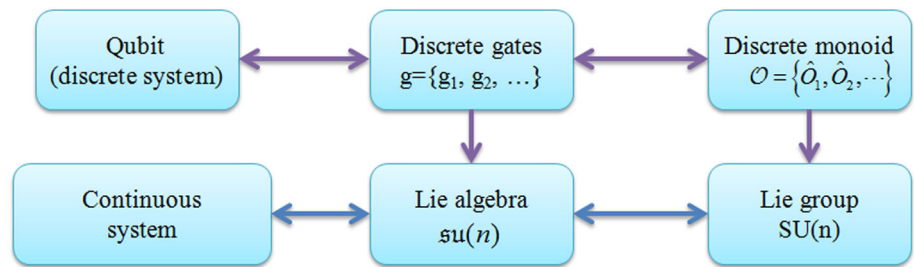
We call this property ‘reversibility’ of \tilde{F} following the mathematical literature, for example, [25]. In Appendix D, we will provide two more methods to support Eq. (4.2). Geometrically speaking, for a given path connecting A and B , it is the constraint Eq. (4.2) that gives the same length when we go from A to B and from B to A .

4.1 Independence of left/right generators from unitary invariance

In this subsection we consider the effect of the unitary invariance of the quantum field theory on the Finsler metric, cost, and complexity. Let us consider an arbitrary quantum field Φ with a Hilbert space \mathcal{H} and a vacuum $|\Omega\rangle$, which are collectively denoted by $\{\Phi, \mathcal{H}, |\Omega\rangle\}$. Its unitary partner is $\tilde{\Phi}(\vec{x}, t) := \hat{U}\Phi(\vec{x}, t)\hat{U}^\dagger$, $\tilde{\mathcal{H}} := \{\hat{U}|\psi\rangle \mid \forall |\psi\rangle \in \mathcal{H}\}$ and $|\tilde{\Omega}\rangle := \hat{U}|\Omega\rangle$, which are denoted by $\{\tilde{\Phi}, \tilde{\mathcal{H}}, |\tilde{\Omega}\rangle\}$.

⁷ The CPT symmetry is a theorem for local relativistic quantum field theories in Minkowski space-time. Here, C means ‘charge conjugation’, P ‘parity transformation’ (‘space inversion’), and T ‘time reversal’. This theorem states that the local Lorentz quantum field theories are invariant under the combined transformations of C, P, and T.

Fig. 6 Schematic diagram for the relation and difference between qubit systems and $SU(n)$ groups



where $C_\alpha(\tilde{g}; \tilde{O})$ denotes the complexity of $\forall \tilde{O} \in \tilde{O}$ measured by \tilde{g} .

In general, the gates set is not invariant under the unitary transformation, i.e.

$$\tilde{g} = \hat{U}g\hat{U}^\dagger \neq g, \quad (g \text{ is the gates set rather than a gate}) \quad (4.12)$$

so we will see that the complexity of \hat{O} and $\tilde{\hat{O}}$, measured by same gates set g , will not be the same, i.e.

$$C_\alpha(g; \hat{O}) = C_\alpha(\tilde{g}; \tilde{\hat{O}}) \neq C_\alpha(g; \tilde{\hat{O}}) \quad (4.13)$$

This shows that the complexity for qubit system will not be invariant under $\hat{O} \rightarrow \hat{U}\hat{O}\hat{U}^\dagger$ if we use the same gates set.

For $SU(n)$ groups, we still obtain an equation similar to Eq. (4.11),

$$C_\alpha(\mathfrak{su}(n); \hat{O}) = C_\alpha(\tilde{\mathfrak{su}}(n); \tilde{\hat{O}}). \quad (4.14)$$

However, unlike Eq. (4.12) in qubit systems, we have the following equality⁹

$$\tilde{\mathfrak{su}}(n) := \hat{U}\mathfrak{su}(n)\hat{U}^\dagger = \mathfrak{su}(n), \quad \forall \hat{U} \in SU(n). \quad (4.15)$$

Thus, we see that Eq. (4.14) implies that

$$C_\alpha(\mathfrak{su}(n); \hat{O}) = C_\alpha(\mathfrak{su}(n); \tilde{\hat{O}}), \quad \forall \hat{O} \in SU(n), \quad (4.16)$$

which means the complexity of $SU(n)$ group will be invariant under the adjoint transformation, $\hat{O} \rightarrow \hat{U}\hat{O}\hat{U}^\dagger$.

It is the difference between Eqs. (4.12) and (4.15) that leads the difference between qubit systems and $SU(n)$ regarding the invariance under adjoint transformations. Because Eq. (4.16) is valid also for any infinitesimal operator, it implies Eq. (4.6). This is another derivation of Eq. (4.6). We have presented two arguments to support the idea that the complexity of $SU(n)$ group should be invariant under adjoint

transformations. In Appendix D, we will give the third and the fourth arguments to support this conclusion.

To understand the validity of the adjoint invariance of the complexity, one useful question is the following: what will happen if we restrict our operators set to some subgroup of $SU(n)$? Let G to be a connected real subgroup and its Lie algebra to be \mathfrak{g} . In this case, we can still obtain the following equation under a general unitary transformation $\tilde{G} = \hat{X}G\hat{X}^{-1}$ and $\tilde{\mathfrak{g}} = \hat{X}\mathfrak{g}\hat{X}^{-1}$,

$$C_\alpha(\mathfrak{g}; \hat{O}) = C_\alpha(\tilde{\mathfrak{g}}; \tilde{\hat{O}}), \quad \forall \hat{X} \in SU(n), \quad \forall \hat{O} \in G. \quad (4.17)$$

If \mathfrak{g} is an ideal of $\mathfrak{su}(n)$, $\tilde{\mathfrak{g}} = \mathfrak{g}$ for all $\hat{X} \in SU(n)$. However, because the $\mathfrak{su}(n)$ is simple it does not have other ideals except for the trivial $\{0\}$. Thus, if we restrict the operators set to any real subgroup of $SU(n)$, the complexity may not be invariant under a adjoint transformation. For qubit systems such as a quantum circuit, the gates set is discrete, which can form only a subgroup of $SU(n)$. As $SU(n)$ group does not have non-trivial normal subgroup, the complexity for qubit systems is not invariant under the general adjoint transformation.

4.3 Reversibility of Finsler metric from the CPT symmetry

In this subsection we consider the effect of the CPT symmetry of the quantum field theory on the Finsler metric, cost, and complexity. Let us denote the CPT partner of $\Phi(\vec{x}, t)$ by $\bar{\Phi}(\vec{x}, t)$, i.e. $\bar{\Phi}(\vec{x}, t) := C \circ P \circ T[\Phi(\vec{x}, t)] = C[\Phi(-\vec{x}, -t)]$. By using Eq. (4.3), we have

$$\begin{aligned} \bar{\Phi}(\vec{x}, t) &= C \circ P \circ T[c(t)^\dagger \Phi(\vec{x}, 0)c(t)] \\ &= c(-t)^\dagger C[\Phi(-\vec{x}, 0)]c(-t) \\ &= c(-t)^\dagger \bar{\Phi}(\vec{x}, 0)c(-t), \end{aligned} \quad (4.18)$$

where we use the fact that $c(t)$ does not have charge and spatial variable \vec{x} . Thus, the evolution of the CPT partner is given by $\bar{c}(t) := c(-t)$. Given the CPT symmetry of the theory, it is natural to assume that the costs of $c(t)$ and $\bar{c}(t)$ are the same, i.e.,

$$L_\alpha[c(t)] = L_\alpha[\bar{c}(t)]. \quad (4.19)$$

⁹ The equality means the equality between two sets, i.e. $\tilde{\mathfrak{su}}(n)$ and $\mathfrak{su}(n)$ contain the same elements.

Similarly to the unitary symmetry case in Sect. 4.1, as a way to understand the general structure of \tilde{F} , we consider a special curve, the time evolution given by an arbitrary constant generator H_α . Because the generators of $\tilde{c}(s)$ are given by $\tilde{H}_\alpha = -H_\alpha$, Eq. (4.19) reads, by the definition of the cost Eq. (3.7),

$$\int_0^1 \tilde{F}(H_\alpha) dt = \int_0^1 \tilde{F}(-H_\alpha) dt. \quad (4.20)$$

which implies

$$\tilde{F}(H_\alpha) = \tilde{F}(-H_\alpha). \quad (4.21)$$

Path-reversal symmetry If we combine the result of the CTP symmetry and the unitary symmetry, Eqs. (4.21) and (4.9) respectively, one can prove the “path-reversal symmetry” for an arbitrary curve:

$$L_\alpha[c] = L_\alpha[c^{-1}], \quad \forall c(s). \quad (4.22)$$

Note that in general $c^{-1}(s) := [c(s)]^{-1}$ is not the curve generated by $-H_\alpha(s)$ when the curve $c(s)$ is generated by $H_\alpha(s)$. For example, for the right-invariant case, we can show

$$\tilde{F}(H_r(c^{-1})) = \tilde{F}(-c^{-1}H_r(c)c) = \tilde{F}(H_r(c)), \quad (4.23)$$

which gives Eq. (4.22). Here, we used $H_r(c^{-1}) = (dc^{-1}/ds)c = -c^{-1}(\dot{c}c^{-1})c = -c^{-1}H_r(c)c$ in the first equality and Eqs. (4.21) and (4.9) in the second equality. In fact, the reverse also holds, i.e. Eq. (4.22) implies Eqs. (4.21) and (4.9). First, by considering the special case $c = e^{Hs}$ with a constant H , Eqs. (4.21) can be derived from Eq. (4.22). Then, we end up with $L_r[c] = L_l[c]$, which implies Eq. (4.9) by the same logic in Sect. 4.1.

Thus, we have the following equivalence between the path-reversal symmetry and the adjoint invariance with the reversibility of the Finsler metric:

$$\begin{aligned} \forall c(s), \quad L_\alpha[c] = L_\alpha[c^{-1}] \\ \Leftrightarrow \begin{cases} \tilde{F}(H_\alpha) = \tilde{F}(\hat{U}H_\alpha\hat{U}^\dagger), & \forall H_\alpha, \quad \forall \hat{U} \in \text{SU}(n); \\ \tilde{F}(H_\alpha) = \tilde{F}(-H_\alpha), & \forall H_\alpha. \end{cases} \end{aligned} \quad (4.24)$$

The path-reversal symmetry also can be justified by other ways, for example, based on the inverse symmetry of the relative complexity or the “ket-world”-“bra-world” symmetry. These two arguments are presented in Appendix D in detail. They may server other supporting evidence for Eqs. (4.21) and (4.9) because of the equivalence in Eq. (4.24).

5 Complexity of $\text{SU}(n)$ operators

5.1 Finsler metric of $\text{SU}(n)$ operators

From here we will drop the indexes α, r , and l based on Eq. (4.10). We have found two constraints Eqs. (4.9) and (4.21) by considering basic physical symmetries. These constraints (plus **G3**) turn out to be strong enough to determine the Finsler metric in the operator space of any $\text{SU}(n)$ groups as follows

$$F(c(s), \dot{c}(s)) = \tilde{F}(H(s)) = \lambda \text{Tr} \sqrt{H(s)H(s)^\dagger}, \quad (5.1)$$

where $H(s) = H_r(s)$ or $H_l(s)$ for the curve $c(s)$ and λ is arbitrary constant. (see Appendix C for a proof.)

The Finsler metric Eq. (5.1) is representation independent and to find the explicit Finsler metric tensor for the $\text{SU}(n)$ group we consider the fundamental representation. An arbitrary generator $H(s)$ can be expanded as

$$H(s) = iH^a(s)T_a, \quad H^a(s) \in \mathbb{R}, \quad (5.2)$$

where $\{T_a\}$ are basis of $\mathfrak{su}(n)$ in the fundamental representation with the following property.

$$T_a T_b = \frac{1}{2n} \delta_{ab} \mathbb{I} + \frac{1}{2} \sum_{c=1}^{n^2-1} (i f_{ab}^c + d_{ab}^c) T_c, \quad (5.3)$$

where f_{ab}^c is the structure constants antisymmetric in all indices, while d_{ab}^c , which is nonzero only when $n > 2$, is symmetric in all indices and traceless. Thus, Eq. (5.1) reads

$$F(c, \dot{c}) = \frac{1}{\sqrt{2n}} \text{Tr} \sqrt{H^a(s)H^b(s)[\delta_{ab} \mathbb{I} + n d_{ab}^c T_c]}, \quad (5.4)$$

with

$$H^a(s) = 2\text{Tr}[H(s)T^a] = 2\text{Tr}[\dot{c}(s)c^{-1}(s)T^a], \quad (5.5)$$

and

$$T^a := T_b \delta^{ab}. \quad (5.6)$$

Without loss the generality, we have set $\lambda = 1$.

For $n = 2$, $d_{ab}^c = 0$ so Eq. (5.4) is simplified to

$$\begin{aligned} F(c, \dot{c}) &= \frac{1}{2} \text{Tr} \sqrt{H^a(s)H^b(s)\delta_{ab} \mathbb{I}} = 2\sqrt{H^a(s)H^b(s)\delta_{ab}} \\ &= \sqrt{\text{Tr}[\dot{c}(s)c^{-1}(s)T^a]\text{Tr}[\dot{c}(s)c^{-1}(s)T^b]\delta_{ab}}, \end{aligned} \quad (5.7)$$

where Eq. (5.5) is used. It contains only quadratic terms of \dot{c} so it gives a Riemannian geometry. For $n > 2$

$$\begin{aligned}
 F(c, \dot{c}) &= \frac{2}{\sqrt{2n}} \text{Tr} \sqrt{\text{Tr}[H(s)T^a] \text{Tr}[H(s)T^b] [\delta_{ab} \hat{\mathbb{I}} + n d_{ab}^c T_c]} \\
 &= \frac{2}{\sqrt{2n}} \text{Tr} \sqrt{\text{Tr}(\dot{c} c^{-1} T^a) \text{Tr}(\dot{c} c^{-1} T^b) [\delta_{ab} \hat{\mathbb{I}} + n d_{ab}^c T_c]}.
 \end{aligned}
 \quad (5.8)$$

As $\hat{\mathbb{I}}$ and T_c are $n \times n$ matrixes, the line element in Eq. (5.8) is not the quadratic form of \dot{c} so it is not Riemannian but Finsler.

5.2 Geodesics and complexity of $SU(n)$ operators

Even though we have the precise Finsler metric, to compute the complexity, we still have to find a geodesic path as shown in (3.9). This minimization procedure is greatly simplified thanks to *bi-invariance* implied by (4.10). The bi-invariance means the geometry is both right and left invariant. It has been shown that, in bi-invariant Finsler geometry, the curve $c(s)$ is a geodesic if and only if there is a constant generator $H(s) = \bar{H}$ such that [34,35]

$$\dot{c}(s) = \bar{H}c(s) \text{ or } c(s) = \exp(s\bar{H}). \quad (5.9)$$

With the condition $\hat{O} = c(1) = \exp(\bar{H})$, we can solve \bar{H} formally $\bar{H} = \ln \hat{O}$. The logarithm of a unitary operator always exists but may not be unique (theorem 1.27 in Ref. [36]). Because \bar{H} is constant, from Eqs. (3.7),

$$L[c] = \tilde{F}(\bar{H}) = \text{Tr} \sqrt{\bar{H} \bar{H}^\dagger}. \quad (5.10)$$

Finally, the complexity of \hat{O} in Eq. (3.9) is given by

$$\mathcal{C}(\hat{O}) = \min\{\text{Tr} \sqrt{\bar{H} \bar{H}^\dagger} \mid \forall \bar{H} = \ln \hat{O}\}, \quad (5.11)$$

The minimization ‘min’ in (3.9) in the sense of ‘geodesic’ is already taken care of in (5.9). Here ‘min’ means the minimal value due to multi-valuedness of $\ln \hat{O}$.

For example, let us consider the $SU(2)$ group in the fundamental representation. For any operator $\hat{O} \in SU(2)$, there is a unit vector \vec{n} and a real number θ such that,

$$\hat{O} = \exp(i\theta \vec{n} \cdot \vec{\sigma}) = \hat{\mathbb{I}} \cos \theta + i(\vec{n} \cdot \vec{\sigma}) \sin \theta, \quad (5.12)$$

where $\vec{\sigma} := (\sigma_x, \sigma_y, \sigma_z)$ stands for three Pauli matrixes. Because $\ln \hat{O} = i\theta_m \vec{n} \cdot \vec{\sigma}$ with

$$\theta_m = \arccos[\text{Tr}(\hat{O})/2] + 2m\pi, \quad (5.13)$$

for $\forall m \in \mathbb{N}$, the complexity of \hat{O} is given by

$$\mathcal{C}(\hat{O}) = 2 \arccos[\text{Tr}(\hat{O})/2], \quad (5.14)$$

where $\bar{H} \bar{H}^\dagger = \theta_m^2 \hat{\mathbb{I}}$ is used.

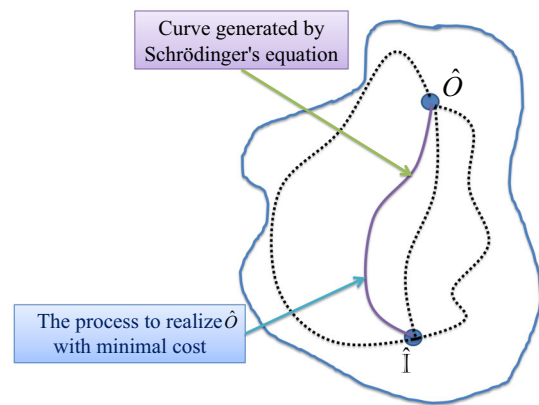


Fig. 7 The schematic figure showing the curve of a time evolution operator $\hat{U}(t)$. To reach the same target operator at $t = t_0$, there are many different curves. The curve given by Schrödinger's equation coincides with the curve of complexity

6 Minimal cost principle

The fact that the process of complexity is generated by a constant generator allows an interesting interpretation of the Schrödinger's equation. In a quantum system with a time-independent Hamilton \mathbb{H} , the time evolution of the quantum state $|\psi(t)\rangle$ is given by a time evolution operator $\hat{U}(t)$, i.e. $|\psi(t)\rangle = \hat{U}(t)|\psi(0)\rangle$, where $\hat{U}(t)$ satisfies the Schrödinger's equation,

$$\frac{d}{dt} \hat{U}(t) = -i\hbar^{-1} \mathbb{H} \hat{U}(t), \quad \hat{U}(0) = \hat{\mathbb{I}}. \quad (6.1)$$

Comparing with Eq. (5.9), we may say that $\hat{U}(t)$ is a geodesic generated by $-i\hbar^{-1} \mathbb{H}$. Thus, the time-evolution operator $\hat{U}(t)$ is along the curve of minimal (at least locally minimal) complexity.

Now let us re-consider the problem in the following way. Assume that after a short time $t = t_0$, the time-evolution operator becomes $\hat{U}(t_0) = \hat{O}$. As there are many different curves which connect the $\hat{\mathbb{I}}$ and \hat{O} (see the Fig. 7), how can we find the real curve $\hat{U}(t)$ during $t \in (0, t_0)$? One answer is that we assume the time evolution operator will obey the Schrödinger's equation (6.1). Alternatively, we may replace the Schrödinger's equation with the following principle:

Minimal cost principle: For isolated systems, the time-evolution operator $\hat{U}(t)$ will be along the curve to reach the target operator so that the cost during this process is locally minimal, i.e. the evolution curve will make the following integral to be locally minimal:

$$L[\hat{U}(t)] = \int_0^{t_0} \text{Tr} \sqrt{\dot{\hat{U}}(t) [\dot{\hat{U}}(t)]^\dagger} dt, \quad \hat{U}(0) = \hat{\mathbb{I}}, \quad \hat{U}(t_0) = \hat{O}, \quad (6.2)$$

where we used Eq. (5.1), where $H \rightarrow \dot{\hat{U}}(t) [\dot{\hat{U}}(t)]^\dagger$.

As a result, the time evolution operator $\hat{U}(t)$ will satisfy Eq. (6.1). In other words, by this principle, the Schrödinger's equation is not the first-principle but a consequence of the complexity principle.¹⁰

7 Comparison with the complexity for K -local qubit systems

For a better understanding of the novel aspects of our work compared to previous research it is useful to compare our complexity and the complexity for K -qubit systems [27]. In particular, our complexity is bi-invariant but the complexity geometry in Ref. [27] is only right-invariant. At first glance, two theories may look different, but the difference in complexity turns out to be little and most of physical results in Ref. [27] can also appear in our theory.

For K -qubit system, the operators form a $SU(2^K)$ group and can be generated only by a right Hamiltonian

$$H_r(s) = \sum_I H_r^a(s) T_a, \quad (7.1)$$

where $a = 1, 2, 3, \dots, 4^K - 1$ and T_a is a series of generalized Pauli operators which can span the Lie algebra $\mathfrak{su}(2^K)$. In Ref. [27], for the $SU(2^K)$ group, the following Riemannian metric was proposed

$$dl^2 = d\Omega_a \mathcal{I}^{ab} d\Omega_b, \quad (7.2)$$

where

$$d\Omega_a := i\text{Tr}(dc^\dagger(s) T_a c(s)). \quad (7.3)$$

Here, the matrix \mathcal{I}^{ab} should be chosen as a block diagonal matrix with one block corresponding to the unpenalized k -local directions, and the other block corresponding to the directions T_a containing more than k single qubit operators. Note that, for given a and b , $i T_a$ is matrix-valued vector in the representation space of $\mathfrak{su}(2^K)$ and \mathcal{I}^{ab} is a real number.

The curve $c(s)$ is assumed to be generated by a right generator $H_r(s)$ such that

$$dc(s) = H_r(s) c(s) ds, \quad H_r(s) \in \mathfrak{su}(2^K). \quad (7.4)$$

Taking Eq. (7.4) into Eq. (7.3), we have

$$\begin{aligned} d\Omega_a &= \text{Tr}[c^\dagger(s) H_r(s) T_a c(s)] ds \\ &= \text{Tr}[H_r(s) T_a] ds, \end{aligned} \quad (7.5)$$

so Eq. (7.2) becomes

¹⁰ If the Hamiltonian is not static, the complexity curve and the curve generated by Hamiltonian will be different. In this case, we may extend the system to include back-reaction of the system on its background, rendering the total Hamiltonian static.

$$dl^2 = \text{Tr}[H_r(s) T_a] \text{Tr}[H_r(s) T_b] \mathcal{I}^{ab} ds^2, \quad (7.6)$$

which yields the following Finsler (or just Riemannian) metric

$$F(c, \dot{c}) = \sqrt{\text{Tr}[H_r(s) T_a] \text{Tr}[H_r(s) T_b] \mathcal{I}^{ab}}. \quad (7.7)$$

It has two differences compared with our result (5.8). One is that the “Tr” is inside the square root. The other is that there is a matrix structure \mathcal{I}^{ab} which is not determined by the Lie algebra uniquely. In the following, we make comparisons between two complexities based on two different Finsler metrics.

- (1) In our paper, the only basic assumptions are **G1–G4**. All conclusions such as Finsler geometry and the Finsler metric Eq. (5.1) are the results of these four assumptions and fundamental symmetries of QFTs. In Ref. [27], the Riemannian geometry and metric (7.6) in K -qubit system were proposed directly as the basic assumptions.
- (2) The complexity given by Eq. (7.7) satisfies our axioms **G1** and **G2** but breaks **G3**. It can be shown by considering a simple case, $\mathcal{I}^{ab} = \delta^{ab}$, which corresponds to bi-invariant case without any penalty (isotropy). In this case, the complexity of the operator $\hat{O} = \exp(H)$ is given by $F(H)$ because the geodesic is generated by a constant generator (due to bi-invariance) i.e.

$$C(\hat{O}) = F(H). \quad (7.8)$$

By using Eqs. (7.8) and (7.7) we have

$$\begin{aligned} C(\hat{O}_1 \oplus \hat{O}_2) &= \sqrt{C^2(\hat{O}_1 \oplus \hat{\mathbb{I}}_2) + C^2(\hat{\mathbb{I}}_1 \oplus \hat{O}_2)} \\ &= \sqrt{C^2(\hat{O}_1) + C^2(\hat{O}_2)}, \end{aligned} \quad (7.9)$$

and in more general cases,

$$C\left(\bigoplus_i \hat{O}_i\right) = \sqrt{\sum_i p_i [C(\hat{O}_i)]^2} \neq \sum_i C(\hat{O}_i), \quad (7.10)$$

where p_i is a weighting factor if $\mathcal{I}^{ab} \neq \delta^{ab}$. This means that the total complexity of *parallel* operations is not the sum of the complexity of the individual operations, so breaks **G3**. We want to stress again that **G3** is a very natural requirement that has not been considered in previous research.

- (3) For the same curve in $SU(n)$ group, the tangent vector at a point is unique but the generator is not. It can be a left or right generator. We admit of two ways (left generator or right generator) but Ref. [27] considers only one way (right generator). As there is no reason to assume that

physics favors “left world” or “right world”, a simple and natural possibility is that two generators yield the same complexity. It is the case that is realized in our framework unlike the complexities in [27] and Nielsen’s original works [20–22] where the left and right generator will give different complexities.

One may argue that Eq. (7.7) could be valid only for the right generator and, for the left generator, there might be another left-invariant metric which has different penalty \mathcal{I}^{ab} and could give out the same curve length. In our upcoming work Ref. [33], we will show this is possible only if the geometry is bi-invariant. This gives us another argument that the complexity for $SU(n)$ group should be bi-invariant.

- (4) When Ref. [27] discusses some particular physical situations such as “particle on complexity geometry”, “complexity equals to action” and “the complexity growth”, the authors restricted the generators in a “ k -local” subspace \mathfrak{g}_k and assumed $\mathcal{I}_{ab}|_{\mathfrak{g}_k} = \delta_{ab}$ (see the Sect. IV.C and V in Ref. [27] for detailed explanations). As a result, the geodesics in the sub-manifold generated by \mathfrak{g}_k are also given by constant generators, which are the same as our bi-invariant Finsler geometry. The lengths of such geodesics in Ref. [27] and in our paper are only different by multiplicative factors, which implies that all the results given by “ k -local” subspace can also appear similarly in our bi-invariant Finsler geometry.

Moreover, in order to obtain the complexity geometry as was proposed in Ref. [27], we can choose some two-dimensional sub-manifold in $SU(n)$ geometry. As described in our upcoming paper [33], by using the Gauss–Codazzi equation, we can show that the sub-manifold can have negative induced sectional curvature somewhere despite the $SU(n)$ geometry is positively curved. So it satisfies the same properties as shown in Ref. [27], where the sectional curvature is made negative near the identity by choosing an appropriate penalty factor.

8 Discussion and outlook

In this paper we proposed four basic axioms for the complexity of operators: nonnegativity (**G1**), series decomposition rule (triangle inequality) (**G2**), parallel decomposition rule (**G3**) and smoothness (**G4**). Combining these four axioms and basic symmetries in QFT, we have obtained the complexity of the $SU(n)$ operator without ambiguity: Eq. (5.11). In our derivation the bi-invariance of the Finsler structure plays an important role and this bi-invariance is a natural implication of the symmetry in QFTs rather than an artificial assumption. Our logical flows are shown in the Fig. 1.

We argue the importance of the bi-invariance in four ways based on: (i) the unitary-invariance of QFTs (Sect. 4.1); (ii) the nature of continuous operators rather than discrete ones (Sect. 4.2); (iii) inverse-invariance of the relative complexity (Appendix D.1); and (iv) the “ket-world” - “bra-world” equivalence (Appendix D.2). The bi-invariance here is different from the only right-invariance for qubit systems [20–22, 27]. We clarify the differences and similarities of our proposal (bi-invariance) from previous researches (only right-invariance) in Sects. 4.2 and 7. It can be shown that most of results in only right-invariant complexity geometry proposed by Ref. [27] can also appear in our framework. We want to emphasize that the complexity cannot be a well-defined physical observable in general finite dimensional systems if the complexity geometry is not bi-invariant.

Thanks to the bi-invariance of the Finsler metric the process of minimal cost (complexity) is generated by a constant generator. This observation leads us to make a novel interpretation for the Schrödinger’s equation: the quantum state evolves by the process of minimizing “computational cost,” which we call “minimal cost principle.”

As an application of the complexity of the $SU(n)$ operator, the complexity between two states described by density matrices ρ_1 and ρ_2 may be defined naturally as

$$\mathcal{C}(\rho_1, \rho_2) := \min\{\mathcal{C}(\hat{O}) \mid \rho_2 = \hat{O}\rho_1\hat{O}^\dagger, \forall \hat{O} \in SU(n)\}. \quad (8.1)$$

In our forthcoming paper [19] our proposal turns out to be general enough to include and unify other recent developments for the complexity in QFT: cMERA tensor network [37–40], Fubini-Study metric [41] and path-integral method [42, 43]. Furthermore, it can be shown that our proposal also correctly reproduces the holographic complexity for thermofield double state (TFD).

In a more general context, geometrizing the complexity in continuous operators sets amounts to giving positive homogeneous norms in some Lie algebras. Our paper deals with only $SU(n)$ group so we gives the norm for Lie algebra $\mathfrak{su}(n)$. For more general Lie algebra \mathfrak{g} , though we cannot determine the norm uniquely, it is natural that such a norm is determined only by the properties of \mathfrak{g} , for example the structure constants, without any other extra information. As in general relativity where the spacetime metric is determined by matter distribution through Einstein’s equations, can we find any physical equation to determine this norm?

Acknowledgements The work of K.-Y. Kim was supported by Basic Science Research Program through the National Research Foundation of Korea (NRF) funded by the Ministry of Science, ICT and Future Planning (NRF-2017R1A2B4004810) and GIST Research Institute (GRI) Grant funded by the GIST in 2018 and 2019. C. Y. Zhang is supported by National Postdoctoral Program for Innovative Talents 938 BX201600005 and China Postdoctoral Science Foundation. C. Niu is

supported by the Natural Science Foundation of China under Grant No. 11805083.

Data Availability Statement This manuscript has no associated data or the data will not be deposited. [Authors' comment: Data sharing not applicable to this article as no datasets were generated or analysed during the current study.]

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A brief introduction to the Finsler manifolds

This appendix introduces some basic concepts in the Finsler geometry. It is not meant to be a complete or rigorous introduction. The readers can find more details in textbooks such as Refs. [23–25] and a physics friendly introduction in Ref. [26].

A.1 Fundamentals

Suppose that M is an n -dimensional smooth manifold and TM is its tangent bundle. Each element of TM is given by a pair (x^i, v^i) , where $x^i \in M$ and $v^i \in T_x M$. In this appendix, x, y will be used to stand for the points in M and u, v will be used to stand for the tangent vectors at some points in M . For convenience, sometimes their indices will be dropped if there is no ambiguity.

A Finsler metric of M is a function $F : TM \mapsto [0, \infty)$ such that:

- (1) F is smooth on $TM \setminus \{\cdot, 0\}$;
- (2) $F(x, \lambda v) = \lambda F(x, v)$ for arbitrary $\lambda > 0$;
- (3) The fundamental tensor (metric)

$$g_{ij}(x, v) := \frac{1}{2} \frac{\partial^2 F^2(x, v)}{\partial v^i \partial v^j} \quad (\text{A.1})$$

is positive definite when $v \neq 0$.

The manifold M with a Finsler metric F is called a *Finsler manifold*. The requirement (3) can be relaxed in physics. If $g_{ij}(x, v)$ has negative eigenvalues but no zero eigenvalue, then (M, F) is called a *pseudo-Finsler manifold*; if $g_{ij}(x, v)$ has only one negative eigenvalue and has no zero eigenvalue, then (M, F) is called *pseudo-Finsler spacetime*. In this paper, we only consider the case that $g_{ij}(x, v)$ is positive definite.

We want to emphasize that the metric tensor $g_{ij}(x, v)$ is defined in the tangent bundle TM rather than the manifold M , which is the essential difference between a general Finsler

manifold and Riemannian manifold. The relation between the Finsler metric F and the metric tensor g_{ij} reads

$$F^2(x, v) = g_{ij}(x, v) v^i v^j. \quad (\text{A.2})$$

For a curve $x(t) = c(t)$ in M , its line element is given by

$$ds := \sqrt{g_{ij}(c, \dot{c}) \dot{c}^i \dot{c}^j} dt = F(c, \dot{c}) dt, \quad (\text{A.3})$$

and the curve length for $t \in (0, 1)$ is

$$L[c(t)] := \int_0^1 F(c, \dot{c}) dt = \int_0^1 \sqrt{g_{ij}(c, \dot{c}) \dot{c}^i \dot{c}^j} dt. \quad (\text{A.4})$$

In a Riemannian case, the curves $x(t) = c(t)$ and $x(t) = c(1-t)$ with $t \in (0, 1)$ have the same length. However, this is no longer true for general Finsler geometries because

$$\begin{aligned} L[c(1-t)] &:= \int_0^1 \sqrt{g_{ij}(c, -\dot{c}) \dot{c}^i \dot{c}^j} dt \\ &\neq \int_0^1 \sqrt{g_{ij}(c, \dot{c}) \dot{c}^i \dot{c}^j} dt = L[c(t)]. \end{aligned} \quad (\text{A.5})$$

If the metric $g_{ij}(x, v)$ is independent of v , then a Finsler manifold becomes a Riemannian manifold. To describe how much different a Finsler manifold is from a Riemannian manifold, we can introduce the Cartan's tensor,¹¹

$$A_{ijk}(x, v) := \frac{1}{2} \frac{\partial}{\partial v^k} g_{ij}(x, v) = \frac{1}{4} \frac{\partial^3 F^2(x, v)}{\partial v^i \partial v^j \partial v^k}, \quad (\text{A.6})$$

which is a fully symmetric covariant tensor.

Thus, a Finsler manifold is a Riemannian manifold if and only if its Cartan tensor is zero. The Cartan form (η_k) is defined by the contraction of Cartan tensor and metric:

$$\eta_k(x, v) := g^{ij}(x, v) A_{ijk}(x, v), \quad (\text{A.7})$$

where $g^{ij}(x, v)$ is the inverse of $g_{ij}(x, v)$. The Cartan tensor and Cartan form have three important properties:

$$\begin{aligned} \eta_k(x, v) &= \frac{\partial}{\partial v^k} \ln \sqrt{\det(g_{ij})}, \\ u = v &\Rightarrow A_{ijk}(x, v) u^j = 0, \\ \eta_k &= 0 \Leftrightarrow A_{ijk} = 0. \end{aligned} \quad (\text{A.8})$$

Example

In order to give readers a “feeling” about the Finsler geometry, let us show a simple example. Assume $a_{ij}(x)$ to be a positive definite tensor and a 1-form $b_i(x)$ satisfying

¹¹ In some textbook, there is an overall factor F in the definition so that the Cartan tensor is scale invariant under the transformation $v \rightarrow \lambda v$.

$b_i(x)b_j(x)a^{ij}(x) < 1$.¹² With $\alpha(x, v) = \sqrt{a_{ij}(x)v^i v^j}$ and $\beta(x, v) = b_i(x)v^i$,

$$F(x, v) = \alpha(x, v) + \beta(x, v) \quad (\text{A.9})$$

is a Finsler metric. If $b_i(x) = 0$, this is just a Riemannian geometry. By the definition (A.1), the metric tensor reads

$$g_{ij}(x, v) = \frac{F}{\alpha} \left(a_{ij} - \frac{v_i v_j}{\alpha} \right) + \left(\frac{v_i}{\alpha} + b_i \right) \left(\frac{v_j}{\alpha} + b_j \right), \quad (\text{A.10})$$

where $v_i := a_{ij}v^j$. One can check $g_{ij}(x, v)v^i v^j = F(x, v)^2$ and the Cartan form is

$$\eta_k(x, v) = \frac{n+1}{2F} \left(b_k - \frac{\beta}{\alpha^2} v_k \right). \quad (\text{A.11})$$

It can be shown that $\forall v, \eta_k(x, v) = 0 \Leftrightarrow b_k(x) = 0 \Leftrightarrow g_{ij}(x, v) = a_{ij}(x)$. The Finsler geometry (A.9) is called Randers geometry, which was first proposed by physicist Gunnar Randers regarding the uni-direction of time in general relativity [44].

In Finsler geometry, we can also define some geometric quantities such as connection and curvature. As this paper will not consider such quantities, we will not introduce them. The readers can refer to some textbooks, e.g., Refs. [23–25].

A.2 Invariant Finsler geometries for a Lie group

Now let us consider the case the manifold $M = G$ is a Lie group with the identity $\hat{\mathbb{I}}$. The Lie algebra is denoted by \mathfrak{g} , which is also the tangent space of identity $\hat{\mathbb{I}}$, i.e., $\mathfrak{g} = T_{\hat{\mathbb{I}}}G$. Because this manifold has group structure, i.e., the product rule, we can pick out a group-relevant special Finsler metric.

In a Lie group G , we can define a left translation $\mathcal{L}_x : G \mapsto G$ such that $\forall y \in G, \mathcal{L}_x(y) = xy$. As the left translation is the isomorphism of the group G , it is natural to expect that it is also the isometry of G . We call a Finsler manifold (G, F) is left-invariant if the left translation is an isometry, i.e.,

$$\forall x \in G, \quad F(y, v) = F(\mathcal{L}_x(y), \mathcal{L}_{x*}(v)), \quad (\text{A.12})$$

where \mathcal{L}_{x*} is the push-forward map induced by \mathcal{L}_x and $\mathcal{L}_{x*}(v) = xv$. Thus, the left-invariant Finsler metric will satisfy

$$\forall x \in G, \quad F(y, v) = F(xy, xv) \quad (\text{A.13})$$

Particularly, by taking $x = y^{-1}$ we have

$$\forall y \in G, \quad F(y, v) = F(\hat{\mathbb{I}}, y^{-1}v). \quad (\text{A.14})$$

¹² This condition is necessary to have a positive definite metric tensor.

Similarity, we can define the right translation and right-invariant Finsler geometry. The right-invariant Finsler geometry satisfies

$$\forall y \in G, \quad F(y, v) = F(\hat{\mathbb{I}}, vy^{-1}). \quad (\text{A.15})$$

Note that both left-invariant and right-invariant Finsler geometry are determined by the group product rule and the Finsler metric at the identity. A Finsler manifold is called bi-invariant if the Finsler geometry is both left and right invariant.

The Finsler metric at the identity can be used as a norm at the Lie algebra \mathfrak{g} . For a given Finsler metric at the identity, we have two natural ways to obtain the Finsler geometry according to Eqs. (A.14) and (A.15). The left and right invariant Finsler metric F_l and F_r can be determined uniquely

$$F_l(x, v) = \tilde{F}(x^{-1}v), \quad F_r(x, v) = \tilde{F}(vx^{-1}), \quad (\text{A.16})$$

where \tilde{F} denotes a norm for Lie algebra \mathfrak{g} . Thus, if we know how to define the norm \tilde{F} for a Lie algebra \mathfrak{g} , we can obtain two natural Finsler geometries for the Lie group G . Such two Finsler geometries are the same only when the norm \tilde{F} satisfies

$$\forall v \in \mathfrak{g}, \forall x \in G, \quad \tilde{F}(v) = \tilde{F}(xvx^{-1}). \quad (\text{A.17})$$

In this case, we can find $F_l = F_r$ and so the Finsler geometry is bi-invariant.

If we know the norm \tilde{F} for Lie algebra \mathfrak{g} , then for arbitrary curve $c(s)$ with $s \in (0, 1)$, we have two ways to compute its length

$$L_\alpha[c] := \int_0^1 F_\alpha(c(s), \dot{c}(s)) ds. \quad (\text{A.18})$$

If the curve is generated by the left generator $H_l(s)$ and right generator $H_r(s)$ defined as

$$c(s)H_l(s) = \dot{c}(s), \quad H_r(s)c(s) = \dot{c}(s), \quad (\text{A.19})$$

we see that Eq. (A.18) reads

$$L_\alpha[c] = \int_0^1 \tilde{F}(H_\alpha(s)) ds. \quad (\text{A.20})$$

In general, we have two different lengths for the same curve if we obtain the norm \tilde{F} of the Lie algebra \mathfrak{g} . Such two lengths are the same if \tilde{F} satisfies Eq. (A.17), i.e., the Finsler geometry is bi-invariant.

B Proof for the emergence of the Finsler metrics

In this appendix, we first show three properties **F1–F3** of the Minkowski norm, \tilde{F} , implied by the general axioms **G1**, **G2** and the assumption **G4**. **F1–F3** readily show that the Finsler geometry $F_\alpha(c, \dot{c})$ emerges. As we have clarified below Eq. (3.6), for an infinitesimal operator $\delta\hat{O} = \exp(H\varepsilon)$ with $|\varepsilon| \ll 1$, the indexes $\alpha = \{r, l\}$ are not necessary, so we omit it not to clutter notations.

F1: $\forall H, \tilde{F}(H) \geq 0$ and $\tilde{F}(H) = 0 \Leftrightarrow H = 0$.

Proof ① If $H = 0$, $\exp(Hs) = \mathbb{I}$ so $\mathcal{C}(\exp(Hs)) = 0$ for $\forall s \in [0, 1]$ by **G1**. Thus, $\tilde{F}(H) = \lim_{s \rightarrow 0^+} \mathcal{C}(\exp(Hs))/s = 0$ by (3.6). \square

② If $H \neq 0$, $\exists \lambda > 0$ such that $\exp(H\lambda) \neq \mathbb{I}$ so by **G1**

$$0 < \mathcal{C}(\exp(H\lambda)) = \mathcal{C}((\exp(H\lambda/N))^N),$$

where $N \in \mathbb{N}^+$. By **G2** we have the following inequality

$$\mathcal{C}[(\exp(H\lambda/N))^N] \leq N\mathcal{C}[\exp(H\lambda/N)].$$

Thus,

$$0 < \lim_{N \rightarrow \infty} \mathcal{C}[(\exp(H\lambda/N))^N] \leq \lim_{N \rightarrow \infty} N\mathcal{C}[\exp(H\lambda/N)]. \quad (\text{B.1})$$

With $\varepsilon := \lambda/N$, Eq. (B.1) means

$$0 < \lim_{\varepsilon \rightarrow 0^+} \lambda \frac{\mathcal{C}[\exp(H\varepsilon)]}{\varepsilon} = \lambda \tilde{F}(H), \quad (\text{B.2})$$

where \tilde{F} is defined in (3.6). Thus, $\tilde{F}(H) > 0$. \square

F2: $\forall \lambda \in \mathbb{R}^+$, $\tilde{F}(\lambda H) = \lambda \tilde{F}(H)$.

Proof By (3.6), for an arbitrary generator H and infinitesimal parameter $\varepsilon > 0$

$$\mathcal{C}(\exp(H\varepsilon)) = \tilde{F}(H)\varepsilon + \mathcal{O}(\varepsilon^2). \quad (\text{B.3})$$

For an arbitrary $\lambda > 0$,

$$\mathcal{C}[\exp(H \cdot \lambda\varepsilon)] = \mathcal{C}[\exp(\lambda H \cdot \varepsilon)], \quad (\text{B.4})$$

which implies $\tilde{F}(\lambda H) = \lambda \tilde{F}(H)$ by Eq. (B.3). \square

F3: $\forall H_1 \neq 0$ and $\forall H_2 \neq 0$, $\tilde{F}(H_1) + \tilde{F}(H_2) \geq \tilde{F}(H_1 + H_2)$

Proof By **G2**, for arbitrary generators H_1 and H_2

$$\mathcal{C}(\exp(H_1\varepsilon)) + \mathcal{C}(\exp(H_2\varepsilon)) \geq \mathcal{C}(\exp(H_1\varepsilon)\exp(H_2\varepsilon)).$$

It yields, up to order $\mathcal{O}(\varepsilon^2)$,

$$\mathcal{C}(\exp(H_1\varepsilon)) + \mathcal{C}(\exp(H_2\varepsilon)) \geq \mathcal{C}[\exp((H_1 + H_2)\varepsilon)], \quad (\text{B.5})$$

which implies $\tilde{F}(H_1) + \tilde{F}(H_2) \geq \tilde{F}(H_1 + H_2)$ by Eq. (B.3). \square

By the relation between F_α and \tilde{F} in Eq. (3.8), we can also prove that F_α satisfies the following properties for $\forall \hat{O} \in \text{SU}(n)$ and two arbitrary tangent vectors V, W at \hat{O} :

F1': $F_\alpha(\hat{O}, V) \geq 0$ and $F_\alpha(\hat{O}, V) = 0 \Leftrightarrow V = 0$.

F2': $\forall \lambda \in \mathbb{R}^+$, we have $F_\alpha(\hat{O}, \lambda V) = \lambda F_\alpha(\hat{O}, V)$.

F3': $F_\alpha(\hat{O}, V) + F_\alpha(\hat{O}, W) \geq F_\alpha(\hat{O}, V + W)$.

These imply that $F_\alpha(c, \dot{c})$ is Finsler metric.

C General Finsler metric for $\text{SU}(n)$: proof of Eq. (5.1)

Proof We prove (5.1) by three steps.

(1) Let us first show that $\tilde{F}(H)$ is only the function of eigenvalues of H and independent of the permutations of the eigenvalues. Notice that H always can be diagonalized under the transformation by an $\text{SU}(n)$ operator \hat{U} :

$$\hat{U}H\hat{U}^{-1} = \text{diag}(\gamma_1, \gamma_2, \dots, \gamma_n), \quad (\text{C.1})$$

where γ_i are the eigenvalues of H . Thus

$$\tilde{F}(H) = \tilde{F}(\hat{U}H\hat{U}^{-1}) = \tilde{F}(\text{diag}(\gamma_1, \gamma_2, \dots, \gamma_n)). \quad (\text{C.2})$$

where the first equality comes from the adjoint invariance (4.10). To see the independence of the permutations of the eigenvalues, we consider a $\text{SU}(n)$ operator \hat{U}' yielding

$$\hat{U}'\text{diag}(\gamma_1, \gamma_2, \dots, \gamma_n)\hat{U}'^{-1} = \text{diag}(\gamma_2, \gamma_1, \gamma_3, \dots, \gamma_n). \quad (\text{C.3})$$

where $\hat{U}'_{11} = \hat{U}'_{22} = 0$, $\hat{U}'_{12} = -\hat{U}'_{21} = 1$ and $\hat{U}'_{ij} = \delta_{ij}$ for $i, j > 2$. Again by (4.10) \tilde{F} is invariant under the permutation $\gamma_1 \leftrightarrow \gamma_2$. Similarity, $\tilde{F}[\text{diag}(\gamma_1, \gamma_2, \dots, \gamma_n)]$ is invariant under any kind of permutation on $\{\gamma_1, \gamma_2, \dots, \gamma_n\}$.

(2) Next, let us prove that if $H = H_1 \oplus H_2$ then $\tilde{F}(H) = \tilde{F}_1(H_1) + \tilde{F}_2(H_2)$, where $\tilde{F}_1(H_1) := \tilde{F}(H_1 \oplus \mathbf{0}_{n-k})$, $\tilde{F}_2(H_2) := \tilde{F}(\mathbf{0}_k \oplus H_2)$, $k = \text{Dim}(H_1)$ and $\mathbf{0}_i$ stands for the i -dimensional zero matrix.

Assume \hat{O}_i ($i = 1, 2$) to be the representations of $\hat{x}_i \in \mathcal{N}_i$ and one matrix representation of $(\hat{x}_1, \hat{x}_2) \in \mathcal{N}_1 \times \mathcal{N}_2$ can be given by $\hat{O} = \hat{O}_1 \oplus \hat{O}_2$. Then the axiom **G3** implies that the complexity should satisfy

$$\mathcal{C}(\hat{O}_1 \oplus \hat{O}_2) = \mathcal{C}_1(\hat{O}_1) + \mathcal{C}_2(\hat{O}_2). \quad (\text{C.4})$$

Here $\mathcal{C}_1(\hat{O}_1) := \mathcal{C}(\hat{O}_1 \oplus \mathbb{I}_2)$ and $\mathcal{C}_2(\hat{O}_2) := \mathcal{C}(\mathbb{I}_1 \oplus \hat{O}_2)$. To be specific, let us consider two 1-parameter subgroups $\mathcal{N}_1 = \exp(sH_1)$ and $\mathcal{N}_2 = \exp(sH_2)$ with $s \in \mathbb{R}$. Then $\mathcal{N} := \mathcal{N}_1 \times \mathcal{N}_2$ is the n -dimensional subgroup of \mathcal{O} . As

$H = H_1 \oplus H_2$, $\exp(\epsilon H) = \exp(\epsilon H_1) \oplus \exp(\epsilon H_2)$. From Eq. (C.4), we have

$$\mathcal{C}[\exp(\epsilon H)] = \mathcal{C}_1[\exp(\epsilon H_1)] + \mathcal{C}_2[\exp(\epsilon H_2)], \quad (\text{C.5})$$

which, in the limit $\epsilon \rightarrow 0$, reduces to

$$\tilde{F}(H) = \tilde{F}(H_1 \oplus H_2) = \tilde{F}_1(H_1) + \tilde{F}_2(H_2). \quad (\text{C.6})$$

In general, $H = \oplus_j H_j$ so Eq. (C.6) can be generalized as $\tilde{F} = \sum_j \tilde{F}_j(H_j)$.

(3) Finally, we combine the results in previous two steps. Noting the fact that $\text{diag}(\gamma_1, \gamma_2, \dots, \gamma_n) = \bigoplus_{j=1}^n \gamma_j$, we have

$$\tilde{F}(H) = \tilde{F}\left(\bigoplus_{j=1}^n \gamma_j\right) = \sum_{j=1}^n \tilde{F}_j(\gamma_j), \quad (\text{C.7})$$

where

$$\tilde{F}_j(\gamma_j) := \tilde{F}(\underbrace{\text{diag}(0, 0, \dots, 0, \gamma_j, 0, \dots, 0)}_{n-j}). \quad (\text{C.8})$$

Because \tilde{F} is invariant under the permutations of diagonal elements as shown in (C.3), $\tilde{F}_j(\gamma_j)$ is independent of the position of γ_j , which means there is a non-negative function f such that $\tilde{F}_j(\gamma_j) = f(\gamma_j)$. Thus

$$\tilde{F}(H) = \sum_{j=1}^n f(\gamma_j) = \sum_{j=1}^n f(i\text{Im}\gamma_j) = \sum_{j=1}^n f(i|\gamma_j|), \quad (\text{C.9})$$

where we used the fact that the eigenvalues of H are all pure imaginary: $\gamma_j = i\text{Im}\gamma_j$ and $f(i\text{Im}\gamma_j) = f(-i\text{Im}\gamma_j)$ from (4.2). Because the Finsler metric satisfies the homogeneity $\tilde{F}(\lambda H) = \lambda \tilde{F}(H)$ for $\lambda > 0$, one can find that $f(\gamma) = |\gamma|f(i)$ for arbitrary pure imaginary number γ ,¹³ which yields

$$F(c, \dot{c}) = \tilde{F}(H) = f(i) \sum_{j=1}^n |\gamma_j| = f(i) \text{Tr}(\sqrt{HH^\dagger}), \quad (\text{C.10})$$

where, without loss of generality, we may set the overall constant $f(i) = \lambda$. It proves (5.1). \square

¹³ This can be obtained, for example, by setting $H = \text{diag}(\gamma, -\gamma, 0, 0, \dots, 0)$ in Eq. (C.9).

D Other arguments for path-reversal symmetry

D.1 Inverse-invariance of relative complexity

Let us denote the set of all admitted gates g_i by g and the set of all g_i^{-1} by g^{-1} , i.e.,

$$g := \{g_i | \forall g_i \in g\}, \quad g^{-1} := \{g_i^{-1} | \forall g_i \in g\}, \quad (\text{D.1})$$

In discrete qubit systems, if g_i is one admitted gate then g_i^{-1} is also one admitted gate because quantum circuits are invertible, so $g^{-1} = g$. Let us investigate what this property implies for the circuit complexity.

For the unitary operators pair (\hat{U}, \hat{V}) and the fundamental gates set g , the “relative complexity” $d_\alpha(g; \hat{U}, \hat{V})$ is defined by the minimal number of required gates to transform from \hat{V} to \hat{U} under the gates set g . The index α is l or r , which is explained in Sect. 3. The relative complexities can be written in terms of the complexity as

$$d_r(g; \hat{U}, \hat{V}) = \mathcal{C}_r(g; \hat{U}\hat{V}^{-1}), \quad d_l(g; \hat{U}, \hat{V}) = \mathcal{C}_l(g; \hat{V}^{-1}\hat{U}). \quad (\text{D.2})$$

where $\mathcal{C}_\alpha(g; \hat{U})$ is the complexity of \hat{U} under the gates set g . As g_i and g_i^{-1} are both the elements in g , $\mathcal{C}_\alpha(g; \hat{U}) = \mathcal{C}_\alpha(g; \hat{U}^{-1})$ so it follows that $d_\alpha(g; \hat{U}, \hat{V}) = d_\alpha(g; \hat{V}, \hat{U})$.

Now let us consider the relative complexities of (\hat{U}, \hat{V}) and $(\hat{U}^{-1}, \hat{V}^{-1})$. If $\{g_1, g_2, \dots, g_n\}$ are some gates transforming \hat{V} to \hat{U} by the right-invariant way ($\alpha = r$), i.e.,

$$\hat{U} = g_n g_{n-1} \dots g_2 g_1 \hat{V}, \quad (\text{D.3})$$

we have

$$\hat{U}^{-1} = \hat{V}^{-1} g_1^{-1} g_2^{-1} \dots g_{n-1}^{-1} g_n^{-1} = \tilde{g}_1 \tilde{g}_2 \dots \tilde{g}_{n-1} \tilde{g}_n \hat{V}^{-1}, \quad (\text{D.4})$$

where $\tilde{g}_i := \hat{V}^{-1} g_i^{-1} \hat{V}$. By defining a new gates set

$$\tilde{g} := \{\hat{V}^{-1} g_i^{-1} \hat{V} | \forall g_i \in g\},$$

we have

$$\tilde{g} = \hat{V}^{-1} g^{-1} \hat{V} = \hat{V}^{-1} g \hat{V}.$$

Equations (D.3) and (D.4) are one-to-one correspondent. If we have a method by n gates to convert \hat{V} to \hat{U} under the gates set g by the right-invariant way, then we have also a corresponding method by n gates to convert \hat{V}^{-1} to \hat{U}^{-1} under the gates set \tilde{g} by the right-invariant way. The converse is also true. For the left-invariant way ($\alpha = l$), we have a similar conclusion. Thus, we have the following equality

$$d_\alpha(g; \hat{U}, \hat{V}) = d_\alpha(\tilde{g}; \hat{U}^{-1}, \hat{V}^{-1}). \quad (\text{D.5})$$

Here $\tilde{g} = \hat{V}^{-1} g^{-1} \hat{V}$ for $\alpha = r$ and $\tilde{g} = \hat{V} g^{-1} \hat{V}^{-1}$ for $\alpha = l$. This is a fundamental symmetry for relative complexity in quantum circuits.

Similar to the discussion in Sect. 4.2, for discrete qubit systems, we have

$$\tilde{g} \neq g$$

in general so

$$d_\alpha(g; \hat{U}, \hat{V}) \neq d_\alpha(g; \hat{U}^{-1}, \hat{V}^{-1}).$$

However, if we consider the $SU(n)$ group, as what we have argued in Sect. 4.2, the gates set should be replaced by Lie algebra $\mathfrak{su}(n)$. Because of $\mathfrak{su}(n) = \hat{V}^{-1} \mathfrak{su}(n) \hat{V} = \hat{V} \mathfrak{su}(n) \hat{V}^{-1}$, a symmetry Eq. (D.5) suggests that we should have

$$d_\alpha(\mathfrak{su}(n); \hat{U}, \hat{V}) = d_\alpha(\mathfrak{su}(n); \hat{U}^{-1}, \hat{V}^{-1}). \quad (\text{D.6})$$

In geometrical theory of complexity, the relative complexity in a $SU(n)$ group is naturally defined by the minimal length of curves connecting two operators such as

$$d_\alpha(\hat{U}, \hat{V}) := \min\{L_\alpha[c] | \forall c, s.t., c(0) = \hat{V}, c(1) = \hat{U}\}, \quad (\text{D.7})$$

where we omit the argument $\mathfrak{su}(n)$ in the relative complexity. Eq. (D.6) shows that the inverse map $\hat{x} \mapsto \hat{x}^{-1}$ should be a “distance-preserving map”, i.e.,

$$d_\alpha(\hat{U}, \hat{V}) = d_\alpha(\hat{U}^{-1}, \hat{V}^{-1}). \quad (\text{D.8})$$

The Finsler version of the *Myers–Steenrod theorem* says that every surjective distance-preserving map in a Finsler manifold is also an isometric map (see the theorem 3.2 of chapter 3 in Ref. [45]). Thus, we see that the inverse map should be an isometry of complexity geometry. As a result, we have

$$\forall c, \quad L_\alpha[c] = L_\alpha[c^{-1}]. \quad (\text{D.9})$$

This shows that the *Myers–Steenrod theorem* with basic symmetry (D.5) can guarantee the path-reversal symmetry for $SU(n)$ groups. Then according to Eq. (4.24), we find that $\tilde{F}(H) = \tilde{F}(\hat{U} H \hat{U}^{-1})$ and $\tilde{F}(H) = \tilde{F}(-H)$. This argument gives us the third method to show bi-invariance and the second method to show reversibility.

D.2 Equivalence between “bra-world” and “ket-world”

To describe a quantum system in a pure state, we usually use a “ket” state vector $|\cdot\rangle$. The time evolution of the system is governed by a unitary operator $c(t)$:

$$|\psi(t)\rangle = c(t)|\psi(0)\rangle. \quad (\text{D.10})$$

However, it is artificial to choose the “ket” vector to present physics and we can equivalently use the “bra” state vector $\langle\cdot|$. In the “bra-world”, the time evolution is given by

$$\langle\psi(t)| = \langle\psi_0|c(t)^\dagger = \langle\psi_0|c(t)^{-1}. \quad (\text{D.11})$$

If $c(t)$, the curve in $SU(n)$ group, presents the time evolution of a system in “ket-world”, then $c(t)^{-1}$ presents the time evolution of the same system in “bra-world”. Because all the physics should be invariant under the change of our formalism from “ket-world” to “bra-world” it is natural to expect that the “length(cost)” of $c(t)$ and $c(t)^{-1}$ are also the same:

$$L_\alpha[c] = L_\alpha[c^{-1}]. \quad (\text{D.12})$$

Starting with Eq. (D.12), we can derive other symmetries of complexity. This offers the fourth method to show bi-invariance and the third method to show reversibility.

E Invariance of the cost function

It seems that the complexity is invariant under some transformation is weaker than the requirement that the cost function is invariant under that transformation. In this appendix we explain indeed they are equivalent.

The argument is based on *Myers–Steenrod theorem*, which says that every surjective distance-preserving map in a Finsler manifold is also a diffeomorphism (the theorem 3.2 of chapter 3 in Ref. [45]).

Therefore, by *Myers–Steenrod theorem*, if we can show that the transformations relevant to (4.6), (4.19), and (4.22) are surjective distance-preserving maps, then we can say that they are diffeomorphisms so the cost is also invariant because those transformations are just diffeomorphisms.

Now we only need to prove that the transformations relevant to (4.6), (4.19), and (4.22) are surjective *distance-preserving* maps. For the proof, let us consider arbitrary two points \hat{V} and \hat{W} in an $SU(n)$ group manifold and denote the *distance* (shortest geodesic length) from \hat{V} to \hat{W} by $d(\hat{V}, \hat{W})$.

Note that the Finsler geometry is required to be right-invariant or left-invariant, which makes a connection between the distance and the complexity as follows

$$d_r(\hat{V}, \hat{W}) = d_r(\hat{\mathbb{I}}, \hat{W} \hat{V}^{-1}) = C_r(\hat{W} \hat{V}^{-1}), \quad (\text{E.1})$$

where the subscript r means we consider the right-invariant case. We explain only for the right invariant case because a left-invariant case works similarly.

For (4.6), we only need to show that the unitary transformation preserve the distance i.e.

$$d_r(\hat{V}, \hat{W}) = d_r(\hat{U} \hat{V} \hat{U}^{-1}, \hat{U} \hat{W} \hat{U}^{-1}). \quad (\text{E.2})$$

It can be shown as follows.

$$\begin{aligned} d_r(\hat{V}, \hat{W}) &= C_r(\hat{W} \hat{V}^{-1}) = C_r(\hat{U} \hat{W} \hat{V}^{-1} \hat{U}^{-1}) \\ &= C_r(\hat{U} \hat{W} \hat{U}^{-1} (\hat{U} \hat{V} \hat{U}^{-1})^{-1}) \\ &= d_r(\hat{U} \hat{V} \hat{U}^{-1}, \hat{U} \hat{W} \hat{U}^{-1}), \end{aligned} \quad (\text{E.3})$$

where for the second equality we used the fact that the complexity is invariant under a unitary transformation. By *Myers–Steenrod theorem* the unitary transformation is a diffeomorphism so $L[c(s)] = L[\hat{U}c(s)\hat{U}^{-1}]$.

For (4.22) we only need to show that the distance is invariant under the inverse map: $\hat{x} \mapsto \hat{x}^{-1}$ i.e.

$$d_r(\hat{V}, \hat{W}) = d_r(\hat{V}^{-1}, \hat{W}^{-1}). \quad (\text{E.4})$$

It can be shown as follows.

$$\begin{aligned} d_r(\hat{V}, \hat{W}) &= C_r(\hat{W} \hat{V}^{-1}) = C_r(\hat{V} \hat{W}^{-1}) = C_r(\hat{W}^{-1} \hat{W} \hat{V} \hat{W}^{-1}) \\ &= d_r(\hat{W} \hat{V}^{-1} \hat{W}^{-1}, \hat{W}^{-1}) = d_r(\hat{V}^{-1}, \hat{W}^{-1}), \end{aligned} \quad (\text{E.5})$$

where for the second equality we used the fact that the complexity is invariant under an inverse map and for the fifth equality we used the fact that the distance is invariant under a unitary transformation with $\hat{U} = \hat{W}^{-1}$ in (E.3).

Finally (4.19) is also valid because of (4.22) and (4.24).

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